THE PETROLOGICAL EVOLUTION OF THE LATE PALEOZOIC A-TYPE GRANITES OF THE COBEQUID HIGHLANDS, CANADA

by

Angeliki Papoutsa

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ABSTRACT

A-type granites comprise a distinct group of granitoid rocks the origin of which is well-debated. Their major alkaline characteristics are well-defined, however, little is known about the origins of the trace element and mineralogical variations they present. This thesis aims to determine the geological processes responsible for geochemical and mineralogical variations observed in the late Paleozoic A-type granites of the Cobequid Highlands. The eastern Cobequid Highlands contain complex plutons (Wentworth and Pleasant Hills) with calcic amphibole-bearing granites (calcic granites), whereas in the west, the North River and Cape Chignecto plutons contain only biotite-bearing granites. Sodic amphibole-bearing granites (sodic-granites) are found only in the Wentworth pluton.

The Wentworth calcic and sodic granites show systematic differences in wholerock chemistry, ε_{Nd} , magmatic temperatures, volatile contents and viscosity of the melt and, therefore, indicate derivation from different sources. Similar systematic differences have been identified between the biotite and calcic granites on a regional scale and, therefore, the mineralogically different granite types are not comagmatic. The biotite and calcic granites were both derived by similar degrees of partial melting of the crust, with the calcic granite being formed under elevated temperatures. Chemical and isotopic variations between the biotite and calcic granites correspond with the contrasting character of the two Neoproterozoic crustal blocks in the area. The sodic granites are the extreme fractionates of a coeval mafic melt that was underplated during extension.

In post-magmatic stages the granites were affected by several events of hydrothermal alteration. Hydrothermal fluids repeatedly mobilized the REE from the granites, resulting in the formation of distinct hydrothermal assemblages that can be correlated with specific types of hydrothermal alteration and the evolution of the shear zone. Regionally, precipitation of REE increases with time as the fluids became progressively more oxidizing and Fe-rich. The highest REE concentrations are found close to the Cobequid Fault and are possibly associated with a ca 320 Ma IOCG mineralizing event.

The early crustal evolution of Avalonia is further reviewed and new paleoreconstruction models are proposed over the last billion years. This may have implications on the current concepts of "proto-Avalonia" and engage new future research.

LIST OF ABBREVIATIONS USED

a.p.f.u.	atoms per formula unit
ACNK	Al ₂ O ₃ / (CaO+Na ₂ O+K ₂ O)
Aln	Allanite
Атр	Amphibole
An	Anorthite
And	Andradite
ANK	Al ₂ O ₃ / (Na ₂ O+K ₂ O)
Arf	Arfvedsonite
ASI	Aluminum Saturation Index
b.d.	below detection limit
Bar	Barite
BSE	Back-Scattered Electron
Bt	Biotite
CAB	Calc-Alkaline Basalt
Cal	Calcite
CAM	Cambrian
CARB	Carboniferous
ССР	Cape Chignecto pluton
Cer	Cerianite
Cher	Chernovite
Chl	Chlorite
CHUR	Chondrite Uniform Reservoir
Crs	Coarse-grained

DCHZ Deep Crustal Hot Zone DEV Devonian Depleted mantle DMM Debert River pluton DR EDI Ediacaran Energy Dispersive Spectrometry EDS EM Enriched Mantle EMB Eight Mile Brook pluton Epidote Ep Equigranular Eqgrnr Ferro-actinolite Fe-actin Ferro-eckermannite Fe-eck Ferro-edenite Fe-ed Fe-richt Ferro-richterite Ferro-winchite Fe-winch Frog Lake pluton FGL FLG Fountain Lake Group Fine-grained Fn FOZO Focal Zone mantle reservoir Feldspar Fsp Gain Brook pluton GB Gbr Gabbro Gr Granite Grnlr Granular Grnph Granophyric

GSB	Gunshot Brook pluton		
GSC	Geological Survey of Canada		
Hbl	Hornblende		
HFSE	High Field Strength Elements		
HG	Horton Group		
HIMU	High-µ mantle		
Hing	Hingganite		
HP-LT	High pressure-Low temperature		
HREE	Heavy Rare-Earth Elements		
Hy-bas	Hydroxylbastnäsite		
IAT	Island Arc Tholeiite		
ICP	Ion Coupled Plasma		
ICS	International Commission of Stratigraphy		
Ilm	Ilmenite		
INAA	Instrumental Neutron Activation Analysis		
IOCG	Iron oxide-Copper-Gold		
LOI	Loss On Ignition		
LREE	Light Rare Earth Elements		
Mag	Magnetite		
MALI	Modified Alkali-Lime Index		
MCS	McCallum Settlement pluton		
ME	Mount Ephraim pluton		
Med	Medium-grained		
MORB	Mid-Ocean Ridge Basalt		
NRP	North River pluton		

OFB	Ocean Floor Basalt		
OIA	Ocean Island Alkali basalt		
ΟΙΤ	Ocean Island Tholeiite		
ORD	Ordovician		
Par	Parisite		
РНР	Pleasant Hills pluton		
Plg	Plagioclase		
PM	Polson Mountain pluton		
PREMA	Prevalent Mantle		
Ру	Pyrite		
Pyr	Pyrochroite		
QFM	Quartz-Fayalite-Magnetite buffer		
Qz	Quartz		
RBF	Rockland Brook Fault		
Rbk	Riebeckite		
REE	Rare Earth Elements		
SAMBA	South America-BAltica		
SEM	Scanning Electron Microscope		
SHRIMP	Super High Resolution Ion Microprobe		
SIL	Silurian		
SMB	Six Mile Brook pluton		
Syn	Synchysite		
Thr	Thorite		
Ti-min	TiO ₂ mineral		
Tit	Titanite		

- UTM Universal Transverse Mercator coordinate
- **WDS** Wavelength Dispersive Spectrometry
- **WP** Wentworth pluton
- Zrn Zircon

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CHAPTER 1 INTRODUCTION

The granitoid rocks have been extensively studied in igneous petrology since they provide evidence of continental crustal growth and are closely related to geodynamics. Chappell and White (1974) proposed the first "genetic alphabet" classification scheme, in which all granites can be classified into four major types related to specific conditions of crustal formation. These proposed granitic types, which were different in terms of composition and magmatic sources (Table 1.1), included the S-type (sedimentary protolith- peraluminous composition), I-type (igneous protolith- metaluminous composition), M-type (mantle conditions – calc-alkaline chemistry), and a fourth alkaline type. No letter was attributed to the latter at the time, but it included alkaline granites enriched in incompatible elements. It was suggested that a probable source was a crustal residue from a previous event of partial melting and that these granites "are always formed late in the history of a fold belt". Thus, the alphabet classification scheme was originally based predominantly on geochemical criteria. Soon after the pioneering work of Chappell and White (1974), Loiselle and Wones (1979) related the fourth granitic type to rift zones and generation within stable continental blocks. They proposed the letter "A" for this granitoid group and the term "A-type" was, therefore, established predominantly due to their anorogenic affinity.

The mainly tectonic affiliation of the term "A-type" was a deviation from the original concept of "genetic alphabet" classification scheme which was based predominantly on geochemical characteristics and magmatic sources (Table 1.1). Furthermore the I-, S-, and M-types were all associated only with crustal formation. Although M-type is often confused as "mantle-derived", its original definition

Granite type	Characteristic minerals	Geochemical features	Magma sources
S	cordierite, garnet, muscovite, biotite	Peraluminous	Meta-sedimentary sequences
Ι	hornblende	Metaluminous	Igneous crustal rocks
М	hornblende	Calc-alkaline, volcanic arc signature	Subducted oceanic crust
A*	arfvedsonite, aegirine, hornblende biotite	Alkaline, within-plate affinities	fractionation of mafic melts, partial melting of granodiorite/ tonalite, granulitic residue from previous melting event

Table 1. 1 Mineralogical and geochemical criteria for the alphabet classification scheme for granites.

Notes: * A-type term was not introduced with the original classification scheme.

referred to "crustal sources associated with mantle conditions", such as melting of subducted oceanic crust. As stated in the submitted abstract of Chappell and White (1974) "The chemical composition of granites precludes the possibility of direct mantle derivation". One of the proposed origins of A-type granites was the fractionation of alkali basalts (Loiselle and Wones, 1979) and, therefore, this granitoid group was not of a strictly crustal origin, a further deviation from the original concept of the alphabet classification scheme. Despite their original definition, A-type granites now include not only anorogenic but post-orogenic rocks as well (Maniar and Piccoli, 1989; Creaser et al., 1991), a fact that led researchers to propose a geochemical classification of these rocks instead of a tectonic one (e.g. Whalen et al., 1987, Fig 1.1).

Frost and Frost (2010) proposed the term "ferroan" instead of "A-type", since in the literature the majority of the described A-type granites, from different tectonic settings, are ferroan rocks. Several petrogenetic models have been proposed in the past decades for these granitoids, and as a result, the term itself and the origin of A-type granites became the subject of one of the biggest debates in igneous petrology.



Chapter 1. 1 Geochemical classification of the S-, I-, M-, and A-type granites according to Whalen et al. (1987).

1.1 THEORETICAL BACKGROUND OF A-TYPE GRANITES

The term "A-type granite" initially referred to granites that were generated along continental rift zones with mildly alkaline geochemistry and crystallized under low water fugacities (Loiselle and Wones, 1979). The fact that this type of granite is richer in Fe, K, rare metals and REE compared to other type of granites (Fig. 1.1) implies a distinctive petrogenesis. The first petrogenetic model suggested that A-type granites derive from fractionation of mantle-derived alkali basalt with or a without crustal contribution. When a crustal source was involved, it was hypothesized to be anhydrous granulitic lower crust (Collins et al., 1982). The A-type granite group was further proposed to be a highly fractionated subgroup of I-type granites (Whalen et al., 1987).

Creaser et al. (1991), however, demonstrated that partial melting of granulitic crust cannot produce A-type magmas, because such a melt would be depleted in Si, Fe and K and enriched in Ca and Al compared to the protolith; such geochemical features are not found in A-type granites. Partial melting of tonalitic to granodioritic crust, instead, can produce A-type magmas (Creaser et al., 1991; Frost and Frost, 2010). Experimental results suggest that the dehydration melting of calc-alkaline granitoids in the shallow crust gives rise to A-type granitic melts (Patiño-Douce, 1997). The high temperatures required for such melting imply the involvement of hot mafic magma close to the Earth's surface, which can also have a chemical contribution (Patiño-Douce, 1997). Metaluminous A-type granites can also form by differentiation of a tholeitic basalt magma, whereas peralkaline granites may form by differentiation of transitional and alkali basalt (Frost and Frost, 2010).

A-type granites, although they have specific chemical characteristics, may present important mineral variations. The major mineralogical diversity observed in A-type granites worldwide is that of primary ferromagnesian minerals hosted in these rocks. These granitoid rocks may contain biotite, amphibole and/or clinopyroxene of varying compositions, that led researchers in the last few years to identify contrasting A-type associations and magmatic conditions in alkaline provinces (Gualda and Vlach, 2007; Shellnutt and Iizuka, 2011).

A-type granites are related to mineralization of REEs and rare metals such as Nb, Y, Ta, Th and U (Schmitt et al., 2002). However, the conditions under which these

mineral deposits formed are still poorly understood. The development of a magmatic model that could explain the enrichment of A-type granitic magmas in these metals could give an economic potential to the study of these rocks.

1.2 STATEMENT OF THE PROBLEM

A large number of different geological processes and origins have been attributed to the formation of the A-type granites, and the mineralogical diversity in these rocks appears to be potentially of great importance for understanding their petrogenesis. The main purpose of this thesis is to investigate mineralogical and geochemical variations observed in A-type granites and correlate them with specific magmatic processes and sources, thus providing further insight to their complicated petrogenesis, which has been the subject of debate for years. Furthermore, this work aims to correlate the petrogenesis of these rocks with the evolution of the crust, and its structure as the result of its previous tectonic history.

The Cobequid Highlands in Nova Scotia provide a unique opportunity for such a study. A large number of coeval A-type granite plutons were emplaced along the active faults of a developing shear zone in the late Paleozoic. These granitic plutons were emplaced under the same regional tectonic regime, which is well understood, and although they have similar overall geochemistry, they present important differences in mineralogy and trace element geochemistry. The Wentworth pluton is richer in Zr, Nb and Y than the other plutons, whereas the North River pluton appears to be the most depleted in these elements. The A-type granites of the area also present a distinct bimodality in Mg, a trend that is observed principally in the Pleasant Hills and

Wentworth plutons, and is best defined in the latter (Fig. 1.2). This bimodality is not the result of post-magmatic alteration and represents a primary feature of these rocks.



Figure 1. 2 Harker-type diagram showing the bimodal distribution of Mg in the granites of the Wentworth pluton (Papoutsa, 2012).

In terms of mineralogical composition, the Pleasant Hills and the Wentworth plutons are the only intrusions in which the granites contain amphibole as ferromagnesian phase (Pe-Piper et al., 1998a; Pe-Piper, 2007), whereas the majority of the A-type granites of the other Late Paleozoic plutons contain principally biotite (Pe-Piper, 1991; Pe-Piper, 1995b). Furthermore, the granites of the Wentworth Pluton are the only ones regionally that contain sodic amphibole as a primary magmatic phase (Pe-Piper, 2007). In the same intrusion a complete sequence of REE-Nb-Y minerals from magmatic to hydrothermal stages has been identified (Papoutsa and Pe-Piper, 2013). However, no similar work has been done on the other plutons of the shear zone.

Certain questions arise when it comes to a correlation between mineralogy and geochemistry. Geochemically all the late Paleozoic plutons of the Cobequid Shear Zone show, in general, the same alkalinity ranging from peralkaline to peraluminous (Pe-Piper and Piper, 2003), however it is only the Wentworth Pluton that contains alkali minerals such as sodic amphibole (Pe-Piper, 2007). The cause for the apparent bimodality of Mg and the reason why this is the only element that presents such a trend is not understood. The REE-Y-Nb enrichment of A-type granites is considered one of the primary diagnostic features of these rocks and has been commonly used in the classification of these rocks (Eby, 1992; Whalen et al., 1987). However, there is no clear correlation between whole-rock geochemistry and the mineralogy of REE-host phases that could shed light into the exact nature and conditions of this enrichment.

The mineralogical diversity of the A-type granites in the Cobequid Highlands may, thus, reflect a rather complex magmatic system, providing the opportunity to investigate the impact of source heterogeneity and contrasting magmatic conditions on the mineralogy of the A-type granites, with limited control of tectonic variations. Such a study would provide further insight into the origin of some primary mineralogical features of A-type granites, and lead to a better understanding of their petrogenesis.

1.3 GEOLOGICAL SETTING

1.3.1 COBEQUID HIGHLANDS

The Cobequid Highlands are located in the Avalon terrane of the Canadian Appalachians, just north of the Cobequid Fault, which separates this terrane from the Meguma terrane (Fig. 1.3). The late Paleozoic plutons and their extrusive equivalents in the Cobequid Highlands are underlain by Neoproterozoic rocks, minor Silurian to Lower Devonian sedimentary rocks(Pe-Piper and Piper, 2003). Within the Cobequid Highlands the Cobequid Shear Zone consists of a series of late Paleozoic faults that were active

during latest Devonian to earliest Carboniferous (Pe-Piper and Piper, 2003). The main faults within theCobequid Shear Zone (Kirkhill, Rockland Brook and Portapique faults) are sub-parallel to each other and to the Cobequid fault, while second order shear zones form splays (Pe-Piper and Piper, 2003). This fault system continues northwestward to the Hollow Fault in the Antigonish Highlands and the fault zone along the northwestern coast of Cape Breton Island (Fig. 1.3). The faults of the Cobequid Shear Zone merge to the southwest with the Cobequid-Chedabucto fault zone, which was initiated in the late Carboniferous and marks the northern limit of the Minas basin (Mawer and White, 1987; Pe-Piper and Piper, 2003). The late Paleozoic plutons were emplaced along the faults of the actively deforming Cobequid Shear Zone and consist mainly of granitic and gabbroic bodies (Fig. 1.3).

In the late Devonian, the Rheic Ocean was subducted beneath Nova Scotia producing the voluminous South Mountain Batholith and satellite plutons (Clarke et al., 1997). Towards the end of the Devonian, incipient closure of the Rheic Ocean at promontories led to orogen-parallel dextral strike slip faulting (Hibbard and Waldron, 2009), resulting in distributed extension in the Magdalen Basin (Fig. 1.3) as a step-over zone. Crustal thinning continued through at least the early Carboniferous, accompanied by underplating of a thick gabbroic layer beneath the Gulf of St Lawrence (Marillier and Verhoef, 1989; Marillier and Reid, 1990). It is suggested that this magma reached the surface through the basin-margin faults to give flood basalts >1 km thick in the Fisset Brook Formation of western Cape Breton Island and the Fountain Lake Group of the Cobequid Highlands (Pe-Piper and Piper, 1998a).

All late Paleozoic granites in the Cobequid Highlands have A-type affinities, whereas the gabbros are ferro-gabbros. Synchronous with the plutons are the bimodal volcanic rocks of the Fountain Lake Group that show similar chemical affinities with their intrusive equivalents (Dessureau et al., 2000; Pe-Piper and Piper, 2003).



Figure 1.3 Simplified geological map of the Canadian Maritimes showing the location of the Cobequid Highlands, the Avalon and Meguma terranes and the Magdalen basin (modified from Pe-Piper and Piper, 2003). HF= Hollow Fault, AH= Antigonish Highlands.

1.3.2 AVALONIAN COUNTRY ROCK

The Avalonian crust, exposed as the basement rocks in the Cobequid Highlands,

may represent the lower-crustal rocks that had potentially been involved in the

petrogenesis of the late Paleozoic A-type granites. The Avalon terrane of the northern

Appalachians in mainland Nova Scotia is characterized by Neoproterozoic (653-570 Ma)

arc-related sequences, uncomformably overlain by thin lower Paleozoic supracrustal

rocks (Murphy and Nance, 2002). This terrane comprises a series of collisional blocks including I-type plutons, coeval volcanic rocks, minor oceanic crust and clastic sedimentary successions (Pe-Piper and Murphy, 1989; Murphy, 2002).

In the Cobequid Highlands, the Neoproterozoic basement has been divided into two major blocks, separated by the Rockland Brook fault (Pe-Piper and Piper, 2003). To the north lies the Jeffers Block, whereas the Bass River Block is located just south of the Rockland Brook fault (Fig. 1.4). The Rockland Brook fault appears to be a reactivated Neoproterozoic fault (Pe-Piper and Piper, 2003) and is considered to represent a structural front, the straight segments of which define a steep, parallel-walled ductile shear zone (Miller, 1991).



Figure 1. 4 Simplified geological map of the Cobequid Highlands showing the location of the studied plutons, major faults and the Neoproterozoic crustal blocks (modified from Pe-Piper and Piper, 2003).

The two Neoproterozoic blocks show differences in stratigraphy and radiogenic (Pb and Nd) isotope signature (Pe-Piper and Piper, 1998b; MacHattie and White, 2012). The Jeffers Block comprises a basement of Cryogenian greenschist to amphibolite facies gneisses, covered by late Neoproterozoic volcanic and volcaniclastic rocks, interbedded with sedimentary rocks, and with small high-level (Ediacaran-Ordovician) plutons. The Bass River Block consists of a Tonian basement of greenschist- lower greenschist, platformal metasedimentary rocks and ocean floor metabasalts, and late Neoproterozoic (Ediacaran) plutonic rocks (Pe-Piper et al., 1996; MacHattie and White, 2012).

Lithostratigraphic correlations of volcanic and sedimentary rocks between the Cobequid and the Antigonish Highlands suggest that these two areas share the same basement and similar volcanic and depositional history (Murphy et al., 1991). It is therefore argued that the Cobequid and Antigonish Highlands were parts of the same volcanic arc during the Neoproterozoic. Furthermore, the Cobequid Highlands were juxtaposed against the Antigonish Highlands until early Carboniferous, before the 30 km dextral displacement on the Cobequid fault and the development of the Stellarton Basin (Murphy et al., 1991).

The Antigonish Highlands are representative of typical west Avalonian crust and provide a record of four major igneous events. The oldest episode is the 615 Ma igneous event which records early rifting, and resulted in the emplacement of mafic, intermediate and felsic rocks (Murphy et al., 2008). The second igneous event occurred in early Cambrian in a post-arc, transtensional setting, represented by bimodal volcanic and plutonic rocks (Murphy et al., 2008). Middle Ordovician magmatism in the Antigonish Highlands is expressed by bimodal alkaline magmatism that includes A-type granites and A-type rhyolites. The youngest magmatic episode occurred in middle Devonian after the collision of Avalonia with Laurentia and resulted in the extrusion of the McArras Brook basalts (Murphy et al., 2008).

The only recognized A-type granitic plutons in the Antigonish Highlands are, therefore, of Ordovician age. These granites are interpreted to have been derived from

crystal fractionation of a parent mafic magma and assimilation with crustal materials, during extension in an ensialic island arc regime (Escarraga et al., 2011).

1.3.3 Previous work on the late Paleozoic plutons of the Cobequid Highlands

The Cobequid Highlands comprise a large number of plutonic bodies with varying complexity. Specific late Paleozoic granitic plutons were selected for detailed study throughout the region, on the basis of variable petrology, distribution of igneous phases and relative age. The late Paleozoic plutons selected for this study are the Wentworth and Pleasant Hills plutons (complex intrusions with significant amounts of gabbro) in the eastern part, the North River pluton in the central part (small intrusion with minor gabbro), and the Cape Chignecto pluton in the western part of the Cobequid Highlands (granite intrusion with moderate amounts of gabbro).

The Wentworth pluton is located just north of the Rockland Brook fault and consists of a zone of granite of different generations in its northeastern part, that has a gradational contact with a large gabbroic body in the southwestern part (Fig. 1.4) (Pe-Piper and Piper, 2003). At its northern edge, the Wentworth pluton intrudes Silurian country rock and is onlapped by Carboniferous sedimentary rocks (Koukouvelas et al.,2002). The main granite of the Wentworth pluton is an alkali-feldspar granite that is found is outcrops of hundreds of meters within the pluton. Fine-grained granite and porphyritic rhyolite occur as E-W striking sheets tens to hundreds of meters wide in the eastern part of the pluton (Pe-Piper, 1998). The pluton has an ESE-striking contact with coeval volcanic rocks of the Byers Brook Formation (Fig. 1.4). The transition from granite to Byers Brook rhyolite is represented by sheets of homogeneous very fine-

grained granite and porphyritic rhyolite exposed at the western part of the Wentworth pluton. These are locally flow banded and interlayered with minor volcaniclastic sedimentary rocks and basalt flows (Pe-Piper, 1998). The main granite of the Wentworth pluton is in places intruded by gabbro bodies with chilled margins. The gabbro is younger (357±4 Ma, Table 1.2) than most mafic intrusions in the Cobequid Highlands and postdates the emplacement of most Wentworth Pluton granites (362 ± 2 Ma, Table 1.2). In the transition zone between the gabbro and the granite, the two lithologies show irregular, lobate contacts in which the gabbro is chilled against the granite (Pe-Piper, 1998). Late granites have lobate contacts with chilled margins against the gabbro, or occur as pods in the latter (Koukouvelas et al., 2002). A few contacts show mixing between the two magmas and the hybridized rocks appear inhomogeneous and have the composition of a granodiorite (Pe-Piper, 1998). Syn-magmatic deformation is observed through precrystallization fabrics where oriented feldspars were found in granodiorites close to the contact with the gabbro. Aplite veins are found throughout the pluton and crosscut all main lithologies (Pe-Piper, 1998).

The Pleasant Hills pluton $(361 \pm 2 \text{ Ma}, \text{Table 1.2})$ is located just southwest of the Wentworth pluton (Fig. 1.4). The southeastern margin of the Pleasant Hills pluton is in contact with the Rockland Brook fault whereas its southwestern part is just north of the Cobequid fault (Fig. 1.4). It is a composite intrusion consisting of gabbro, granodiorite and granites some of which are high-level porphyries (Pe-Piper et al., 1989). The pluton intrudes sedimentary rocks of the synchronous Horton Group, Silurian sedimentary rocks and Neoproterozoic volcaniclastic rocks (Pe-Piper et al., 1998). Geochronological data (Table 1.2) suggest that the pluton was emplaced within a period of a few million years.

The granite and the gabbro in the Pleasant Hills pluton, however, appear to be of several generations, based on cross-cutting relationships (Pe-Piper and Piper, 1998a). The Pleasant Hills pluton is divided by a shear zone into a western and eastern part. The mafic rocks are dominant in the eastern part of the pluton and along the southern margin of the intrusion. The main body of the pluton is a coarse grained alkali-feldspar granite that shows locally rapakivi texture, elongate sub-parallel to the length of the pluton (Pe-Piper and Piper, 1998a). Finer-grained granite occurs mainly at the northern and northeastern parts of the pluton along with small bodies of gabbro that predate the granite. The main granite is cut by porphyritic rhyolite bodies at the northern part of the pluton (Pe-Piper and Piper, 1998a). Lobate contacts between granite and gabbro suggest contemporaneous emplacement. In the eastern part of the pluton, dominant gabbro is cut by the main coarse-grained alkali feldspar granite. Younger mafic dykes are common and are in places deformed with the granites, whereas aplite veins cut all other lithologies and are oriented in conjugate sets striking NE and NW (Pe-Piper and Piper, 1998a). The pluton was emplaced during dextral shear and along its southern margin planar mylonitic shear zones are deformed by ductile folds that have been interpreted as deep-level deformation (Koukouvelas et al., 2002). Younger movement of the Rockland Brook fault resulted in brittle deformation that overprints the ductile structures (Miller, 1991). Koukouvelas et al. (1996) interpreted that flat-lying planes that cut all the lithologies in the pluton as transtensional features. C-S structures and asymmetrical folds suggest a southward direction of movement. Syn-magmatic deformation was also observed at a microscopic scale where pre-full orientation of plagioclase in gabbros and granites is parallel to the

foliation of small granite dykes and overprinted by solid-state deformation of plagioclase and quartz (Koukouvelas et al., 1996).

The North River pluton (356 ± 17 Ma, Rb-Sr errorchron, Pe-Piper et al., 1989) is a small teardrop-shaped intrusion in the central part of the Cobequid Shear Zone, between the Rockland Brook and Kirkhill faults (Fig. 1.4). The long axis of the pluton is subparallel to the trace of the Cobequid fault and no more than 2 km away from it. The southern part of the North River pluton is pervasively fractured (Pe-Piper, 1991). The dominant igneous phase in the North River pluton is a coarse grained alkali feldspar granophyric granite that becomes finer grained equigranular or porphyritic, towards the margins of the intrusion (Pe-Piper, 1991). Gabbroic rocks in the North River pluton appear to predate the main granite, since the former is cut by the latter and the granite shows chilled margins against the gabbro. Coeval granite and gabbro, however, are observed in this pluton, in areas where the granite occurs as pods within the gabbro with evidence of hybridization (Pe-Piper, 1991). Late microgranite sheets and gabbro dykes cut other lithologies in the pluton and also the country rock. In the northern part of the pluton, south-dipping small granite dykes are isoclinally folded under ductile-solid-state conditions, indicating northward thrusting (Koukouvelas et al., 2002). Sigmoidal foliation planes that are at an angle to the rock foliation are interpreted as solid-state foliation while flame structures at the rim of mafic xenoliths suggest magmatic flow during NW thrust movements. Tension gashes in the granite that were created during that movement are filled with secondary epidote and quartz and are ductilely folded (Koukouvelas et al., 1996). Brittle deformation in the North River pluton is expressed by flat-lying E-W shear zones that are chatacterized by cataclastic zones in which the displacement of aplite

dykes suggests a displacement up to hundreds of metres top-to-south (Koukouvelas et al., 1996).

The Cape Chignecto pluton is located in the western part of the Cobequid Highlands (Fig. 1.4). It lies north of the Cobequid fault and west of the Kirkhill fault (Pe-Piper, 1995). The main phases of the pluton are a variably deformed, medium-grained alkali-feldspar pink granite and gabbro (Pe-Piper, 1995b). The gabbro may occur as small bodies or sheet-like intrusions in the granites and has a thickness of hundreds of metres along the southern margin of the pluton. Most contacts with granites are tectonized, however igneous contacts are present as well where the gabbro is chilled against the granite (Waldron et al., 1989). The presence of intermediate hybrid rocks indicates coeval emplacement of mafic and felsic magmas. The granites of the Cape Chignecto pluton show a variety of textures including equigranular, inequigranular, graphic, and granophyric. Locally, granites were found to contain miarolitic cavities, suggesting a rather shallow emplacement (Pe-Piper, 1995b). Late mafic sheets in granite have developed strong foliation parallel to their margins and contain elongated pods and lenses of fractured foliated granite (Waldron et al., 1989). Foliated mafic sheets to the north of the pluton appear to form an anastomosing network of subhorizontal orientation, while strongly foliated granites to the south show C-S structures indicative of N-vergent overthrusting on the subhorizontal to SE-dipping shear planes (Koukouvelas et al., 1996). Brittle fractures in the Cape Chignecto pluton are associated with hematite staining and green secondary biotite (Waldron et al., 1989). To the north of the pluton, foliations and lineations are weakly developed but mm-scale shear zones show intense mylonitic foliation. The orientation of these shear zones is variable but most of them dip E or SE.
The alkaline affinity of the late Paleozoic granites in the Cobequid Highlands has been interpreted as the result of magma genesis from partial melting of the lower crust during crustal extension (Pe-Piper et al., 1988; 1991; 1998). Pe-Piper et al. (1991) concluded that variations in trace elements such as Nb and Zn are related to variable conditions of partial melting. Variations in large ion lithophile elements (LILE) such as Th and Rb were taken to have been influenced by segregation during magma expulsion from a crystal mush during shear (Koukouvelas et al., 2002). In the Wentworth pluton, Pe-Piper (2007) examined the mineralogical variations of amphiboles and concluded that sodic amphiboles are dominant in the early granites of the Wentworth pluton, whereas calcic amphibole is present in most late granites. Therefore it was interpreted that anatexis of pre-existing granites by the intrusion of voluminous gabbro influenced the type of amphibole crystallized in the Wentworth pluton. In the same intrusion Papoutsa and Pe-Piper (2013) identified a paragenetic sequence of REE-Nb-Y-rich minerals from magmatic to hydrothermal stages, suggesting a post-magmatic mobilization of these elements. In the Pleasant Hills pluton Pe-Piper et al. (1998) concluded that the granites show an enrichment in high field strength elements (HFSE) and a decrease in Sr, Ba and Ca that was associated with probable fractionation of hornblende and calcic plagioclase. The main granite of the North River pluton, on the other hand, shows a distinct geochemical character in being less alkali than the rest of the plutons and has smaller concentrations of Nb, Y, Zr. However the high alkalies and SiO2 are still indicative of a subalkalic A-type granite (Pe-Piper, 1991).

Table 1. 2 Published geochronological controls for late Paleozoic magmatic and related postmagmatic events in the Cobequid Highlands.

Geological Unit	Mineral	Method	Age	Reference
	Main magmati	c phase		
Early granites of the Wentworth				
pluton	zircon	U-Pb	362±2	Doig et al. (1996)
-	arfvedsonite	Ar-Ar	368±4	Pe-Piper et al. (2004)
Late granites (probable-syn gabbro				
granites)	arfvedsonite	Ar-Ar	356±3	Pe-Piper et al. (2004)
Wentworth gabbro	hornblende	Ar-Ar	357±4	Pe-Piper et al. (2004)
-	biotite	Ar-Ar	353±2	Pe-Piper et al. (2004)
Byers Brook Formation (Upper)	zircon	U-Pb	358±1	Dunning et al. (2002)
Diamond Durals Formation (Unres)		malamalama	24012	Utting et al. (1989),
Diamond Brook Formation (Upper)		palynology	348±3	Menning et al. (2006)
Diamond Brook Formation (Middle)	zircon	U-Pb	353±3	Dunning et al. (2002)
Diamond Brook Formation (Middle-				
Lower)		palynology	359±2	Martel et al. (1993)
Pleasant Hills pluton main granite	zircon	U-Pb	358±2	Doig et al. (1996)
Pleasant Hills pluton porphyritic				
rhyolite	zircon	U-Pb	356±3	Dunning et al. (2002)
Pleasant Hills pluton fine-grained				
granite	zircon	U-Pb	360±2	Dunning et al. (2002)
Pleasant Hills pluton- granodiorite	hornblende	Ar-Ar	358±4	Pe-Piper et al. (2004)
Pleasant Hills pluton- diorite	hornblende	Ar-Ar	359±4	Pe-Piper et al. (2004)
Cape Chignecto pluton main granite	zircon	U-Pb	361 ± 2	Doig et al. (1996)
Cape Chignecto pluton- rhyolite	zircon	U-Pb	355 ± 2	Dunning et al. (2002)
Post	t-emplacement o	lated events		
Albitized granite, West Moose River	hydrothermal	$\Delta r_{-} \Delta r$	353+4	$Pe_Piper et al (2004)$
pluton	riebeckite	7 11 -7 11	JJJ-7	1 c-1 iper et al. (2004)
Deformed diorite, Cape Chignecto	biotized	Ar-Ar	347+3	Pe-Piper et al. (2004)
pluton, metamorphic biotite	hornblende	7 H 7 H	517-5	1 e 1 iper et ul. (200 i)
Altered diorite, southwestern Cape	low-Ti	Ar-Ar	343+3	Pe-Piper et al (2004)
Chignecto pluton	biotite		0.0-0	1 • 1 ·p•1 •• ••• (200 ·)
	mixed			
Gneissose granodiorite south of the	hornblende-	Ar-Ar	336-	Pe-Piper et al. (2004)
Rockland Brook fault	biotite		343 ± 4	r r · · · · · · · · · · · · · · · · · ·
	separates			
Medium-grained diorite, Cape	hydrothermal	Ar-Ar	342±3	Pe-Piper et al. (2004)
Chignecto pluton	biotite			1 ()
Lamprophyre dyke in the central	whole-rock	Ar-Ar	334±3	Pe-Piper et al. (2004)
Cobequid Highlands				1 ()
Lamprophyre dyke in the central	green biotite	Ar-Ar	326±2	Pe-Piper et al. (2004)
Cobequid Highlands	e			1 ()
Carbonate-suphide vein in Copper	pyrite	Re-Os	323±8	Kontak et al. (2008)
	<u>.</u> .			
Altered suitstone adjacent to a				
carbonate-sulphide vein with	hydrothermal	A A	309.5	Kantal at 1 (2000)
intergrowths of nydrothermal	muscovite	Ar-Ar	±1.6	Kontak et al. (2008)
I ako donosit				
LAKE GEDOSIL				

Notes: *: All Ar-Ar ages published prior 2008 have been corrected from the original data according to Kuiper et al. (2008), due to new intercalibrations in Ar-Ar dating.

1.4 OBJECTIVES AND THESIS ORGANIZATION

In order to understand the evolution of the late Paleozoic A-type granites in the Cobequid Highlands and their petrogenesis, this thesis is focused on specific mineralogical and geochemical differences, as discussed above. In this section the specific objectives of this thesis are described. This thesis is organized in a paper-based format and comprises three research manuscripts, presented in chapters 2-4.

1.4.1 CHAPTER 2

The origin of the variability in the Wentworth A-type granites is investigated in chapter 2. This chapter is entitled "Geochemical variation of amphiboles in A-type granites as an indicator of complex magmatic systems: Wentworth pluton, Nova Scotia, Canada" and a version of this chapter has been published as Papoutsa, A., and Pe-Piper, G., 2014, Geochemical variation of amphiboles in A-type granites as an indicator of complex magmatic systems: Wentworth pluton, Nova Scotia, Canada: Chemical Geology, v. 384, p. 120-134. This study investigates why the Wentworth granite has a bimodal abundance of Mg, and is, regionally, the only granite that contains both primary sodic and calcic amphiboles, and thus has more geochemical and mineralogical diversity than other plutons. Microprobe analyses of amphiboles were compared to the modal, wholerock composition and Sm/Nd isotope data of the granites. Temperature (from zircon and amphibole), pressure (from amphibole), fluorine and water-in-melt contents have been calculated, to investigate the magmatic conditions. Correlation between whole-rock chemistry and modal composition revealed that the low-Mg granites are those containing sodic amphibole, whereas the high-Mg group includes granites with calcic amphiboles. Further systematic variations between the two groups are identified in terms of trace

element geochemistry, isotopes, and inferred magmatic conditions. The systematic differences between the granites indicate the presence of two magmatic systems in the Wentworth pluton. This study shows that a compositional variation in amphiboles of Atype granites can also result from the contribution of more than one magmatic source, not just extreme differentiation, and that the Wentworth magmatic system is more complex than previously proposed.

My contributions in this chapter include the development of the ideas presented, petrographic descriptions of the samples, mineral analyses, chemical modelling and interpretations. Geological maps and samples were provided by the co-author Georgia Pe-Piper. Furthermore as a co-author, she contributed in the writing of the manuscript by providing specific information and through constructive discussions that helped to develop the ideas presented.

1.4.2 CHAPTER 3

In chapter 3, the geochemical and mineralogical variations of the A-type granites from all studied plutons are examined in detail. This Chapter is entitled "*Systematic mineralogical diversity in A-type granitic intrusions: Control of magmatic source and geological processes*" and is currently in press in the Geological Society of America Bulletin. In this chapter, the ideas and hypotheses introduced for the Wentworth pluton are further tested in the A-type granites from all plutons, which are classified into mineralogical types according to the ferromagnesian mineral present. The whole-rock geochemistry, radiogenic and stable isotope chemistry, and estimates of magmatic parameters for these types are compared and evaluated. Systematic differences between these different types suggest that they do not have the same source. The origin of each

mineralogical type is approached by chemical modelling of partial melting and fractionation processes and potential magmatic sources are identified. As a result, a regional petrogenetic model is proposed that includes the variable modes of origins required for the late Paleozoic A-type granites of the Cobequid Highlands, by correlating specific geological processes with the observed variations in mineralogy and geochemistry.

My contributions in this chapter include the original development of the ideas presented regarding origins of the different granites, geochemical modelling and synthesizing a petrogenetic model. The co-authors Georgia Pe-Piper and David Piper provided all the geological maps, samples, and most whole-rock analyses. Furthermore, several discussions with the co-authors and their constructive comments helped to refine the proposed models and improve the presentation of the new ideas in this chapter.

1.4.3 CHAPTER 4

The REE-enrichment of the A-type granites is one of the most important diagnostic features that is often used as a petrogenetic indicator of these rocks but the correlation with the mineralogy of the granites, particularly in the Cobequid Highlands is not well established. In order to evaluate the nature and conditions of this enrichment, an understanding of the formation of the REE-hosting minerals present is necessary. In chapter 4 the petrographic study of REE-enriched samples from the Wentworth, Pleasant Hills, North River and Cape Chignecto plutons revealed that although the Wentworth pluton contains the highest amounts of primary REE-minerals (Papoutsa and Pe-Piper, 2013), the rest of the plutons record a predominantly hydrothermal enrichment. As described in the previous section, the A-type granites of the Cobequid Highlands show

strong deformation and fracturing close to the main faults. Therefore it is important to evaluate the conditions of REE-mobility in post-magmatic stages and investigate the effect of repeated hydrothermal activity on the mineralogy and geochemistry of the Atype granites. The chapter is entitled "Variation of the REE-hydrothermal Circulation in Complex Shear Zones: The Cobequid Highlands, Nova Scotia" and a version of this chapter has been published as Papoutsa, A., and Pe-Piper, G., 2015, Variation of the REE-hydrothermal Circulation in Complex Shear Zones: The Cobequid Highlands, Nova Scotia: Canadian Mineralogist, v. 52, p. 943-968. This study correlates the presence of specific hydrothermal REE-minerals with types of fractures, providing insight into the hydrothermal behavior of these elements. Furthermore, observations on textural relationships and the distribution of different fractures in the studied plutons provide information about the evolution of the regional hydrothermal system during the Carboniferous. A paragenetic sequence of hydrothermal REE-minerals is proposed and correlated with different dated hydrothermal events and the tectonic evolution of the Cobequid Shear Zone.

My contributions as a first author in this chapter, besides the writing of the manuscript, include the analytical work, mineral identification and interpretation, and the development of the ideas regarding the proposed paragenetic sequence of the minerals as well as the correlations with dated hydrothermal events. As a co-author, Georgia Pe-Piper provided the geological maps, samples, additional whole-rock analyses, as well as important information and comments about the geology of the study area that greatly helped the development of a model for the regional hydrothermal system.

An extension of this project includes a detailed petrographic study of fractured granites in the West Moose River pluton in the central Cobequid Highlands. In this pluton the hydrothermal precipitation of REE was studied in specifically selected fractured granites, and was correlated with the formation of associated late hydrothermal Fe-oxyhydroxides. This study has been published as "*Pe-Piper, G., Wisen, J., Papoutsa, A., Piper, D.J.W., 2015. Mineralisation of fractured granites along the Cobequid Fault Zone: West Moose River Pluton, Cumberland Co., Nova Scotia; Geological Survey of Canada, Open File 7759, 252 p."* but is not included in this thesis.

1.4.5 CHAPTER 5

Besides the geochemical and mineralogical assessment of the A-type granites in an attempt to determine their genesis, it is equally important to understand the evolution of the lower crust since it played an important role in the petrogenesis of these rocks in terms of magmatic sources and crustal structure (faults and formation of deep crustal hot zones). Therefore the geological evolution of the lower crust must be evaluated as well. This chapter is entitled "*An evolutionary history of Avalonia: new insights from the Cobequid Highlands*". The data and the concepts of the petrogenesis of the A-type granites developed in this thesis can be used to constrain the geodynamic evolution of the deep crust in the Cobequid Highlands. The contrasting character of the Neoproterozoic basement in the two crustal blocks of the area is inherited by the late Paleozoic granites. In this chapter, the known geochemistry and stratigraphy of the Neoproterozoic basement in the Cobequid Highlands is reviewed and compared with other parts of the Avalon terrane. The correlations made between different areas are then used for paleogeographic and geodynamic reconstructions of Avalonia over the last billion years. Although the

modelling presented here may be a subject for future study and will be refined, it has implications for the early evolution stages of this terrane, and the current concept of "proto-Avalonia".

CHAPTER 2 GEOCHEMICAL VARIATION OF AMPHIBOLES IN A-TYPE GRANITES AS AN INDICATOR OF COMPLEX MAGMATIC SYSTEMS: WENTWORTH PLUTON, NOVA SCOTIA, CANADA

PREFACE

A version of this chapter has been published as: "Papoutsa, A., and Pe-Piper, G., 2014, Geochemical variation of amphiboles in A-type granites as an indicator of complex magmatic systems: Wentworth pluton, Nova Scotia, Canada: Chemical Geology, v. 384, p. 120-134".

2.1 ABSTRACT

The Wentworth pluton is the most complex intrusion among a series of A-type granitic plutons emplaced along an active shear zone in the Cobequid Highlands of Nova Scotia during the latest Devonian-earliest Carboniferous. This pluton consists of A-type granites of different generations and a large gabbroic body. Among all late Paleozoic plutons in the Cobequid Highlands, only the Wentworth granites contain both primary calcic (edenite, hornblende) and sodic (arfvedsonite) amphiboles. Whole-rock and mineral chemical data were examined along with estimated magmatic parameters, such as temperature, emplacement pressure, oxygen fugacity and volatile contents, in order to understand the geological factors responsible for this mineralogical variation. The granites with calcic amphiboles show systematically lower \mathcal{E}_{Nd} values, magmatic temperatures, and F-in-melt contents. All granites, however, appear geochemically homogeneous with limited evidence of fractionation. Specific chemical differences between these two types of granites indicate the presence of two coeval but distinct granitic systems in the Wentworth pluton. The granites with the calcic amphiboles formed from a geochemically less evolved, hydrous, relatively calcic melt, whereas the

granites with sodic amphiboles were derived from a relatively drier sodic melt, which had a larger mantle component and was enriched in F. Coexistence of these two magmas is indicated by the presence of interstitial sodic-calcic amphiboles in a late granitic dyke. This study provides evidence that in A-type granites extreme variation in the type of amphibole can be the result of a very complex magmatic history.

2.2 INTRODUCTION

Amphiboles are the major ferromagnesian phase in many igneous rocks. In Atype granites, the presence of amphiboles not only indicates derivation from hydrous melts but also is considered to play an important role in the geochemical evolution of the parent magma (Martin, 2007). Compositional variation of amphiboles in felsic rocks has been interpreted as the result of either magmatic crystallization or oxidative reequilibration (Strong and Taylor, 1984). In peralkaline rocks, in particular, the magmatic trend of amphiboles is represented by a continuous change from katophorite through richterite to arfvedsonite, whereas the post-solidus oxidative trend is from actinolite towards riebeckite (Table 2.1).

Even though sodic amphiboles appear to be the dominant ferromagnesian minerals in many A-type granites (Wang et al., 2001; Schmitt et al., 2002; Ogunleye et al., 2005; Kinicky et al., 2011), calcic amphiboles are also present in several A-type granites (Han et al., 1996; Dall'Agnol et al., 1999; Wu et al., 2002). The presence of both sodic and calcic amphiboles in A-type granites in the same area has been reported in the literature (Han et al., 1997; Wu et al., 2002). In most cases the sodic amphiboles are found either in different plutons than the calcic amphiboles and belong to different ages

and tectonic regimes or they are found in the same intrusion as the result of fractionation of calcic-amphibole-bearing syenites (Han et al., 1996; Vlach and Gualda, 2007).

Amphibole mineral	Type	Chemical formula
Ferro-edenite	Calcic	NaCa ₂ Fe ²⁺ ₅ Si ₇ AlO ₂₂ (OH) ₂
Ferrohornblende	Calcic	$\Box Ca_2(Fe^{2+4}(Al,Fe^{3+}))Si_7AlO_{22}(OH)_2$
Magnesiohornblende	Calcic	$\Box Ca_2(Mg_4(A1,Fe^{3+}))Si_7AlO_{22}(OH)_2$
Ferro-actinolite	Calcic	$\Box Ca_2Fe^{2+}_5Si_8AlO_{22}(OH)_2$
Actinolite	Calcic	$\Box Ca_2(Mg,Fe^{2+}{}_{51}Si_8AlO_{22}(OH)_2$
Richterite	Sodic-calcic	Na(Ca,Na)Mg ₅ Si ₈ O ₂₂ (OH) ₂
Ferro-richterite	Sodic-calcic	Na(Ca,Na)Fe ²⁺ ₅ Si ₈ O ₂₂ (OH) ₂
Ferro-winchite	Sodic-calcic	\Box (Ca,Na)Fe ²⁺ ₄ (Al,Fe ³⁺)Si ₈ O ₂₂ (OH) ₂
Katophorite	Sodic-calcic	Na(Ca,Na)Fe ²⁺ 4(Al,Fe ³⁺)Si ₇ AlO ₂₂ (OH) ₂
Barroisite	Sodic-calcic	□(Ca,Na)Mg ₃ AlFe ³⁺ Si ₇ AlO ₂₂ (OH) ₂
Arfvedsonite	Sodic	$NaNa_2(Fe^{2+}_4Fe^{3+})Si_8O_{22}(OH)_2$
Ferro-eckermannite	Sodic	NaNa ₂ (Fe ²⁺ ₄ Al)Si ₈ O ₂₂ (OH) ₂
Riebeckite	Sodic	$\Box Na_2(Fe^{2+}{}_3Fe^{3+})Si_8O_{22}(OH)_2$
Notes: □= vacancy in cr	ystallographic site.	

Table 2.1 Ideal chemical formulae of the types of amphibole mentioned in this work.

The Wentworth pluton in the Cobequid Highlands, in mainland Nova Scotia, belongs to a series of plutonic bodies which were emplaced along an active shear zone in the late Paleozoic (Pe-Piper and Piper, 2003). All late Paleozoic granites of the Cobequid Highlands have A-type affinities and similar major oxides compositions. The Wentworth pluton, however, is the only composite intrusion hosting both primary sodic and calcic amphiboles in different granitic bodies within the pluton (Pe-Piper, 2007). Previous mapping (Koukouvelas et al., 2002) showed that coeval sodic amphibole-bearing and calcic amphibole-bearing granites do not show a systematic distribution within the pluton. Contrasting types of granites have been described in the literature from alkaline provinces. Gualda and Vlach (2007) reported the two mineralogically different, alkaline and aluminous associations of the felsic rocks in the Graciosa Province in Brazil. In the alkaline association, calcic amphibole occurs only in the alkali-feldspar and alkalifeldspar-quartz syenites, whereas peralkaline alkali-feldspar granites present a wide range of amphibole compositions, from sodic-calcic to sodic, and are related to the syenites through magmatic differentiation. The aluminous petrographic association contains metaluminous biotite granites with only calcic amphiboles in the less evolved varieties and with a narrow compositional range, suggesting contrasting sources between the two associations.

The amphiboles of the Wentworth pluton were studied by Pe-Piper (2007). The magmatic amphiboles include arfvedsonite (sodic) and edenite (calcic), whereas secondary amphiboles are represented by riebeckite (sodic) and actinolite (calcic). Even though there is a compositional variation in the amphiboles of this pluton, these minerals are not zoned. The lack of zoning was considered evidence of limited fractionation of the parental magma (Pe-Piper, 2007).

It is not clear why the Wentworth granites present such an extreme chemical variation of magmatic amphiboles, in a pluton with limited evidence of magmatic differentiation. Several factors can control the type of amphibole that will form, such as the composition of the magma, oxygen fugacity, temperature, pressure, and the presence or type of volatiles (Martin, 2007). The purpose of this study is to examine and evaluate the role of each of these factors on the composition of the amphiboles present, and

thereby provide insight into the geological history of this distinct and complex magmatic system.

2.2.1 REGIONAL GEOLOGY

The late Paleozoic Cobequid Highlands provide a record of a large scale alkaline magmatic event, which is spatially associated with strike-slip motion on the main faults of the coeval Cobequid Shear Zone (Pe-Piper and Piper, 2003). Underplating of mafic melts beneath the centre of the Magdalen Basin during the late Paleozoic has been inferred from seismic refraction and gravity data (Marillier and Reid, 1990) and is believed to be associated with the magmatism in the Cobequid Highlands (Piper et al., 1993). The underplating is spatially related to distributed regional extension and crustal thinning in the late Devonian and early Carboniferous between bounding master faults in southern New Brunswick in the west and in the Cobequid Shear Zone, western Cape Breton Island, and Newfoundland in the east (Fig. 2.1A) (Hibbert and Waldron, 2009).

A number of plutonic bodies and chemically equivalent extrusive rocks were emplaced along active faults of the Cobequid Shear zone within the Cobequid Highlands. These plutons, which vary in size and complexity, consist of granitic rocks with various amounts of coeval gabbro (Pe-Piper and Piper, 2003). The granites are mostly alkalicalcic to alkalic (using the classification of Frost et al., 2001), with geochemistry resembling A-type granites, whereas the mafic rocks are ferro-gabbros. The coeval volcanic rocks are rhyolites with chemistry similar to A-type granites and basalts with continental tholeiite chemistry (Pe-Piper and Piper, 2003). The Wentworth pluton is the largest and most complex of the plutons in the Cobequid Shear Zone, and is located in the eastern part of the Cobequid Highlands along with the nearby Pleasant Hills pluton (Fig.

2.1B). The Wentworth pluton is bounded by the Rockland Brook fault to the south, and consists of a major gabbroic body to the southwest, passing through a complex zone with gabbro and granite bodies to a granitic zone to the northeast. The gabbro is younger (357±4 Ma Ar-Ar on hornblende, Pe-Piper et al., 2004; Murphy et al., 2011) than the other mafic intrusions in the Cobequid Highlands and based on cross-cutting relationships, post-dates the emplacement of the main Wentworth granite (362±2 Ma U-Pb on zircon, Doig et al., 1996).



Figure 2. 1 Simplified maps (A) of eastern Canada showing the location of the Avalon and Meguma terranes and the Cobequid-Chedabucto fault zone and (B) the distribution of igneous phases and sample locations in the Wentworth Pluton (modified from Koukouvelas et al. 2002). Open symbols represent early granites, half-filled symbols are used for syn-gabbro granites, and post-gabbro granites are shown in filled symbols.

The late Paleozoic granites of the western part of the Cobequid Highlands are of

similar age to the main Wentworth granite (Doig et al., 1996; Dunning et al, 2002).

These granites contain biotite as their ferromagnesian phase, whereas the Pleasant Hills and Wentworth granites contain additional amphibole.

2.2.2 PREVIOUS WORK ON THE WENTWORTH GRANITES

The A-type granites of the Wentworth pluton, based on their field relationship with the gabbro, are classified either as early (predating the gabbro) or late granites (synchronous with or post-dating the gabbro). Syn-gabbro granites show lobate, mutually chilled contacts with gabbro, but previous studies have shown little evidence for geochemical exchange between the gabbro and the granite (Pe-Piper, 1998). Hybrid rocks formed by mixing and mingling of the mafic and felsic magmas are relatively uncommon. Mafic enclaves are common in the late granites, which commonly have sharp linear contacts with gabbro.

The early granite of the Wentworth pluton is mainly a K-feldspar-rich, mediumgrained granite that contains sodic and calcic amphibole, has an average of 76% SiO₂, and relatively high F (up to 1600 ppm) (Koukouvelas et al., 2002; Pe-Piper, 2007). A few granites, interpreted as synchronous with the gabbro, occur as pods within the gabbro and are geochemically distinct, with a wide range of SiO₂ contents and have somewhat higher TiO₂ contents than the alkali granites of the early phase (Koukouvelas et al., 2002). The syn-gabbro and post-gabbro granites have lower concentrations of F than the early granites, a geochemical feature that is also reflected in the composition of the amphiboles. These subtle differences in the geochemistry of the late granites were interpreted by Koukouvelas et al. (2002) to have resulted from partial melting and remobilization of the early granite by heat from the gabbro.

2.3 SAMPLES AND METHODS

2.3.1 SAMPLE SELECTION AND CLASSIFICATION

The samples described in this study were selected on the basis of relative age, as deduced from field relationships, the type of amphibole present, and the availability of chemical and mineralogical data. The nine selected samples include two early granites (7814 and 4636), two syn-gabbro granites (6490 and 5056), and five post-gabbro granites (4641, 7658, 7710, 6419 and 6518). These samples are located on the 1:50000 scale geological maps of Pe-Piper and Piper (2005). Except for sample 5056, which appears to be of hybrid origin, all granite samples were collected away from gabbro contacts and their geochemistry shows no evidence of mixing with gabbro.

2.3.2 RADIOGENIC ISOTOPE ANALYSES

With the exception of sample 7814 for which Sm-Nd isotope data are reported in Pe-Piper and Piper 1998, all other samples have been analyzed for Sm-Nd isotopes for this study by Activation Laboratories Ltd. Rock powders for Sm-Nd isotope analyses were dissolved in a mixture of HF, HNO₃ and HClO₄. Before the decomposition sample was totally spiked with ¹⁴⁹Sm-¹⁴⁶Nd mixed solution. Rare earth elements were separated using conventional cation-exchange techniques. Samarium and Nd were separated by extraction chromatography on HDEHP covered Teflon powder. Accuracy of the measurements of Sm and Nd contents is $\pm 0.5\%$. The ¹⁴³Nd/¹⁴⁴Nd ratios are relative to the value of 0.511860 for the La Jolla standard. Analyses were performed on Triton-MC massspectrometer.

2.3.3 OPTICAL MINERALOGY

The samples presented in this study were examined under the petrographic microscope, scanning electron microscope (SEM) and electron microprobe (EMP) for the determination of their mineralogy. The microscope used is a Nikon Eclipse E400 POL. Micro-pointcounting of the samples was performed using an automated stepping stage of Conwy Valley Systems and the acquired data were processed with the use of PETROG software for Windows. The density of the pointcounting grid and the distance between incremental movements of the stepping stage are automatically calculated by the operating program, based on the total surface of the thin section and the number of points to be counted. Selection of the number of points is based on the coarseness of the rock and the desired level of precision. Based on the grain size of these rocks (medium- to coarse-grained granites) a minimum of 300 points should be counted. The point counting runs in this study were, done with 300 and 1000 points. The difference in the occurrence of common minerals from the two runs was less than 2%, however, only the results of the 1000-point run are reported (Table 2.3), since the precision of the method increases with the number of points counted. No reference rocks were examined for comparison, however the results herein are comparable to those from other samples from the same pluton reported in Pe-Piper (1998), for which macro-pointcounting on stained slabs was employed.

2.3.4 EDS AND WDS CHEMICAL ANALYSES

All polished thin sections were carbon-coated and analyzed by energy dispersive spectroscopy (EDS). For these analyses, a LEO 1450 VP SEM scanning electron microscope was used, with a maximum resolution of 3.5 nm at 30 kV. The EDS, semi-

quantitative analyses were performed using an Oxford X-mac 80 mm² silicon-drift detector (SDD) with the X-ray production radius being less than 10 microns. Wavelength dispersive spectroscopy (WDS) geochemical analyses of the ferromagnesian minerals present, such as amphibole and biotite, were made using a JEOL-8200 electron microprobe with five wavelength spectrometers and a Noran 133 eV energy dispersion detector. The operating conditions were 15kV of accelerating voltage with a 20nA beam current, and a beam diameter of 1 micron. The crystals used for the analyses were the LIFH for Mn and Fe, PETJ for K, Cl, Ca and Ti, TAP for Al and Si, and the TAPH for F, Na and Mg. The standards used for the calibration of the instrument were sanidine (Al, K, and Si), pyrolusite (Mn), fluorapatite (F), kaersutite (Ca, Mg and Ti), tugtupite (Cl), jadeite (Na), and garnet (Fe). The raw data were corrected with the ZAF method for matrix effects (Armstrong, 1988). Calculation of amphibole formulas from electron microprobe data is challenging because of uncertainty in the oxidation state of Fe and Mn, as recently reviewed by Giesting and Filiberto (2014) and Locock (2014). Various normalization procedures are imprecise compared to the measured values of Fe3+ and Fe2+, but better than the default assumption of all Fe being either Fe2+ or Fe3⁺ (Hawthorn and Oberti, 2007). The calculations follow Leake et al. (1997). All cations were initially normalized on the basis of 23 oxygens producing a formula where all iron is assumed to be ferrous. According to Leake et al. (1997), ferric iron may be estimated if the following conditions are met: a) Si \leq 8 a.p.f.u., b) $\Sigma cat \leq 16$, and c) Σcat (excluding Na and K) \leq 15 a.p.f.u. All these conditions are met for the amphibole analyses included in this study. The most widely used cation normalization procedure for estimating Fe3+ (Hawthorn and Oberti, 2007) is the 13eCNK method, in which 13 cations excluding Ca,

Na and K are assumed to fill the tetrahedral and octahedral sites. The 13eCNK method is specifically applicable to calcic and sodic amphiboles of the type found in this study (Giesting and Filiberto, 2014). This procedure was carried out using MINPET software for Windows (Richard, 1995). Amphibole nomenclature follows Leake et al. (1997); all names used in this study are retained in the newer nomenclature of Hawthorne et al. (2012).

2.3.5 ESTIMATE OF INTENSIVE CRYSTALLIZATION PARAMETERS

The temperature and pressure for the Wentworth granites have been estimated by the use of the zircon-saturation thermometer (Boehnke et al., 2013, Table 2.2), Al-inhornblende thermobarometer (Anderson and Smith 1995, Table 2.2), and the amphibole thermometer of Ridolfi and Rinzulli (2012) (Table 2.2). This new zircon-saturation thermometer is a revised version of the Watson and Harrison (1983) calibration. This thermometer works under the principle that zircon solubility in crustal magmas is a simple function of temperature, and composition of the melt (both in terms of Zr, and alkalinity represented as the M-value). The use of this method requires the presence of magmatic zircon in the rocks, a composition of the melt that is within the calibration range (M-value from 1 to 2.2), and an accurate measurement of Zr in the melt (Boehnke et al., 2013). In the absence of in situ analyses on volcanic glass or melt inclusions, an approximation of the melt composition is through whole-rock analysis (Hanchar and Watson, 2003). Saturation of zircon in the Wentworth granites is indicated by the presence of magmatic zircon. Zircon grains in the Wentworth granites are found both as inclusions on major minerals and in between grains. Watson (1979) demonstrated that most magmatic accessory minerals tend to form along grain boundaries. The occurrence

of zircon, on the other hand, as inclusion in major minerals may be related to growth occlusion. The presence of magmatic zircon in the Wentworth granites is not only texturally inferred, but also suggested by geochronological data that demonstrate a zircon crystallization age similar to those for dated amphiboles (Dunning et al., 2002). Furthermore, based on their whole-rock composition, the Wentworth granites are mostly alkalic-calcic to alkalic rocks and well within the calibration range of this method, and thus the application of the zircon-thermometer was considered appropriate.

The hornblende thermometer of Rifolfi and Renzulli (2012) is used to determine the temperature at which the calcic amphiboles of the Wentworth granites crystallized. This internally consistent method relies solely on amphibole chemistry for which a series of thermobarometric equations are applied to retrieve not only temperature, but also pressure and water contents of the melt. The parameters of this method require the recalculation of the amphibole formula using the 13 cation (excluding Ca, Na and K) normalization. Temperature and pressure estimates may be considered reliable if they plot within the defined fields in the T-P diagrams proposed by Ridolfi and Renzulli (2012) (Appendix A.1). These equations are only applicable to euhedral, homogeneous crystals from either volcanic or plutonic rocks and are not recommended for hydrothermal amphiboles or disequilibrium compositions. As it is described in the next section, the calcic amphiboles of the Wentworth pluton satisfy both textural and chemical criteria of this method.

For the Al-in-hornblende thermobarometer to be applied, the mineral assemblage alkali feldspar + plagioclase + hornblende + biotite + iron oxide + titanite must be present and that requirement is satisfied in the granites of the Wentworth pluton. The hornblende

grains chosen had Al₂O₃ > 6 wt. %, more than 1.0 Al_(total) a.p.f.u., and are in contact with quartz or K-feldspar. Anderson and Smith (1995) argued that Fe# is indicative of oxygen fugacity. Amphiboles with high Fe# (Fe_T/(Fe_T+Mg)>0.65) are indicative of low oxygen fugacity, and should be avoided for geobarometric calculations because they tend to yield anomalously high pressures by a factor of \geq 2 kbar. Several analyses of calcic amphiboles from the Wentworth pluton, however, show Fe# > 0.65 and were still used for geobarometric calculations. The reasons for this are discussed later in this work. Even though this geobarometer is based on the principle that the Al-content in amphiboles is indicative of pressure through the tschermakite substitution, increasing Al-content can be also a result of temperature due to edenite substitution. To eliminate the influence of temperature in the estimate of pressure, the recalibration of Anderson and Smith (1995) has been used which enables pressure calculations for igneous amphiboles that have been affected by the edenite substitution.

The initial contents of H₂O in the granitic melt were estimated by the plagioclasemelt hygrometer of Al'meev and Ariskin (1996) (Table 2.2). The hygrometer estimates the initial H₂O contents of a melt as a function of pressure, temperature and chemical composition. The authors developed this model primarily for mafic rocks, however the experimental database that was used as an input included not only basaltic and andesitic compositions but also 49 analyses of granitic rocks for a total range of pressure between 1 and 3.5 kbar and temperatures from 800 to 1300 °C. The equation includes a correction for the dependence of water solubility on temperature, which is different between basaltic and granitic systems. Therefore, the produced equations may be applied to both mafic and felsic systems (Al'meev and Ariskin, 1996). The error of the estimated H₂O contents

is up to 0.3 wt. %. The estimated H₂O contents were also compared with those derived from the plagioclase-melt equilibrium hygrometer of Putirka (2005), where applicable, for a better evaluation of the estimates. The plagioclase-melt hygrometer is based on the anorthitic-albitic exchange of the melt and estimates the magmatic H₂O content, with an error of 1.4 wt. % (Putirka, 2005). For the estimate of the initial magmatic contents of F, the equation of Eby and Charnley (2010) was used (Table 2.2). According to this equation, the concentration ratio D of F between the melt and crystallizing amphiboles is proportional to the Mg number of the latter (Table 2.2). The concentration ratio is the amount of the element measured in the mineral divided by the concentration of that element in the melt. From the estimated concentration ratios and the chemical analyses of the amphiboles, the initial concentrations of fluorine in the melt were thus calculated. The viscosities of the granitic melts were estimated by the empirical model of Giordano et al. (2008) for granitic melts (Table 2.2). The estimates for calcic granites are within an error of ± 0.35 log units whereas viscosities for sodic granites have an associated error of \pm 0.25 log units.

2.4 **RESULTS**

The majority of the studied samples plot within the alkali-feldspar granite field of the modal composition diagram of Streckeisen (1974) with a few exceptions (Fig. 2.2). The textures vary from granular to granophyric and the grain sizes range from fine to coarse grained (Table 2.3). The granites with sodic amphiboles are coarser grained and contain only perthitic K-feldspar, with the composition of orthoclase, indicative of hypersolvus granites. The perthitic albite is almost pure sodic feldspar with a composition of

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Estimate	Equation	Reference	Notes
Zircon- saturation thermometer	$T{=}(10108 \pm 32) \{ \text{In } D_{Zr} + (1.16 \pm 0.15) * (M{-}1) + (1.48 \pm 0.09) \}$	Boehnke et al. (2013)	T is temperature in $^{\circ}$ K, M is the cationic ratio of the melt: (Na+K+2Ca)/(Al*Si). D _{2r} is the concentration ratio of zirconium in zircon over that of the melt, in ppm.
Al-in- hornblende barometer	P= 4.76Al -3.01 -{[T-675]/85} *{0.530 Al+ 0.005294 *(T-675)}	Anderson and Smith (1995)	T is temperature in °C, Al is the total Al (a.p.f.u.) in hornblende.
Hbl thermometer	T= 17098 +1322.3Si+1035.1Ti +1208.2Al+1230.4Fe + 1152.9 Mg+ 130.4 Ca+200.54 Na + 29.408K + 12.410 lnP	Ridolfi and Renzulli (2012)	T is temperature in °C, Pressure is given by the equation lnP = 125.93 + 9.5876Si + 10.116Ti + 8.1735Al + 9.2261Fe + 8.7934Mg + 1.6659Ca + 2.4835Na + 2.5192K. Elements are in a.p.f.u. on the basis of 13 cations excluding Ca,Na and K.
Water-in- melt	$\begin{split} H_2O(wt.\%) &= 24.757 - 2.26 * 10^{-3} * T* \\ ln[An^{plag}/(Ca^{melt} (Al^{melt})^2 * (Si^{melt})^2)] - \\ 3.847* Ab^{plag} + 1.927* \\ [An^{plag}/(Ca^{melt}/(Ca^{melt} Na^{melt}))] \end{split}$	Putirka (2005)	Temperature is in °K, superscripts indicate the cationic fractions of the melt and the plagioclase.
	$lnH_2O=(4.39 \pm 1.5) + [(38438.3 \pm 5402.4) * Si/O- (14710.2\pm 2142.9)] /T + (0.59 \pm 0.01) * lnP- (21.45 \pm 3.8) * Si/O + (3.89 \pm 0.69) * Al/Si$	Al'meev and Ariskin (1996)	$\mathrm{H_2O}$ in wt%, T in °K, P in bars and elements in atom fractions.
F-in-melt	$D^{Famph/Fmelt} = 1.03 + 0.06 * Mg\#$	Eby and Charnley (2010)	D is the concentration ratio of fluorine in amphibole over the amount of fluorine in the melt. Mg#: (Mg/(Mg+Fe)*100, elements in a.p.f.u.
Viscosity	logn= -4.55 + B/(T-C)	Giordano et al. (2008)	η in Pascal seconds (Pa*s), T in °K, B and C parameters are calculated from whole-rock composition with oxides normalized as mol%.

Ab₉₆ to Ab₁₀₀ (Appendix A.2). The granites with calcic amphiboles are finer grained and contain discrete grains of both plagioclase and K-feldspar, characteristic of subsolvus granites. The plagioclase has a composition that ranges from Ab₇₀ (andesine) to Ab₉₆ (albite) (Appendix A.1). Biotite is present in various amounts in many samples with calcic amphiboles, but is completely absent from granites with sodic amphiboles. Most of the studied samples contain magnetite as the dominant iron oxide phase, but a few have ilmenite (Table 2.3).

2.4.1 AMPHIBOLE COMPOSITION

The amphibole-group minerals have a double silicate chain structure and a generic chemical formula of $AB_2C_5^{VI}T_8^{IV}O_{22}(OH)_2$ (Leake et al., 1997). The A-site contains primarily alkalies, whereas the B-site is occupied by Ca and any excess K and Na. Divalent cations such as Mg, Fe^{2+} , Mn^{2+} and Li are distributed between the B- and C-sites (Table 2.1). The trivalent cations such as Al, Fe^{3+} , Mn^{3+} and Cr^{3+} are accommodated in the C-site. The high valency cations such as AI^{4+} and Si^{4+} are hosted in the T-site with any excess Ti, whereas the F and Cl anions substitute for OH in the amphibole structure. According to the Leake et al. (1997) nomenclature scheme, all calcic amphiboles must have Ca+Na >1 a.p.f.u. in the B-site with Na being less than 0.5 a.p.f.u. The sodic-calcic amphiboles have, in general, Ca+Na>1 a.p.f.u. in the B-site (>1.5 a.p.f.u.) and alkalies in the A-site (>0.5 a.p.f.u.) are indicative of sodic amphiboles (Table 2.1). A complete set of EMP analyses of amphiboles from the studied samples is included in Appendix A.3.



Figure 2. 2 Modal classification of the studied granitic samples (after Streckeisen, 1974). Squares indicate samples with calcic amphiboles and circles indicate the presence of sodic amphiboles. A: alkali feldspar, Q: quartz, P: plagioclase.

2.4.1.1 Early granites

The early granites contain magmatic arfvedsonite (sample 4636, Fig. 2.3A), and ferro-edenite and ferro-hornblende (sample 7814, Fig. 2.3B) (Tables 2.3, 2.4). The ferroedenite occurs as subhedral grains and contains inclusions of magnetite. Ferrohornblende forms either discrete subhedral grains or patches in ferro-edenite. Calcic amphiboles are commonly altered along cleavage planes to epidote, titanite, chlorite, actinolite, calcite and in some samples, secondary biotite (7814, Fig. 2.3B). Arfvedsonite, on the other hand, occurs as distinct grains that can reach 400 microns in diameter (Fig. 2.3A). The sodic amphibole contains inclusions of magnetite and is associated with secondary fluorite. The analyses from core-to-rim traverses suggest that the grains are relatively homogenous with no zoning, except for alteration products along the rims that include ferro-eckermannite, riebeckite and chlorite (4636).

2.4.1.2 Syn-gabbro granites

Two studied syn-gabbro granites contain calcic amphiboles represented by Mgrich ferro-hornblende (5056, Fig. 2.3C), and ferro-edenite (6490, Fig. 2.3D). Sample 5056, which has been interpreted as hybrid, chemically resembles granodiorite and is one of the samples that plot within the granite field instead of the alkali feldspar granite field (Fig. 2.2) of Streckeisen (1974). This sample is the only one containing Mg-rich amphiboles and no ferro-edenite (Table 2.4). The dominant amphibole is Mg-rich ferrohornblende with patches of magnesiohornblende along surfaces of weakness such as microfractures and cleavage (Fig. 2.3C). The Mg-rich ferro-hornblende forms large, discrete, subhedral to euhedral grains in contact with Fe-oxides and biotite. The primary amphibole in sample 5056 has altered to chlorite, biotite, titanite, epidote, calcite and actinolite (Fig. 2.3C). Ferro-edenite in sample 6490 occurs as medium to small, subhedral to anhedral grains.

2.4.1.3 Post-gabbro granites

The post-gabbro granites contain ferro-edenite, ferro-hornblende (samples 4641, 7658, 7710), ferro-richterite, ferro-winchite (sample 7710, Fig. 2.3E), and arfvedsonite (sample 6419) (Table 2.3). Like the early and syn-gabbro granites, the calcic amphiboles alter to titanite, epidote and actinolite. Sample 7710 comes from a late equigranular, fine-grained granitic dyke that intruded the Wentworth gabbro, and it is somewhat different from most samples. It contains a wide variety of amphiboles calcic ferro-edenite to sodic-calcic ferro-richterite and ferro-winchite (Table 2.4).



Figure 2. 3 Selected backscattered electron (BSE) images of amphiboles from the Wentworth granites. Figures (a) and (b) are from early granites with arfvedsonite (Arf), and ferro-edenite (Feed) which is altered to titanite (Tnt) and chlorite (Chl). Figure (c) is an Mg-rich ferro-hornblende (Fe-hbl) from a hybrid rock altered to magnesiohornblende (Mg-hbl) and actinolite (Act), and (d) is a euhedral ferro-edenite from a syn-gabbro granite. Figure (e) is an interstitial, sodic-calcic amphibole (Fe-richt) and (f) a megacrystic riebeckite (Rbk) from post gabbro granites. Abbreviations after Whitney and Evans (2010). Numbers indicate positions of analyses.

Furthermore, it is the only sample in which ferro-richterite has patches of hedenbergitic

augite, mainly along cleavage planes. Primary ferro-edenite and ferro-richterite occur as

interstitial, subhedral grains, and contain inclusions of Fe-oxides. Secondary riebeckite,

katophorite and ferro-winchite have formed by the alteration of ferro-richterite. The primary sodic amphibole in sample 6419 is arfvedsonite, occurring as megacrysts with the same alteration products as the sodic amphiboles in other samples. In this sample, arfvedsonite appears more altered compared to other samples and secondary ferro-eckermannite is more abundant. Sample 6518, however contains principally riebeckite with smaller amounts of ferro-winchite and minor actinolite. In this sample, riebeckite occurs in two forms: a) megacrystic riebeckite with inclusions of ilmenite (Fig. 2.3F) and b) slender, independent prisms. The chemical composition of the riebeckitic megacrysts and the prisms is identical.

In all samples, calcic amphiboles like ferro-edenite or ferro-hornblende have not been observed to coexist with primary sodic amphiboles such as arfvedsonite. Furthermore, the analyses from the Wentworth amphiboles show that arfvedsonite has almost double the amount of F compared to ferro-edenite and ferro-hornblende (Table 2.4).

2.4.2 INTENSIVE CRYSTALLIZATION PARAMETERS

With the exception of one sample (7814), all the calculated zircon-saturation temperatures for the studied samples from the Wentworth Pluton appear higher than 850 ± 50 °C (Table 2.3), within the expected range for A-type granites (Clemens et al., 1986). The samples with sodic amphiboles have higher zircon-saturation temperatures than those with calcic amphiboles (Table 2.3). The magmatic temperatures of calcic amphiboles do not vary significantly from those estimated from zircon saturation for the parental rock and all are greater than 760 °C ± 50 °C (Table 2.3).

The Wentworth calcic amphiboles show evidence for both edenite and tschermakite substitutions (Fig. 2.4), and therefore, a temperature correction has been applied when

using the hornblende Al content to estimate pressures, as mentioned above. The pressure estimated by the Al-in hornblende barometer for the Wentworth granites ranges between 2.1 and 3.2 ± 0.6 kbar (Table 2.2).

2.4.3 OTHER MAGMATIC PARAMETERS

The dominance of magmatic magnetite indicates that these granites belong to the magnetite series granites, which formed under oxidizing conditions (Table 2.3). The hygrometric calculations for the Wentworth granites suggest that the calcic amphibolebearing samples formed in a magma richer in H₂O (4–11.6 wt. % H₂O), whereas the samples with sodic amphiboles were derived from a drier magma (3.6–1.3 wt. % H₂O, Table 2.3). There is no correlation between the estimated pressures and the calculated H₂O for these granites to suggest a large influence of PH₂O on the pressure estimates.

The amount of F in the granitic melt of the Wentworth pluton ranged from 0% to 0.8 wt.%. The samples with calcic amphiboles had smaller amounts of F in the melt than the samples with sodic amphiboles (Table 2.3). The estimated viscosities show that there are systematic variations between the two types of granites. The granites with calcic amphiboles were derived from a melt with a higher average viscosity of $10^{9.79\pm0.35}$ Pa*s, whereas the granites with sodic amphiboles show relatively lower viscosities between $10^{7.08}$ and $10^{9.42\pm0.25}$ Pa*s (Table 2.3).

2.5 **DISCUSSION**

The main factors that control the type of amphibole present in the rocks are: a) chemical composition of the melt, b) temperature, c) pressure, d) oxygen fugacity, and e)



Figure 2. 4 Edenitic and tschermakitic substitutions for the Wentworth hornblendes. The tschermak exchange shows a better defined linear trend for all calcic amphiboles, demonstrating a larger influence of pressure over temperature on the chemistry of these minerals. The parameter R in tschermak exchange refers to divalent cations (such as Mg and Fe) in octahedral sites.

presence and type of volatiles (Martin, 2007). In this section, the influence of each of these factors is evaluated in an attempt to understand the geological processes responsible for the wide chemical variation observed in the amphiboles of the Wentworth granites.

2.5.1 CORRELATION BETWEEN THE TYPE OF AMPHIBOLE AND WHOLE-ROCK COMPOSITION

Comparison of the types and amounts of amphiboles present with the whole-rock geochemistry of the samples suggests that there is a correlation with MgO. Most of the samples with relatively high MgO have relatively high modal abundances of calcic amphiboles, whereas the samples with low MgO concentrations contain either sodic amphiboles or very small amounts of calcic amphiboles. Barium is used as an indicator of fractionation of the melt. Barium is incompatible in mafic to intermediate melts, however it is highly compatible in evolved magmas (Rollinson, 1993). The hornblende-bearing rocks have significantly higher amounts of Ba than the samples with sodic amphiboles (Table 2.5). Assuming that Ba content reflects the content of the magma and was not affected by subsolidus alteration, these differences suggest a geochemically less evolved parental magma for the rocks with calcic amphiboles.

In silica-oversaturated rocks, the spectrum of amphibole compositions found in extension-related suites could result from either fractional crystallization of a mafic melt or anatexis of suitably metasomatized crust (Martin, 2007). There is no good evidence for fractional crystallization from a mafic melt in the Wentworth granites. The lack of homogeneous intermediate rock-types in the area or chemical zoning (Appendix A.4) in the rock-forming minerals that could suggest a fractionating melt leads to the conclusion that melting of a suitable source in the deep crust was responsible for the Wentworth granites. In that case, Sm-Nd isotopes are considered an effective method to determine the nature of that source.

The Sm-Nd isotopic composition of the Wentworth granites (Pe-Piper and Piper, 1998b, and new data reported here, Table 2.3) shows a range of \mathcal{E}_{Nd} values from +1.1 to

	Sample	4636	7814	5056	6490	4641	6419	6518	7658	7710
	Relative age	Early	Early	Syn- gabbro	Syn- gabbro	Post- gabbro	Post-gabbro	Post-gabbro	Post- gabbro	Post-gabbro
UTM	Easting	455400	463361	463975	461724	452932	451203	460683	468260	450747
	Northing	5048118	5044863	5044528	5043853	5046282	5049916	5044932	5045537	5040441
uo	dz	35.0	37.6	32.4	36.6	32.4	31.0	31.3	25.4	33.6
itieo	plg (An)	0.0	0.4 (An4)	8.8(An ₂₇₋₄)	1.7 (An ₇₋₄)	6.6 (An ₇₋₃)	0.0	0.0	7.7 (An9-5)	0.0
dw	K-fsp	61.0	61.6	43.2	59.8	57.8	62.0	59.7	61.6	62.5
0)	amp	4.0	0.4	7.9	0.7	2.4	6.0	6.7	0.8	3.9
lsbo	bt	0.0	$\overline{\vee}$	7.8	1.2	3.7	0.0	0.0	4.4	0.0
٥Μ	Fe-oxide ¹	mgt 100%	mgt 91%	mgt 88%	mgt 99%	mgt 99%	mgt 99%	mgt 59%	ilm 65%	mgt 92%
	Streckeisen classification	alkali feldspar granite	alkali feldspar granite	granite	alkali feldspar granite	granite	alkali feldspar granite	alkali feldspar granite	granite	alkali feldspar granite
əlo	Primary	Arf (80%)	Ferro- edenite (96%),	Ferro-hbl (65%)	Ferro- edenite (80%),	Ferro- edenite (29%),	Arf (30%)	Rbk** (63%)	Ferro- edenite (88%),	Ferro-edenite (41%), Ferro-
odinq			Ferro-hbl (4%)		Ferro-hbl (15%)	Ferro-hbl (63%)			ferro-hbl (12%)	richt (28%)
ms to sqvT	Secondary	Ferro-eck (16%), rbk (4%)		Mg-hbl (18%), actin (17%)	Ferro-actin (5%)	Ferro-actin (8%)	Ferro-eck (66%), ibk (4%)	Ferro-winch (32%), Ferro- actin (5%)		Ferro-actin (3%), Ferro- winch (6%), rbk (22%)
		sodic	calcic	calcic	calcic	calcic	sodic	sodic	calcic	calcic, sodic- calcic
	Texture	granular	granophyric	granular	granophyric	granular	granular	granular	granular	equigranular
	Grain Size	coarse	coarse	medium- fine	medium	medium	medium- coarse	medium-coarse	medium	medium-fine
	ASI	0.92	1.03	1.15	1.06	1.43	0.98	0.92	1.08	1.00
	MALI	8.89	7.91	6.29	8.49	6.51	8.36	8.51	8.06	8.28

Table 2. 3 Petrographic description of the studied samples.

$ \begin{array}{c c c c c c c c c c c c c c c c c c c $		Sample	4636	7814	5056	6490	4641	6419	6518	7658	7710
		Relative age	Early	Early	Syn- gabbro	Syn- gabbro	Post- gabbro	Post-gabbro	Post-gabbro	Post- gabbro	Post-gabbi
Evention ± 0.5 3.60 1.36* 2.46 1.15 2.52 3.15 3.13 2.24 3.19 Tz, $(°C) \pm 35$ 891 830 884 836 874 919 1045 863 940 Tz, $(°C) \pm 36°$ 1.4 1.7 1.37 1.13 1.42 1.63 1.44 1.45 M-value 1.56 1.4 1.7 1.37 1.13 1.42 1.63 1.44 1.45 Paineber 2.11 2.55 2.8 3.2 1.42 1.63 1.44 1.45 Paineber 1.56 1.4 1.7 1.37 1.13 1.42 1.43 1.45 Paineber 2.11 2.5 2.8 3.2 3.13 3.13 3.13 3.13 Hold met 0.16 0.2 0.00 0.2 0.00 0.6 1.0 0.5 5.6 5.6 5.6 Wt 90 1.03 3.5 1.16 3.5 1.3 <t< td=""><td></td><td>Ba (ppm)</td><td>39</td><td></td><td>699</td><td>89</td><td>599</td><td>32</td><td>12</td><td>276</td><td>125.3</td></t<>		Ba (ppm)	39		699	89	599	32	12	276	125.3
Tz, $(cC) \pm 35$ 891 830 844 836 874 919 1045 863 940 Tz, $(cC) \pm 35$ 891 830 884 836 874 919 1045 863 940 Two wild 1.56 1.4 1.7 1.37 1.13 1.42 1.63 1.44 1.45 Two (*C) \pm 50°C 857 764 821 795 820 887 940 Poliment 2.1 2.3 2.8 3.2 3.2 3.1 3.2 3.1 3		$\epsilon_{Nd(360)} \pm 0.5$ units	3.60	1.36*	2.46	1.15	2.52	3.15	3.13	2.24	3.19
$ \begin{array}{lcccccccccccccccccccccccccccccccccccc$		T_{Zr} (°C) ± 35	891	830	884	836	874	919	1045	863	940
Thu $(^{\circ}C) \pm 50^{\circ}C$ 857 764 821 795 826 887 PALINALI 0.6 D.1 2.1 2.5 2.8 3.2 3.2 3.1 3.1 PALINALI 0.6 D.1 2.1 2.5 2.8 3.2 3.1 1.3 PALINALI 0.6 D.2 0.0 0.2 0.0 0.6 1.0 0.1 0.5 PALINALI 1.0 0.8 0.2 0.0 0.2 0.0 0.6 1.0 0.1 0.5 PALINALI 1.0 0.8 0.2 0.0 0.2 0.0 0.6 1.0 0.1 0.5 PALINALI 1.0 0.8 0.2 0.0 0.2 0.0 0.6 1.0 0.1 0.5 PALINALI 1.0 0.8 0.2 0.0 0.2 0.0 0.6 1.0 0.1 0.5 PALINALI 1.0 0.8 0.2 0.0 0.2 0.0 0.0 0.6 1.0 0.1 0.1 PALINALI 1.0 0.1 0.2 0.0 10.6 1.0 0.6 1.0 0.1 <	SJ	M-value	1.56	1.4	1.7	1.37	1.13	1.42	1.63	1.44	1.45
FAimble Uo blar2.12.52.83.2331.3Finalet Uo t bar0.80.20.00.20.00.61.00.10.5 ± 0.24 ± 0.24 0.80.20.00.20.00.61.00.10.5 ± 0.24 ± 0.24 0.80.20.00.20.00.61.00.10.5 ± 0.24 ± 0.24 3.611.65.010.64.03.51.36.65.6 ± 0.24 (wt%) ± 0.3 3.611.65.010.64.03.51.36.65.6 $H_{2}0$ in met (wt%) ± 1.2 5.56.35.74.03.51.35.64.3Met viscosity (log (Pa*s))9.4210.359.2510.329.309.037.089.738.51	iəşəu	T_{hbl} (°C) ± 50°C		857	764	821	795			826	887
Financi (wt%) 0.8 0.2 0.0 0.2 0.0 0.6 1.0 0.1 0.5 ± 0.24 0.8 0.2 0.0 0.2 0.0 0.6 1.0 0.1 0.5 ± 0.24 0.2 0.0 0.2 0.0 0.5 0.1 0.5 Wt % $= 0.3$ 3.6 11.6 5.0 10.6 4.0 3.5 1.3 6.6 5.6 Wt % Wt % $= 0.3$ 5.7 4.0 3.5 1.3 5.6 4.3 Wt % Wt % $= 5.5$ 6.3 5.7 4.0 5.6 4.3 Wt % 0.42 10.35 9.25 10.32 9.30 9.03 7.08 9.73 8.51	baraı	Pal-in-hbl±0.6 kbar		2.1	2.5	2.8	3.2			б	1.3
F H2O in met (wt%) $\pm 0.3.5$ 3.6 11.6 5.0 10.6 4.0 3.5 1.3 6.6 5.6 wt.% ^a H2O in met (wt%) ± 1.2 5.5 6.3 5.7 4.0 wt.% ^b Melt viscosity 9.42 10.35 9.25 10.32 9.30 9.03 7.08 9.73 8.51	əvienəti	F in-melt (wt%) ± 0.24	0.8	0.2	0.0	0.2	0.0	0.6	1.0	0.1	0.5
H2O in met (wt%) ± 1.2 5.5 6.3 5.7 4.0 5.6 4.3 wt.%b Melt viscosity 9.42 10.35 9.25 10.32 9.30 9.03 7.08 9.73 8.51	I	H2O in mett (wt%) ± 0.3 wt. % ^a	3.6	11.6	5.0	10.6	4.0	3.5	1.3	6.6	5.6
Melt viscosity 0.42 10.35 9.25 10.32 9.03 7.08 9.73 8.51		H2O in met (wt%) ± 1.2 wt. % ^b		5.5	6.3	5.7	4.0			5.6	4.3
		Melt viscosity (log (Pa*s))	9.42	10.35	9.25	10.32	9.30	9.03	7.08	9.73	8.51
		ilmenite, arf: arfve ferro-winch: ferro- cationic proportion by the Al'meev and	edsonite, ferrc -winchite, fer ns, MALI: M d Ariskin (19	 >-eck: ferro-eck rro-richt: ferro-r odified Alkali I 96) model and 	cermannite, rbk: richterite (after Jime Index (Na; ^b are the magmé	riebeckite, ferr Whitney and Ev 20+K2O-CaO) atic water estim	o-hbl: ferroho vans, 2010), A oxides in wt% ates by the use	urnblende, Mg-hbl: SI: Aluminum Satı (Frost et al., 2001 e of the plagioclase	magnesiohornblend uration Index (Al/(N). ^a represents magm melt hygrometer o:	le, ferro-actin: la+K+Ca-1.67 latic water con f Putirka (200:	ferro-actinoli *P) elements tents estimate 5). Pointcoun
ilmenite, arf: arfvedsonite, ferro-eck: ferro-eckermannite, rbk: riebeckite, ferro-hbl: ferrohnblende, Mg-hbl: magnesiohornblende, ferro-actin: ferro-actinoli ferro-winch: ferro-winchite, ferro-richt: ferro-richterite (after Whitney and Evans, 2010), ASI: Aluminum Saturation Index (Al/(Na+K+Ca-1.67*P) elements cationic proportions, MALI: Modified Alkali Lime Index (Na2O+K ₂ O-CaO) oxides in wt% (Frost et al., 2001). ^a represents magmatic water contents estimate by the Al'meev and Ariskin (1996) model and ^b are the magmatic water estimates by the use of the plagioclase-melt hygrometer of Putirka (2005). Pointcoun		based on 1000 poi	ints.				,	1			ĸ

Sample	7814	4641	5056	5056	5056	7710	7710	4636	6419	6419	6518
Analysis	19	190	202	204	233	296	297	174	461	462	25
Mineral	fe-ed	fe-ed	mg-hb	mg-hbl	act	rcht	rcht	arf	arf	ie- eck	rbk
Oxides (w	t.%)										
SiO ₂	44.68	43.58	45.48	45.26	50.48	47.09	46.84	50.53	50.25	48.78	51.65
TiO ₂	0.77	0.69	1.59	1.32	0.28	1.24	1.18	1.30	0.60	1.32	0.43
Al ₂ O ₃	6.01	7.27	6.91	7.48	2.13	3.43	3.56	1.42	1.30	2.36	0.96
FeOt	28.33	26.00	20.80	21.22	17.05	29.85	27.84	32.50	30.66	30.45	35.33
MnO	1.38	0.86	0.49	0.48	0.41	0.71	0.69	0.73	0.72	0.73	0.92
MgO	4.27	4.70	8.96	8.74	12.81	2.69	4.10	1.28	2.66	2.19	0.45
CaO	10.50	10.79	10.69	10.58	11.85	7.80	8.73	2.17	2.67	3.20	1.61
Na ₂ O	2.18	1.76	1.76	1.51	0.63	3.86	3.75	7.60	7.56	7.30	6.23
K ₂ O	1.18	1.13	0.86	0.92	0.24	1.07	1.09	1.35	1.28	1.31	0.32
F	0.71	0.35	0.23	0.17	0.23	0.58	1.01	1.06	1.38	1.15	0.00
Cl	0.30	0.79	0.15	0.18	0.07	0.07	0.07	0.04	0.04	0.03	0.00
Total	100.3	97.9	97.90	97.86	96.17	98.39	98.84	99.98	99.10	98.82	0.00
Si	6.951	6.897	6.877	6.822	7.551	7.499	7.436	7.925	7.924	7.744	7.984
Al	1.049	1.103	1.123	1.178	0.376	0.501	0.564	0.075	0.076	0.256	0.016
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
Al	0.053	0.253	0.106	0.150	0.000	0.142	0.101	0.187	0.165	0.185	0.159
Fe ³⁺	0.425	0.259	0.511	0.694	0.358	0.000	0.000	0.269	0.298	0.157	1.296
Ti	0.090	0.082	0.181	0.149	0.032	0.149	0.141	0.153	0.071	0.158	0.049
Mg	0.990	1.109	2.020	1.965	2.856	0.638	0.970	0.299	0.625	0.518	0.104
Fe ²⁺	3.261	3.183	2.119	1.981	1.702	3.975	3.695	3.995	3.744	3.885	3.271
Mn	0.181	0.116	0.063	0.061	0.052	0.096	0.093	0.097	0.096	0.097	0.120
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Ca	1.750	1.830	1.732	1.709	1.899	1.331	1.485	0.365	0.451	0.544	0.266
Na	0.250	0.170	0.268	0.291	0.101	0.669	0.515	1.635	1.549	1.456	1.734
B-site	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.408	0.371	0.247	0.149	0.083	0.523	0.638	0.677	0.764	0.790	0.133
K	0.233	0.227	0.166	0.176	0.046	0.217	0.220	0.271	0.257	0.265	0.063
A-site	0.642	0.598	0.413	0.326	0.128	0.741	0.858	0.947	1.021	1.055	0.196
Σ cations	15.64	15.60	15.41	15.33	15.13	15.74	15.86	15.95	16.02	16.06	15.20
Al (total)	1.10	1.36	1.23	1.33	0.38	0.64	0.67	0.26	0.24	0.44	0.18
Fe#	0.79	0.76	0.57	0.58	0.42	0.86	0.79	0.93	0.87	0.89	0.98
R	0.11	0.07	0.19	0.25	0.17			0.06	0.07	0.04	0.28

Table 2. 4 Representative electron microprobe analyses of amphiboles from the Wentworth granites.

Notes: fe-ed: ferro-edenite, mg-hbl: magnesiohornblende, act: actinolite, fe-rcht: ferro-richterite, arf: arfvedsonite, fe-eck: ferro-eckermannite, rbk: riebeckite. Fe#: Fe_T/(Fe_T+Mg). Mg#: Mg/(Mg+Fe_T). R: Fe³⁺/ (Fe³⁺+Fe²⁺). Structural formulae calculated on the basis of 23 oxygens and normalized using the 13eCNK method as in Leake et al. (1997).

+3.6. Rocks with low \mathcal{E}_{Nd} (+1.1 to +2.5) have only calcic amphiboles, whereas samples with sodic-calcic and sodic amphiboles have higher values of \mathcal{E}_{Nd} (Fig. 2.5). This suggests the presence of two different sources for the granitic magma. A calcic magma was derived from a source with a more crustal isotopic signature that crystallized only calcic amphiboles, whereas a relatively more alkaline magma was derived from a source with a larger mantle component, from which sodic amphiboles crystallized.



Figure 2. 5 ϵ_{Nd} versus CaO/(Na₂O+K₂O) from whole-rock composition, showing two distinct groups of granites (separated by dotted lines). Samples with sodic amphiboles show ϵ_{Nd} values higher than +3, whereas all samples with calcic amphiboles have less than +2.5. Isotopic data from Pe-Piper and Piper (1998) and new data from the same laboratory presented in this work.

There is a difference in alkalinity (Agpaitic index, Table 2.5) that distinguishes the samples with sodic amphiboles from the granites with calcic amphiboles. The calculated values for the granites with sodic and calcic amphiboles of the Wentworth pluton

Sample	4636	7814	6490	5056	4641	6419	6518	7658	7710
Type of amp.	sodic	calcic	calcic	calcic	calcic	sodic	sodic	calcic	calcic, sodic- calcic
Relative age	Early	Early	Syn- gabbro	Syn- gabbro	Post- gabbro	Post- gabbro	Post- gabbro	Post- gabbro	Post- gabbro
Major Ox	ides (wt.	%) (Rec	alculated of	on a volatil	e-free basi	s)	0	0	0
SiO ₂	75.69	76.99	76.22	64.88	71.62	77.16	73.41	72.98	76.03
TiO ₂	0.17	0.19	0.17	0.87	0.46	0.18	0.43	0.32	0.23
Al ₂ O ₃	11.51	11.85	12.59	15.91	14.57	11.18	11.79	13.67	11.94
Fe ₂ O ₃ t	2.51	1.88	1.67	5.41	3.56	2.83	4.56	2.73	2.55
MgO	0.82	0.03	0.04	1.06	1.07	0.02	0.02	0.24	0.09
MnO	0.04	0.03	0.03	0.07	0.07	0.05	0.1	0.08	0.04
CaO	0.21	0.53	0.45	2.64	0.94	0.18	0.56	0.9	0.4
Na ₂ O	4.42	3.38	3.5	3.46	2.12	3.77	4.27	3.74	3.92
K ₂ O	4.62	5.09	5.31	5.51	5.48	4.61	4.84	5.27	4.78
P2O5	0.01	0.02	0.02	0.2	0.1	0.01	0.02	0.07	0.02
Total	100.00	99.99	100.0	100.01	99.99	99.99	100.00	100.00	100.00
Trace eler	nents (pj	om)							
Ba	39	b.d	89	669	599	32	12	276	125.3
Rb	224	174	163	176	212	243	170	241	212.6
Sr	10	33	18	148	91	11	12	63	17.7
Y	90	88	65	93	53	84	102	76	93.3
Zr	552	262	232	579	324	635	2136	344	786.5
Nb	69	60	62	50	20	101	77	45	55.7
Pb	3	16	18	30	41	33	14	32	10.3
Ga	30	27	28	23	19	37	34	24	29.2
Zn	131	40	44	68	72	163	138	58	138
Cu	n.d	6	12	10	n.d	13	22	13	18
NI V	4	4	b.d	6	3 15	b.d	b.d	b.d	b.d
V Cr	2	0.0 12	0 hd	49	13	b.d	0.u 7	/ hd	10 h d
	52 71.2	12 n d	n d	nd	nd	n d	232	n d	n d
La Ce	152	122	n d	n d	n d	n d	437	n d	n d
Nd	68	n.d	n.d	n.d	n.d	n.d	176	n.d	n.d
Sm	15.3	n.d	n.d	n.d	n.d	n.d	30.1	n.d	n.d
Eu	0.59	n.d	n.d	n.d	n.d	n.d	1.44	n.d	n.d
Tb	2.3	n.d	n.d	n.d	n.d	n.d	3.3	n.d	n.d
Yb	10.7	n.d	n.d	n.d	n.d	n.d	13.4	n.d	n.d
Lu	1.67	n.d	n.d	n.d	n.d	n.d	2.1	n.d	n.d
Th	16	17	18	19	17	24	21	36	9.9
U	5.9	7	n.d	n.d	n.d	n.d	3.5	n.d	1.3
Agpaitic Index	1.07	0.93	0.91	0.73	0.65	1.00	1.04	0.87	0.97
Fe#	0.73	0.98	0.97	0.82	0.75	0.99	1.00	0.91	0.96

Table 2. 5 Whole-rock analyses of the studied samples from the Wentworth pluton.

Notes: Agpaitic Index= molar (Na₂O+K₂O)/Al₂O₃, Fe# number= FeO_T/ (FeO_T+MgO) in wt% oxides (Frost et al., 2001), n.d.= not determined, b.d.= below detection limit. Whole-rock chemical data previously reported by Pe-Piper (1998).
correspond to those for the alkaline and aluminous petrographic associations, respectively, described by Gualda and Vlach (2007) for the Graciosa granites. Mineralogically, however, there are important differences. The hypersolvus granites are restricted to the granites with sodic amphibole, which in the case of the Wentworth pluton is predominantly primary arfvedsonite, in contrast to dominant riebeckite and minor arfvedsonite found in the Graciosa alkaline association. Furthermore, the wide compositional range in amphiboles found in the alkaline association of the Graciosa syenites and granites implies a link between the syenites and the granites by fractionation. Such evidence is lacking in the granites of the Wentworth pluton, as well as chemical zoning in the studied amphiboles (Appendix A.4). There is no evidence, therefore for a genetic link between the granites with sodic and the granites with calcic amphiboles.

2.5.2 INFLUENCE OF TEMPERATURE AND PRESSURE

There is a positive correlation between estimated zircon-saturation temperature and \mathcal{E}_{Nd} values of the studied samples (Fig. 2.6). Samples with calcic amphiboles have low \mathcal{E}_{Nd} values and low temperatures. As temperature and \mathcal{E}_{Nd} increases the samples pass from calcic amphibole-bearing to more sodic-calcic compositions and finally to sodic compositions in higher temperatures (Fig. 2.6). The calcic amphiboles crystallized at similar temperatures with the zircons, as indicated by the hornblende thermometer, and therefore the two minerals probably crystallized broadly at the same time. The higher T_{Zr} values of the granites with sodic amphiboles do not necessarily mean that the sodic amphiboles formed at higher temperatures. Rather, the sodic amphiboles crystallized from a magma with a higher liquidus temperature than the one that formed the calcic amphiboles.



Figure 2. 6 ε_{Nd} versus temperature diagram, demonstrating the positive correlation of temperature and isotopic signature to the type of amphibole present in the granites of the Wentworth pluton. Temperatures estimated from zircon-saturation thermometer.

2.5.3 The impact of oxygen fugacity on the calcic amphiboles of the Wentworth Pluton

Oxygen fugacity is believed to have an impact on the chemistry of calcic amphiboles. With increasing oxygen fugacity the amphiboles crystallizing from a magma become progressively enriched in Mg (Wones, 1981) as the result of precipitation of magnetite. Amphiboles thus with high Fe# (Fe_T/(Fe_T+Mg) in a.p.f.u.) are believed to have formed under low oxygen fugacity (Anderson and Smith, 1995). The majority of the analyzed amphiboles in the Wentworth granites have moderate to high Fe# (0.5-0.8) (Fig. 2.7).

However, this inference conflicts with the presence in these samples of magnetite as the dominant Fe-oxide, which suggests oxidizing conditions (Table 2.3), like those for the magnetite series of granites. Anderson and Smith (1995) interpreted an oxygen fugacity below QFM for the Enchanted Rock batholith, on the basis of high Fe# in the magmatic amphiboles, even though the Enchanted Rock batholith belongs to the magnetite series. The granites of the Wentworth pluton, however, contain Fe-rich amphiboles and yet the oxygen fugacity indicated by the assemblage of magnetite, titanite and quartz is probably above the QFM, suggesting that for some reason, oxygen fugacity had a limited effect on the Fe# of the amphiboles.



Figure 2. 7 Fe-number (Fe#: $(Fe^{2+} + Fe^{3+})/(Fe^{2+} + Fe^{3+}+Mg)$) versus Al of the hornblendes from the Wentworth granites (fields after Anderson and Smith 1995), suggesting their formation under moderate to low oxygen fugacity. The range of the amphiboles from the Enchanted Rock batholith formed under low oxygen fugacity (as described in Anderson and Smith, 1995) is plotted for comparison.

The molar ratio R (Fe³⁺/ (Fe³⁺+Fe²⁺) of the calcic amphiboles from the Wentworth pluton ranges from 0.07 to 0.25 (Table 2.4). This ratio can be affected by the oxygen fugacity of the parent magma and such values indicate oxygen fugacity above the QFM buffer (Clowe et al., 1988), suggesting crystallization under moderately oxidizing conditions.

With the exception of Mg-rich ferro-hornblende, present in the hybrid sample, all primary calcic amphiboles in the Wentworth pluton show almost constant values of molar Mg (0.9–1.1 a.p.f.u., Table 2.4) suggesting that these minerals did not become as enriched in Mg as amphiboles that form under moderate to high oxygen fugacity do. The data from the calcic amphiboles of the Wentworth pluton, therefore, are ambiguous. The R ratio indicates oxidizing conditions above the QFM buffer, whereas their relatively low Mg amounts suggest crystallization under low oxygen fugacity and consequently affected the calculation of the Fe# values.

The second ferromagnesian mineral in the granites is biotite (Table 2.6, Appendix A.5) and is almost exclusively found in the granites with calcic amphiboles. The biotites present in these granites, even though close to the annite end-member, still contain significant amounts of Mg and therefore the timing of their crystallization needs to be evaluated. Interlocking textures of biotite with hornblende in the granitic samples (Fig. 2.8) indicate that the crystallization of these minerals took place almost simultaneously. As such, the crystallizing biotites could also incorporate significant amounts of Mg from the crystallizing melt, as indicated by their mineral chemistry (Table 2.6). Kretz (1959) demonstrated that the distribution of Mg between coexisting biotite and calcic amphibole can be approached graphically by plotting the XMg ratio (Mg/(Mg+Fe+Mn+Ti) in a.p.f.u.) of both minerals in a Roozeboom diagram (Fig 2.9). A regular distribution of Mg should produce a linear trend with positive slope if the two minerals represent ideal binary mixtures (i.e. under constant temperature or pressure) (Saxena, 1968). The plotted ratios for biotite and coexisting calcic amphiboles from three studied samples do form broadly a linear trend as a whole (Fig 2.9). However, in each sample the plotted data

form negative linear trends that deviate from a regular distribution. These trends show that an increase of Mg in biotite was accompanied by a decrease of Mg in amphibole. Such deviations may be the result of a change in the pressure and temperature, or other chemical changes in the coexisting minerals (Kretz, 1959). Thus a possible explanation for this deviation could be a rapid ascent and emplacement of the granitic magma that could significantly change the pressure, however this cannot be evaluated at this point. Nonetheless, the textural and chemical evidence from the amphiboles and the biotites suggests that during the crystallization of the calcic amphiboles of the Wentworth pluton



Figure 2. 8 Coexisting anhedral calcic amphibole with biotite in the granites of the Wentworth pluton. The two grains are in contact and present an interlocking texture.

there was also simultaneous crystallization of biotite, in which Mg was incorporated as

well. The partitioning of magnesium into the biotite, under high oxygen fugacity,

prevented the Mg-enrichment of the amphiboles, leading to the formation of Fe-rich calcic amphiboles, even under oxidizing conditions.



Figure 2. 9 Roozeboom diagram showing the XMg ratio (octahedral Mg/Mg+Mn+Fe_T+Ti in a.p.f.u) of coexisting biotite and calcic amphibole in the granites of the Wentworth pluton. With the exception of the hybrid sample which contains magnesiohornblende, the samples plot close to the 1:1 curve of the diagram that indicates a regular distribution of Mg between the two minerals. Within each mineral there is a negative correlation between the two ratios.

2.5.4 PRESENCE OF VOLATILES

The fact that the calcic amphiboles are related to a magma with mean 6 wt. % H₂O (Table 2.3) can explain why these samples contain biotite and also why they have lower zirconsaturation temperatures than those with sodic amphiboles. According to experimental data of Baker et al. (2002), zircon solubility is significantly lower in melts with 2 wt. % H_2O at high temperatures, compared to more hydrous melts. Therefore, precipitation of zircon was early in the samples with sodic amphiboles, at temperatures higher than 850 °C, whereas it came later in the melt with calcic amphiboles due to the presence of high amounts of H_2O . The presence of volatiles can affect the stability and composition of amphiboles (Martin, 2007). The H_2O content of the melt can control the mineral chemistry of crystallizing amphiboles; a decrease in H_2O concentrations in the melt can lead to more sodic compositions (Scaillet and Macdonald, 2003; Ridolfi et al., 2010).

Scaillet and Macdonald (2003) demonstrated that a decrease in H_2O of the melt is associated with an increase of F in the amphiboles. The granites with sodic amphiboles show systematically higher initial amounts of F in the melt than the granites with calcic amphiboles, suggesting a negative correlation between these two volatile phases (Fig. 2.10).

The variations in initial H_2O contents between the two granitic magmas may have affected the F enrichment of the amphiboles, besides the availability of F itself. The more hydrous calcic magma produced additional biotite, thus leading F to participate in that mineral as well (Table 2.6) and resulting in smaller F contents in the crystallizing amphiboles. The variations in F contents between sodic and calcic amphiboles reflect the control of different initial magmatic contents of both H_2O and F on the mineral chemistry of the produced amphibole.

2.5.5 IMPLICATIONS FOR THE MAGMATIC EVOLUTION OF THE WENTWORTH GRANITES

The systematic differences in H_2O and F contents of the melt, temperature, \mathcal{E}_{Nd} values, Ba and alkalinity, suggest that the samples with calcic amphiboles and sodic

amphiboles represent two different, coeval, granitic magmas: a relatively sodic and a

relatively calcic melt.

Table 2. 6 Representative microprobe analyses of biotite from the granites of the Wentworth pluton.

%) 7.34 4.12 3.27 3.81 0.19 8.93 0.01 0.14 9.58 0.00 0.43 0.35 1.59	37.04 3.78 13.37 23.95 0.18 8.79 0.03 0.14 9.54 0.00 0.41 0.40	38.45 2.51 14.68 24.76 0.52 6.28 0.03 0.09 9.38 0.00	38.27 2.66 13.60 24.83 0.40 7.41 0.02 0.06 9.35	38.53 2.41 14.26 24.73 0.51 6.69 0.05 0.04 9.53	36.15 3.72 12.79 30.09 0.26 4.19 0.02 0.04 0.04	35.22 4.02 12.49 30.74 0.35 4.08 0.02 0.06		
7.34 4.12 3.27 3.81 0.19 8.93 0.01 0.14 9.58 0.00 0.43 0.35 1.59	37.04 3.78 13.37 23.95 0.18 8.79 0.03 0.14 9.54 0.00 0.41 0.40	38.45 2.51 14.68 24.76 0.52 6.28 0.03 0.09 9.38 0.00	38.27 2.66 13.60 24.83 0.40 7.41 0.02 0.06 9.35	38.53 2.41 14.26 24.73 0.51 6.69 0.05 0.04 9.53	36.15 3.72 12.79 30.09 0.26 4.19 0.02 0.04	35.22 4.02 12.49 30.74 0.35 4.08 0.02 0.06		
4.12 3.27 3.81 0.19 8.93 0.01 0.14 9.58 0.00 0.43 0.35 1.59	3.78 13.37 23.95 0.18 8.79 0.03 0.14 9.54 0.00 0.41 0.40	2.51 14.68 24.76 0.52 6.28 0.03 0.09 9.38 0.00	2.66 13.60 24.83 0.40 7.41 0.02 0.06 9.35	2.41 14.26 24.73 0.51 6.69 0.05 0.04 9.53	3.72 12.79 30.09 0.26 4.19 0.02 0.04	4.02 12.49 30.74 0.35 4.08 0.02 0.06		
3.27 3.81 0.19 8.93 0.01 0.14 9.58 0.00 0.43 0.35 1.59	13.37 23.95 0.18 8.79 0.03 0.14 9.54 0.00 0.41 0.40	14.68 24.76 0.52 6.28 0.03 0.09 9.38 0.00	13.60 24.83 0.40 7.41 0.02 0.06 9.35	14.26 24.73 0.51 6.69 0.05 0.04 9.53	12.79 30.09 0.26 4.19 0.02 0.04	12.49 30.74 0.35 4.08 0.02 0.06		
3.81 0.19 8.93 0.01 0.14 9.58 0.00 0.43 0.35 1.59	23.95 0.18 8.79 0.03 0.14 9.54 0.00 0.41 0.40	24.76 0.52 6.28 0.03 0.09 9.38 0.00	24.83 0.40 7.41 0.02 0.06 9.35	24.73 0.51 6.69 0.05 0.04 9.53	30.09 0.26 4.19 0.02 0.04	30.74 0.35 4.08 0.02 0.06		
0.19 8.93 0.01 0.14 9.58 0.00 0.43 0.35 1.59	0.18 8.79 0.03 0.14 9.54 0.00 0.41 0.40	0.52 6.28 0.03 0.09 9.38 0.00	0.40 7.41 0.02 0.06 9.35	0.51 6.69 0.05 0.04 9.53	0.26 4.19 0.02 0.04	0.35 4.08 0.02 0.06		
 8.93 0.01 0.14 9.58 0.00 0.43 0.35 1.59 	8.79 0.03 0.14 9.54 0.00 0.41	6.28 0.03 0.09 9.38 0.00	7.41 0.02 0.06 9.35	6.69 0.05 0.04 9.53	4.19 0.02 0.04	4.08 0.02 0.06		
0.01 0.14 9.58 0.00 0.43 0.35 1.59	0.03 0.14 9.54 0.00 0.41	0.03 0.09 9.38 0.00	0.02 0.06 9.35	0.05 0.04 9.53	0.02 0.04 9.01	0.02 0.06		
0.14 9.58 0.00 0.43 0.35 1.59	0.14 9.54 0.00 0.41	0.09 9.38 0.00	0.06 9.35	0.04 9.53	0.04	0.06		
9.58 0.00 0.43 0.35 1.59	9.54 0.00 0.41	9.38 0.00	9.35	9.53	0.01			
0.00 0.43 0.35 1.59	0.00 0.41	0.00	0.02		9.01	9.15		
0.43 0.35 1.59	0.41	0 10	0.02	0.02	0.04	0.02		
0.35 1.59	0.40	0.42	0.63	0.41	0.19	0.22		
1.59	0.40	0.14	0.10	0.18	0.27	0.33		
	1.57	1.64	1.54	1.63	1.64	1.59		
0.26	0.26	0.21	0.29	0.21	0.14	0.16		
9.75	99.19	98.89	98.89	98.98	98.39	98.28		
Structural formulae based on 22 oxygens								
.964	5.959	6.173	6.165	6.189	6.027	5.930		
.036	2.041	1.827	1.835	1.811	1.973	2.070		
.461	0.493	0.949	0.744	0.887	0.537	0.407		
.495	0.458	0.304	0.322	0.291	0.466	0.509		
.180	3.222	3.325	3.345	3.322	4.194	4.329		
.026	0.025	0.071	0.055	0.069	0.036	0.050		
.126	2.109	1.503	1.779	1.602	1.042	1.025		
.002	0.005	0.005	0.004	0.008	0.003	0.003		
.042	0.044	0.026	0.017	0.013	0.011	0.019		
.953	1.958	1.921	1.920	1.953	1.916	1.965		
16.3	16.3	16.1	16.2	16.1	16.2	16.3		
0.43	0.42	0.42	0.64	0.41	0.2	0.23		
0.19	0.22	0.07	0.06	0.1	0.15	0.19		
1.69	1.68	1.75	1.65	1.74	1.83	1.79		
2.95	2.95	2.04	2.45	2.25	0.94	0.84		
0.60	0.60	0.69	0.65	0.67	0.80	0.81		
0.19	0.18	0.19	0.27	0.18	0.09	0.10		
	.036 .461 .495 .180 .026 .126 .002 .042 .953 16.3 0.43 0.19 <u>1.69</u> <u>2.95</u> 0.60 0.19 ett/(Fe	.036 2.041 .461 0.493 .495 0.458 .180 3.222 .026 0.025 .126 2.109 .002 0.005 .042 0.044 .953 1.958 16.3 16.3 0.43 0.42 0.19 0.22 1.69 1.68 2.95 2.95 0.60 0.60 0.19 0.18 eT/(Fet+Mg) et	.036 2.041 1.827 .461 0.493 0.949 .495 0.458 0.304 .180 3.222 3.325 .026 0.025 0.071 .126 2.109 1.503 .002 0.005 0.005 .042 0.044 0.026 .953 1.958 1.921 16.316.316.1 0.43 0.42 0.42 0.19 0.22 0.07 1.69 1.68 1.75 2.95 2.95 2.04 0.60 0.60 0.69 0.19 0.18 0.19 $e_T/(Fe_T+Mg)$ elements	.0362.0411.8271.835.461 0.493 0.949 0.744 .495 0.458 0.304 0.322 .180 3.222 3.325 3.345 .026 0.025 0.071 0.055 .126 2.109 1.503 1.779 .002 0.005 0.005 0.004 .042 0.044 0.026 0.017 .953 1.958 1.921 1.920 16.3 16.3 16.1 16.2 0.43 0.42 0.42 0.64 0.19 0.22 0.07 0.06 1.69 1.68 1.75 1.65 2.95 2.95 2.04 2.45 0.60 0.60 0.69 0.65 0.19 0.18 0.19 0.27 $e_T/(Fe_T+Mg)$ elements in a.p.f.	.0362.0411.8271.8351.811.4610.4930.9490.7440.887.4950.4580.3040.3220.291.1803.2223.3253.3453.322.0260.0250.0710.0550.069.1262.1091.5031.7791.602.0020.0050.0050.0040.008.0420.0440.0260.0170.013.9531.9581.9211.9201.95316.316.116.216.10.430.420.420.640.410.190.220.070.060.11.691.681.751.651.742.952.952.042.452.250.600.600.690.650.670.190.180.190.270.18er/(Fer+Mg) elements in a p. f.u.1 indi	.0362.0411.8271.8351.8111.973.4610.4930.9490.7440.8870.537.4950.4580.3040.3220.2910.466.1803.2223.3253.3453.3224.194.0260.0250.0710.0550.0690.036.1262.1091.5031.7791.6021.042.0020.0050.0050.0040.0080.003.0420.4440.0260.0170.0130.011.9531.9581.9211.9201.9531.91616.316.316.116.216.116.20.430.420.420.640.410.20.190.220.070.060.10.151.691.681.751.651.741.832.952.952.042.452.250.940.600.600.690.650.670.800.190.180.190.270.180.09		



Figure 2. 10 Initial fluorine-in-melt contents versus magmatic water contents for the granites of the Wentworth pluton.

The sodic magma shows lower values of estimated viscosity compared to the calcic magma. The presence of H₂O can reduce the viscosity of the melt (Goto et al., 2005) and enhance the magma ascent. However in mid- to upper crustal levels the influence of water may be limited due to the release of volatiles at low pressures and decrease in temperature. The high magmatic temperatures and F can also reduce the viscosity of the melts and drive the emplacement of the melt in the upper crust (Dingwell et al., 1985). Therefore, while the mobility of the calcic melt may have been influenced by the presence of water, in the sodic melts the shallow emplacement was the result of reduced viscosity due to higher temperatures and probably F.

The granites of the Wentworth pluton show a systematic increase of emplacement pressures with decreasing age (Fig. 2.11). The differences in the pressure estimates for granites of different ages could be either the result of differences in the level of emplacement or due to changes in overlying lithostatic load with time. During the

emplacement of the Wentworth early granites (from 368 ± 4 to 362 ± 2 Ma, Table 2.7) the volcanic pile included a part of the Byers Brook rhyolites, less than 3 km thick, (<1 kbar assuming a density of a felsic continental crust) (Fig. 2.11). The syn-gabbro granites $(356\pm3$ Ma, Table 2.7) are probably synchronous with either the uppermost Byers Brook Formation, or the lower Diamond Brook basalts, and also with the intrusion of the gabbro (Fig. 2.12). Therefore, at the time of the emplacement of the syn-gabbro granites, the volcanic pile included 3 km of the Byers Brook Formation and at least some of the Diamond Brook basalts, thus equivalent to 1.0-1.5 kbar lithostatic load. The post-gabbro granites or all of the present volcanic pile (~4.5 km, ~ 1.5 kbar), suggesting that the lithostatic load was the highest during the emplacement of these rocks.

Taking into account the age uncertainties and the mean pressure estimates for the studied samples, it seems that the observed increase of pressure over time broadly correlates with the likely increase of thickness of the volcanic pile (including the thick succession of basalts) and the emplacement of the voluminous and denser gabbro. Thus the absolute structural level of the syn-gabbro and post-gabbro granites probably did not vary significantly. The systematically higher magmatic temperatures (T_{Zr}) in the post-gabbro granites compared to the early and the syn-gabbro granites (Table 2.3) reflects the relatively greater depth of emplacement created by the overlying volcanic pile. The very low pressure of emplacement of the late dyke that intruded the gabbro (sample 7710) may be related to a decreased lithostatic load due to erosion, after the volcanic succession was completed. Rapid erosion and unroofing of the plutons are recorded throughout the

Cobequid Highlands, where A-type granitic clasts are found in the sediments of the Horton Group which is in turn intruded by younger granitic phases (Pe-Piper et al.,



Figure 2. 11 Distribution of estimated pressures of emplacement for the Wentworth granites plotted against age (error bar for pressure estimates equal to 0.6 kbar, Anderson and Smith 1995, average error of age estimates 3 Ma). The approximate lithostatic load of the volcanic pile over time is also plotted for comparison.

1998). The fine grained granitic dyke contains sodic-calcic amphiboles interstitial between larger grains of quartz and feldspar (Table 2.3). Even though it was emplaced at very low pressures, the magmatic temperatures were high, as indicated by the zirconsaturation thermometer (Table 2.3). There is no correlation between the estimated viscosities and the observed increase of pressure with time, which could indicate a dominant influence of the ascent process on the latter.



Figure 2. 12 Geochronological data for the various geological units in the area of the Wentworth pluton, along with the stratigraphic column of the volcanic rocks, indicating the positions of dated samples. Data compiled from Utting et al. (1989), Martel et al. (1993), Doig et al. (1996), Dunning et al. (2002) and Pe-Piper et al. (2004). All Ar-Ar ages have been corrected from the original papers according to Kuiper et al. (2008).

Furthermore, the presence of the interstitial amphiboles in the fine-grained granite dyke, indicates that these minerals were the last to crystallize from a residual melt, in the space between the larger pre-existing quartz and feldspar. The sodic-calcic composition of the interstitial amphiboles in this sample suggests that the residual melt had an intermediate composition between the identified calcic and sodic granitic magmas. This originally calcic magmatic pulse, as indicated by the presence of primary ferro-edenite, may have been partially hybridized during ascent by the sodic melts that had been

Geological Unit	Age (Ma)	Method	Reference
Early granites of the Wentworth pluton	362±2	U-Pb (zrn)	Doig et al. (1996)
	368±4	Ar-Ar (arf)	Pe-Piper et al. (2004)
Late granites (probable syn-gabbro granites)	356±3	Ar-Ar (arf)	Pe-Piper et al. (2004)
Wentworth gabbro	357±4	Ar-Ar (hbl)	Pe-Piper et al. (2004)
	353±2	Ar-Ar (bt)	Pe-Piper et al. (2004)
Byers Brook Formation (Upper)	358±1	U-Pb (zrn)	Dunning et al. (2002)
Diamond Brook Formation (Upper)	348±3	palynology	Utting et al. (1989), Menning et al. (2006)
Diamond Brook Formation (Middle)	353±3	U-Pb (zrn)	Dunning et al. (2002)
Diamond Brook Formation (Middle-Lower)	359±2	palynology	Martel et al. (1993)

Table 2. 7 Chronological data for the geological units in the area of the Wentworth pluton.

Notes: All Ar-Ar ages have been corrected from the original ages reported in the cited papers, according to Kuiper et al. (2008). Biostratigraphic ages are corrected according to ICS 2013. Reported age uncertainties are given in 2σ .

previously emplaced at lower crustal levels, so that the residual melt that finally reached the upper crust has chemical affinities of both granitic melts and thus produced amphiboles of sodic-calcic composition. Granites with sodic-calcic amphiboles are only known from one dyke and as enclaves in outcrops of granites that contain sodic amphiboles. Such a mode of occurrence implies that these granites do not represent a major magmatic phase in Wentworth pluton; rather they are the result of mingling of coexisting sodic and calcic granitic melts.

2.6 CONCLUSIONS

The types of primary amphiboles in the Wentworth pluton are represented principally by calcic and sodic compositions with rare sodic-calcic varieties. The pressure estimates suggest that the pluton was emplaced between 1 to 3 kbar, which corresponds to upper crust. Differences in temperature, E_{Nd} , Ba, H₂O and F in-melt contents suggest the presence of two coeval granitic magmatic systems: one that formed the calcic amphibolebearing granites and another that formed the sodic amphibole-bearing granites. The calcic magma was H₂O -rich producing granites in a range of pressure and temperature and with textures that vary from medium to fine grained granular and granophyric. The sodic magma derived from a source with larger mantle component, was drier, and less viscous due to elevated magmatic temperatures. Coexistence of both magmas is indicated by a late fine-grained granite dyke, with interstitial amphiboles of calcic and sodic-calcic composition that show the lowest estimated pressure. These observations suggest that a younger batch of the calcic melt was mixed with the sodic granitic magma and reached the upper crust before complete crystallization, producing sodic-calcic amphiboles upon emplacement.

The Wentworth pluton is composite, constructed by two distinct granitic magmas and a gabbroic magma. The present work shows that extreme variations in the type of magmatic amphiboles present in an A-type granitic pluton may reflect the complexity of the parental magmatic system. Other A-type granites with such a mineralogical diversity and limited evidence of fractionation may have similarly complex origins.

CHAPTER 3 SYSTEMATIC MINERALOGICAL DIVERSITY IN A-TYPE GRANITIC INTRUSIONS: CONTROL OF MAGMATIC SOURCE AND GEOLOGICAL PROCESSES

PREFACE

This chapter has been submitted as: "Papoutsa, A., Pe-Piper, G., and Piper, D.J.W., Systematic mineralogical diversity in A-type granitic intrusions: control of magmatic source and geological processes", and is currently in press in Geological Society of America Bulletin.

3.1 Abstract

The origin and causes of mineralogical diversity of A-type granites are debated. A series of A-type granite plutons, with distinct mineralogical differences, emplaced along an late Paleozoic crustal-scale shear zone in the Cobequid Highlands, Nova Scotia, provide an opportunity to examine the origin of different A-type plutons in a similar tectonic setting. Based on the ferromagnesian minerals present, the plutons are classified into sodic granites with sodic amphibole, calcic granites with calcic amphibole, and biotite granites. Sodic and calcic granites occur exclusively in complex intrusions with subequal amounts of gabbro in the eastern part of the Cobequid Highlands, whereas plutons in the western part, with lesser gabbro, comprise only biotite granites. Trace elements and radiogenic isotopes show that the three granite types have different sources. Intensive parameters including temperature, pressure, and water-in-melt contents were estimated from mineralogical and geochemical data. Modeling of these geochemical data suggests that the biotite and calcic granites were derived by 20-40% partial melting of intracrustal feldspathic rocks, whereas the sodic granites are extreme fractionates (90%) of coeval mafic magma. We propose that

supply of late Paleozoic mafic magma, probably related to regional extension and underplating beneath the Magdalen Basin, reached the Cobequid Highlands and created a deep crustal hot zone. Extreme fractionation of these underplated mafic melts produced the sodic granites. Heat transfer from crystallizing mafic magma induced partial melting of the surrounding crust, creating batches of biotite and calcic granitic melts in different depths. Fractionated and crustally-derived melts segregated along crustal-scale faults, as indicated from field evidence, constructing the complex plutons in the east. Melting of the crust was further facilitated by the release of water from the crustal rocks upon heating. In the eastern part of the Cobequid Highlands, water was released predominantly by magmatic rocks and in lesser amounts compared to the west where Neoproterozoic sedimentary and volcaniclastic rocks are more abundant. The volatile-rich granitic melts in the western part of the study area crystallized rapidly, stabilizing only biotite. This study demonstrates that the mineralogical variations in A-type granites arise from rather similar magma compositions, but are important petrogenetic indicators of varying sources, specific magmatic processes and emplacement conditions.

3.2 INTRODUCTION

A-type granites comprise a distinct group of granitoid rocks, the origin of which is well-debated. This group of rocks was recognized by Chappell and White (1974) in their proposed genetic "alphabet" classification scheme, in which these granites were separated from the I-, S- and M-types on the basis of their distinct alkaline trace element geochemistry and anorogenic tectonic setting. The authors proposed that these granites originated from a crustal residue that remained after a previous event of partial melting. Loiselle and Wones (1979) introduced the term "A-type" and proposed that differentiation of alkali basalt may also produce a suitable parent magma. However A-type granites, identified solely from their trace elements, also have been reported from post-orogenic tectonic settings, and include rocks of variable alkalinity ranging from peralkaline to even peraluminous compositions (Bonin, 2007).

The term "A-type granite" (Loiselle and Wones, 1979) initially referred to granites with mildly alkaline geochemistry emplaced along continental rift zones and crystallized under low water fugacities. Such granites are richer in Fe, K, rare metals and rare earth elements (REE) compared to other granitoids (Whalen et al., 1987), implying a distinctive petrogenesis. As such, the importance of A-type granites is also economic since they may be associated with polymetallic deposits of Sn, Nb, Ta, U, Th, F, and REE (Bettencourt et al., 2005; Costi et al., 2009; Dall'Agnol et al., 2012). Current concepts of A-type granite petrogenesis include partial melting of tonalitic to granodioritic crust (Creaser et al., 1991; Frost and Frost, 2010), high-temperature dehydration melting of calc-alkaline granitoids in the shallow crust (Patiño Douce, 1997), or differentiation of mafic magma (Loiselle and Wones, 1979; Frost and Frost, 2010). Variation in primary ferromagnesian minerals in Atype granites suggests contrasting sources and magmatic conditions, even within a single alkaline province (Gualda and Vlach, 2007; Shellnutt and Zhou, 2007; Shellnutt and Iizuka, 2011). In this paper we show that systematic mineralogical differences in coeval Atype granites within the same magmatic province are the result of varying sources, specific magmatic processes, and styles of magma emplacement.

The Cobequid Highlands in Nova Scotia provide a suitable opportunity for such a study. Several coeval A-type granite plutons were emplaced along the active faults of a shear zone active in the late Devonian. These granitic plutons were emplaced under the

same regional tectonic regime and have similar overall geochemistry, but they present important mineralogical differences. The plutons include biotite-bearing and both sodicand calcic-amphibole-bearing A-type granites, with only minor chemical zoning in the ferromagnesian minerals. The mineralogical diversity may thus reflect deeper magmatic processes and variations in magmatic sources.

3.3 GEOLOGICAL SETTING

The Cobequid Highlands are located in the Avalon terrane of the Canadian Appalachians, just north of the Cobequid-Chedabucto fault zone, which separates this



Figure 3. 1 Simplified geological map of Nova Scotia (A) showing the position of the study area and (B) geological map of the Cobequid Highlands (modified from Pe-Piper and Piper, 2003).

terrane from the Meguma terrane (Fig. 3.1A). The Cobequid Highlands are underlain by

Neoproterozoic subduction-related igneous and metasedimentary rocks, minor Ordovician

igneous rocks, Silurian to lower Devonian sedimentary rocks and late Paleozoic plutons

and their extrusive equivalents (Pe-Piper and Piper, 2003). Within the Cobequid

Highlands, the Cobequid Shear Zone consists of a series of strike-slip faults that were initially active in the Neoproterozoic and reactived during latest Devonian to earliest Carboniferous. Plutons were emplaced along the active faults of the deforming Cobequid Shear Zone and consist of granite and gabbro bodies. The Rockland Brook fault, the major fault of the Cobequid Shear Zone in the eastern part of the Highlands, separates the Neoproterozoic Jeffers Block to the north from the Bass River Block to the south (Fig. 3.1B). The Cobequid fault marks the southern margin of the Cobequid Highlands (Pe-Piper and Piper, 2003).

Lower crust of the Avalon terrane has a distinctive seismic signature compared to both inboard and outboard Appalachian terranes (Marillier et al., 1989; Keen et al., 1991), and its seismic velocity of 6.8 km/s (Marillier et al., 1994) may indicate a significant mafic component, related to its long history of subduction in the Neoproterozoic.

In the late Devonian, subduction of the Rheic Ocean beneath Nova Scotia was associated with the emplacement of the voluminous South Mountain Batholith and satellite plutons in the Meguma terrane (Clarke et al., 1997). Towards the end of the Devonian, incipient closure of the Rheic Ocean at the Virginia and Hermitage promontories led to dextral strike-slip faulting parallel to the strike of the orogen (Hibbard and Waldron, 2009). This resulted in distributed extension of the basement beneath the Gulf of St. Laurence and the opening of the Magdalen Basin (Fig. 3.1A) as a step-over zone. Crustal thinning continued through at least the early Carboniferous, accompanied by underplating of a thick gabbroic layer beneath the Magdalen Basin (Marillier and Verhoef, 1989; Marillier and Reid, 1990). The Cobequid Shear Zone and its continuation along the west coast of Cape Breton Island marked the southwestern edge of the Magdalen Basin and

acted as a pathway for magma, which formed flood basalts >1 km thick in the Fisset Brook Formation of western Cape Breton Island and 1.5 km thick in the Fountain Lake Group of the eastern Cobequid Highlands (Pe-Piper and Piper, 2003).

All late Paleozoic granites in the Cobequid Highlands have A-type affinities, whereas the gabbros are ferro-gabbros. Synchronous with the plutons were the basalts and rhyolites of the Fountain Lake Group that show similar chemical affinities to their intrusive equivalents (Dessureau et al., 2000; Pe-Piper and Piper, 2003). The plutons selected for this study are the Wentworth (WP) and Pleasant Hills (PHP) plutons (complex intrusions with significant amounts of gabbro) in the eastern part, the North River pluton (NRP) in the central part (small intrusion with minor gabbro), and the Cape Chignecto pluton (CCP) in the western part of the Cobequid Highlands (homogeneous granite intrusion with moderate amounts of gabbro). All are coeval (365±4 Ma to 358±4 Ma), with the latest plutonic body being the large gabbro in the Wentworth pluton (357±4 Ma) (Dunning et al, 2002; Pe-Piper et al., 2004; Murphy et al., 2011).

3.4 METHODS

A total of 127 thin and polished thin sections from the four representative plutons were examined by petrographic microscope. Some samples used in this study were previously analyzed geochemically by X-ray fluorescence (XRF) for major elements and certain trace elements (Ba, Rb, Sr, Y, Zr, Nb, Pb, Ga, Zn, Cu, Ni, V, Cr and Co) and by instrumental neutron activation analysis (INAA) for the REEs and specific trace elements (Cr, Cs, Hf, Sb, Sc, Ta, Th, and U). The analytical precision is better than 2%, except for MgO, Na₂O and Nb for which is better than 5%, and Th which is better than 10% (Pe-Piper and Piper, 1989). Some of these samples, together with some new samples, were re-

analyzed for this study at Activation Laboratories. Major elements were analyzed by fusion ICP-MS with a detection limit of 0.01 wt.% for most oxides except MnO (0.001 wt.%). Trace elements including REE were also analyzed by fusion ICP-MS, except for Ni, Cu, Zn, and Pb for which total digestion ICP-MS was employed. Trace elements have detection limits below 5 ppm, except for Cr (20 ppm). Geochemical data were processed using MINPET 2.2 (Richard, 1995) software for Windows and the plotted data were recalculated to 100% on volatile-free basis.

Oxygen isotope analyses of selected samples were determined at the Laboratory of Stable Isotope Science of the University of Western Ontario, using the ClF₃ extraction method for silicates and oxides with a precision of 0.3 ‰. Oxygen isotope data are presented in the normal delta notation relative to the Vienna Standard Mean Ocean Water (V-SMOW). Neodymium-Sm and Pb isotope analyses were mostly reported by Pe-Piper and Piper (1998), where details of methods and precision are given, and some analyses were made by the same laboratory reported in Papoutsa and Pe-Piper (2014).

3.5 PETROGRAPHY OF THE LATE PALEOZOIC A-TYPE GRANITES

3.5.1 MINERALOGICAL VARIATIONS

The granites are fine- to coarse-grained alkali-feldspar granites with a variety of textures including granular, granophyric, and porphyritic. Most granites are subsolvus with two types of subhedral magmatic feldspar. Plagioclase ranges from oligoclase to calcic albite whereas the K-feldspar is orthoclase. Hypersolvus granites have been identified where the only feldspar present is a perthitic K-feldspar. The granites host variable amounts and types of magmatic ferromagnesian minerals such as biotite and amphibole. Biotite has an annitic composition whereas the amphibole can be calcic, sodic-calcic or

sodic. The primary calcic amphiboles include ferro-edenite and ferro-hornblende, sodiccalcic amphiboles are represented by ferro-richterite and the sodic amphiboles by arfvedsonite (as described in Pe-Piper, 2007; Papoutsa and Pe-Piper, 2014). Secondary minerals include epidote, chlorite, titanite, actinolite, ferro-winchite, ferro-eckermannite, riebeckite, and green biotite, derived mostly from the breakdown of the primary ferromagnesian phases. Magnetite is the dominant opaque mineral, but ilmenite and pyrite are also present in small amounts. Accessory primary minerals include apatite, zircon, titanite, allanite-(Ce), and chevkinite-(Ce).

Based on the type of magmatic ferromagnesian mineral present, five granitic types are recognized: a) sodic granite with arfvedsonite, b) calcic granite with ferro-edenite or ferro-hornblende, with or without biotite, c) sodic-calcic granite with sodic-calcic amphibole coexisting with either sodic or calcic amphibole, d) biotite granite with biotite but no amphibole, and e) leucogranite with ferromagnesian silicate minerals absent.

With the exception of sodic granites, which are exclusively hypersolvus and some leucogranites, all other granite types are subsolvus. The sodic-calcic granites are found as late microgranitic dykes and enclaves in either gabbro or sodic granite. The leucogranites occur exclusively as late dykes cutting all lithologies in the plutons. These two minor lithologies are not considered further in this study. The Cape Chignecto and North River plutons contain only biotite granite, the Pleasant Hills pluton contains calcic and biotite granite, whereas the Wentworth pluton is the largest and most complex intrusion and contains all three major granite types. The three main granite types show no systematic geographic distribution within the composite Pleasant Hills and Wentworth plutons.

3.5.2 WHOLE-ROCK CHEMISTRY

All granite types plot within the A-type field in the geochemical diagrams of Whalen et al. (1987) (Fig. 3.2A), and range from mildly peralkaline to metaluminous and peraluminous compositions (Fig. 3.2B).



Figure 3. 2 Geochemical classification diagrams for the Late Paleozoic granites from the Cobeqiod Highlands. (A) shows the classification of the studied granites as A-type (after Whalen et al., 1987). (B) shows the alkaline character of the sodic, calcic and biotite granites (as in Maniar and Piccoli, 1989).

The sodic granites are the most peralkaline, the calcic granites range from metaluminous to peraluminous, whereas the biotite granites are mainly peraluminous. The sodic granites are the only ones to plot in the A1 field (Fig. 3.3A) in the discrimination diagrams of Eby (1992), whereas the calcic and biotite granites are classified as A2 granites (Fig. 3.3B and C), although all three types plot close to the boundary between the two fields. The chondrite-normalized REE patterns of all granitic types show the typical characteristics of A-type granites such as a strong negative Eu anomaly, strong light (L) REE enrichment and a relatively flat heavy (H) REE pattern (Fig. 3.4A to C). Although there are no significant variations in whole-rock geochemistry of major oxides between the main granitic types (Table 3.1), some trace elements do show systematic variations (Fig. 3.5).



Figure 3. 3 A-type discrimination diagrams (after Eby, 1992) for (A) sodic, (B) calcic and (C) biotite granites. A1 field includes granites derived by fractionation of mafic melts whereas A2 field is for granites derived by partial melting of crustal rocks.



Figure 3. 4 Chondrite-normalized REE patterns for (A) sodic, (B) calcic and (C) biotite granites from the Upper Paleozoic plutons of the Cobequid Highlands. Normalizing values after Sun and MacDonough (1989).

At high SiO₂ values, Ba is generally more concentrated in the biotite and calcic granites, whereas the sodic granites are noticeably poorer (Fig. 3.5A). The calcic granites show higher amounts of Cr compared to the sodic and most biotite granites (Fig. 3.5B). Nb is in higher concentrations in the sodic granites, whereas it is lower in the biotite granites

(Fig. 3.5C). The sodic granites are also the most enriched in Zn (>90 ppm, Table 3.1) and Ga (>30 ppm) (Figs. 3.5D and E), whereas both the biotite and calcic granites show similar amounts of these elements (10–100 ppm and 10–35 ppm, respectively). Strontium shows significant scatter, but is generally in higher concentrations in the calcic granites, whereas the sodic granites show the lowest concentrations (Fig. 3.5F). Whole-rock oxygen isotopes from the North River and Cape Chignecto biotite granites have higher $\delta^{18}O$ (6.1–7.6 ‰, Table 3.2) than the granites of the eastern Cobequid Highlands (mean 5.7 ‰).

3.6 **DISCUSSION**

The presence of mineralogically different A-type granites with systematic trace element variations provides the opportunity to better understand the variable petrogenetic conditions that formed these rocks. Since multiple granite types can coexist in the same pluton, it needs to be established whether these different granite types have the same source. Are the observed chemical variations controlled by the nature of the magmatic source, or do they result from geological processes such as varying degrees of partial melting and fractionation? In this section, the mineralogy and geochemistry of the sodic, calcic and biotite A-type granites are discussed further, in order to define the petrogenetic evolution of these rocks.

3.6.1 ESTIMATES OF INTENSIVE PARAMETERS

The magmatic temperatures for the late Paleozoic granites have been estimated by the use of the zircon-saturation thermometer (Boehnke et al., 2013) within an uncertainty of \pm 30-50°C to account for errors in elemental abundances and ratio precision. Saturation of Zr in all these granites is indicated by the presence of magmatic zircon, the calculated M-ratios (molar (Na+K+2Ca)/(Al*Si)) are within the calibration range (from 0.7 to 1.7,



Figure 3. 5 Harker-type geochemical plots of selected trace elements (A) Ba, (B) Nb, (C) Cr, (D) Zn, (E) Ga and (F) Sr for the different granitic types present in the Upper Paleozoic plutons of the Cobequid Highlands. Sodic granites appear quite distinct from the other types in terms of Ba, Nb and Zn, whereas the calcic and biotite granites show significant overlaps except for Cr.

Table 3.3), and therefore the application of this zircon thermometer was considered appropriate. Textural evidence from the subsolvus granites suggests that plagioclase was one of the first phases to crystallize from the melt. Experimental work shows that plagioclase crystallizes as a near-liquidus phase in A-type granitic partial melts (Dall'Agnol et al., 1999). The plagioclase-melt barometer of Putirka (2005) was, therefore, employed to determine the pressure that best approaches the pressure of magma generation (pressure of partial crystallization) in all subsolvus granites. This barometer is based on albite-anorthite exchange and provides pressure estimates within an error of \pm 1.8 kbar.

The initial contents of water in the granitic melt were estimated by the plagioclase-melt hygrometer of Putirka (2005). The error of the estimated water contents with this method may be up to ± 1.2 wt. %. Estimates in the viscosity of the granitic melts were done using the model of Giordano et al. (2008). The associated error of this method for anhydrous (sodic granites) and hydrous melts (calcic and biotite granites) is ± 0.25 and 0.35 log units, respectively.

Most of the studied granites are related to melts with zircon saturation temperatures between 800 and 950 °C (Table 3.3). However, the alkalinity of the melt is a major factor that controls zircon solubility, and consequently affects the temperature estimates. Solubility of zircon is lower in metaluminous and peraluminous melts, and therefore any inherited zircon may be preserved. Miller et al. (2003) argued that zircon-saturation temperature estimates from highly fractionated felsic melts and melts with limited evidence of zircon inheritance represent minimum estimates, whereas the opposite stands for granitoids with abundant zircon inheritance. Dunning et al. (2002) demonstrated that there is only minor zircon inheritance in the granites of the studied plutons, except for their sample J of a late leucogranite from the Pleasant Hills pluton. Therefore, the estimated magmatic temperatures in this work are taken as minimum estimates.

The estimated pressures of partial crystallization from the plagioclase geobarometer range from 3 to 14 kbar for most granites (Table 3.3). There is a systematic variation, however, in the estimated pressures between the calcic and the biotite granites.

Pluton	CCP	NRP	PHP	WP	WP	PHP	PHP	WP	WP	WP	WP	Detection	on limit
Туре	bio	bio	bio	bio	bio	cal	cal	cal	cal	sod	sod	(%	6)
Sample	6284	1779	4672	9827b	36-2-1	4681	4689	6468	6489	4636	6490	А	В
Method	В	В	В	А	В	В	В	в	В	В	В		
<u>Major ox</u>	des (wt.	<u>%)</u>											
SiO ₂	77.10	74.92	75.16	78.23	72.90	70.77	72.45	76.40	77.16	76.15	75.44	0.01	0.03
TiO ₂	0.16	0.19	0.25	0.13	0.22	0.48	0.38	5.30	5.03	0.17	4.88	0.001	0.005
Al ₂ O ₃	12.20	12.81	12.64	11.92	13.76	13.18	13.17	11.88	12.09	11.58	11.81	0.01	0.03
Fe ₂ O ₃	1.13	1.97	2.42	1.51	1.85	4.09	3.23	2.17	1.90	2.53	2.67	0.01	0.03
MnO	0.10	0.01	0.02	0.02	0.03	0.07	0.07	0.01	0.03	0.04	0.04	0.001	0.005
MgO	0.22	0.19	0.83	0.13	0.89	1.13	0.99	0.15	0.09	0.82	0.13	0.01	0.03
CaO	0.68	0.16	0.26	0.44	0.44	0.91	0.65	0.17	0.16	0.21	0.18	0.01	0.03
Na ₂ O	5.10	3.59	3.74	3.40	4.00	3.45	3.89	3.39	3.33	4.45	3.68	0.01	0.03
K_2O	2.92	5.45	4.79	4.80	5.55	5.49	5.11	0.17	0.35	4.65	0.33	0.01	0.03
P_2O_5	0.04	0.02	0.04	0.02	0.03	0.09	0.07	0.02	0.03	0.01	0.03	0.01	0.005
LOI	0.30	0.31	0.40	0.13	0.40	0.20	0.30	0.49	0.39	0.30	0.56		С
Trace ele	ements (p	<u>pm)</u>											
Method	А	Α	С	А	С	С	С	А	А	С	А		
Ba	504	529	186	64	285	360	223	152	108	39	73	3	25
Rb	180	32	211	226	240	206	262	174	180	224	203	1	2
Sr	54	66	26	36	37	51	27	14	10	10	8	2	5
Y	39.8	63.4	111	55.9	60	78	88	50.5	57.0	90	60.1	0.5	3
Zr	273	162	337	273	212	831	361	365	202	552	446	1	4
Nb	19	20	50	56	23	37	48	55	52	69	43	0.2	l
Pb	59	13	20	6	17	3	45	8	9	3	18	5	3
Ga	21	18	25	31	20	25	24	28	27	30	35	1	5
Zn	b.d.	60	22	17	32	55	80	21	52	131	90	1	5
Cu	b.d.	b.d.	b.d.	10	b.d	b.d	2	7	10	0	7	l	4
N1	b.d.	b.d.	10	5	5	12	9	, 5	, 5	4	6	I T	3
V	8	8	6	. 7	21	10	9	b.d.	b.d.	2	b.d.	5	4
Cr	b.d.	b.d.	44	b.d.	21	3/	29	40	30	32	40	20	4
La	68.40	28.90	109.0	33.20	46.0	91.2	//.8	41.50	40.30	/1.2	43.90	0.05	0.5
Ce	122.0	97.10	124	97.20	102	213	169	112.0	102.0	152	118.0	0.05	3
Pr NJ	15.80	/.26	n.a.	9.44	n.d.	n.a.	n.a	12.60	12.20	n.d.	13.60	0.01	-
Na	50.50 10.00	25.30	82	33.30	n.d.	8/	08 12.7	51.00	49.60	08 15 2	56.40	0.05	5
Sm E	10.90	0.10	1/.2	8.41	8./	1 / .2	13./	10.90	0.(1	15.5	13.10	0.01	0.1
Eu	0.90	0.34	1.49	0.33	0.62	1.72	1.1	0.79	0.01	0.0	$\frac{0.71}{12.00}$	0.005	0.2
Ga Th	8.45	0.05	n.a. 2.51	1.99	n.a.	n.a.	n.a	10.30	10.90	n.d.	12.00	0.01	0.5
10 Du	1.32 8.10	1.50	5.51 m.d	1.74	1.04 nd	2.95 nd	2.0	1.78	1.95	2.3 nd	2.21	0.01	0.5
Dy Lo	0.19	9.93	n.a.	2 20	n.a.	n.a.	n.u n.d	2.12	2.30	n.d.	2.61	0.01	
П0 Бе	1.50	2.09	n.d.	6.02	n.u.	n.d.	n.u	2.15	2.40	n.d.	2.01	0.01	
EI Tm	4.07	0.10	n.d.	0.95	n.u.	n.d.	n.u	1.04	7.00	n.d.	1.39	0.01	
Thi Vh	0.39	6.56	12.0	1.10 8.22	11.u. 5.8	11.u. 8 0	0.3	1.04	7.80	10.7	1.10 8.41	0.005	0.2
IU In	0.62	1.00	1 62	0.55	0.85	1.27	1.22	1.06	1.20	167	1 20	0.01	0.2
Lu Co	1.00	84.0	0.63	1.40 h.d	0.05 h.d	1.2/ h.d	1.32 h.d	1.00 h.d	1.20	1.07 nd	1.29	1	5
Co	1.00	04.0	2 1	0.u. 24	nd	3 D.U	0.u. 3 1	0.u. 2 1	3 2	n d	2^{1}	01	0.2
US Hf	т.5 7 Л	5.5	∠. 4 12.7	∠. 4 8.5	7.0	5.∠ 22.4	12.6	$\frac{2.1}{12.2}$	5.5 8 7	n d	∠.1 12.6	0.1	0.2
Sc	nd	5.5 n d	2 13	1.00	3 23	4 15	3 68	12.2	1.0	n d	12.0 h.d	0.1	0.2
Та	n d	n d	55	5.80	26	30	5.00	4.15	3.97	n d	3.06	0.1	0.01
Th	20.4	24 5	23.0	197	22.0	17.0	21.0	157	174	16.0	11.8	0.01	0.3
U	n.d.	n.d.	4.3	4.43	n.d.	3.6	4.2	3.28	3.06	5.9	2.88	0.01	0.5
-						2.0			2.00	~ • • /		U.U I	···

Table 3. 1 Representative whole-rock analyses of A-type granites from the Cobequid Highlands.

Note: All major oxides were measured with XRF, trace elements were determined with INAA and ICP-MS. A: Major and trace elements analyzed by ICP-MS, B: Major elements analyzed by XRF, C: trace elements from Ba to Cr and Co analyzed by XRF, REE and elements from Cs to U by INAA. Abbreviations: b.d.=below detection limit, n.d.= not determined, bio= biotite granite, cal= calcic granite, sod= sodic granite, CCP= Cape Chignecto pluton, NRP= North River pluton, PHP= Pleasant Hills pluton, WP= Wentworth pluton.

		Late Pal	leozoic (granites		Late Pal	eozoic 1	<u>ocks</u>				Neopro	oterozoia	<u>c rocks</u>		
nit	CCP	NRP	WP	WP	WP	WP	Fou Lake	ntain Group		Jeffe	<u>rs Block</u>			Bas	s River Bl	<u>ock</u>
ample	3927	1779	4641	4636	6419	9826	4601	6061	1347	3709	2106	3500	449	3135	3171	4156
ype	bio	bio	cal	sod	pos	gabbro	bas	bas	gabbro	quartz diorite	grd	grd	rhy	gabbro	grd	granite
tensive pai	ameters															
Zr (± 30-																
0 °C)	814	805	874	891	919	I	I	I	I	I	I	I	I	I	I	I
$(\pm 0.18$																
-Pa)	12	7	5.4	Ι	Ι	Ι	Ι	Ι	I	I	Ι	Ι	I	I	Ι	I
2Oin melt (±																
2 wt. %)	5.28	3.86	3.97	Ι	I	I	Ι	I	Ι	I	Ι	I	Ι	I	I	I
Aelt fr.																
(0)	27.5	28.2	37.8	I	I	Ι	I	I	I	I	Ι	I	Ι	Ι	Ι	I
otopic con	positior	<u> </u>														
vd (360)	+2	+0.2	+2.5	+3.6	$^{+3.1}$	+3.8	+3.4	+3.9	+3.6	+2.5	-0.6	-4.7	-0.4	+0.6	-1.75	-2.05
¹⁸ O (‰)	7.6	6.1	5.7	5.5	6.0	4.5	I	I	I	I	Ι	I	I	I	I	I

Table 3. 2 Summary of intensive parameters, for the late Paleozoic A-type granites, and available isotope data for the late Paleozoic and Neonroterozoic rocks of the Cobequid Highlands.

minor variations in both late Paleozoic and Neoproterozoic rocks, and are therefore taken to represent primary magmatic contents. Abbreviations: bio= biotite granite, cal= calcic granite, sod= sodic granite, bas= basalt, grd= granodiorite.

The latter appear to yield higher pressures (9 to 13 kbar) than the calcic granites, for which pressures as low as 3 kbar were estimated (Fig. 3.6). The biotite granites also show higher amounts of initial H₂O content in the melt than the calcic granites (Tables 3.2, Appendix B.1). The estimated pressures from plagioclase are significantly higher than those from Al-in-hbl barometer (2–3 kbar: Chapter 2). Textural evidence suggests that plagioclase in the calcic granites crystallized before amphibole. Therefore the large differences in pressure estimates between the two barometers may reflect a rapid ascent from deep crustal levels, with Al-in-hbl estimates reflecting the level of emplacement of these plutons.

3.6.2 DID THE MAIN GRANITE TYPES HAVE A SIMILAR SOURCE?

The available geochronological data from the Wentworth, Pleasant Hills and Cape Chignecto plutons (U-Pb on zircon, Doig et al., 1996; Dunning et al., 2002) suggest that the A-type granites and their extrusive equivalents are contemporaneous and emplaced at ca 360 Ma. Furthermore, the ages yielded by both zircons and amphiboles in these rocks are quite comparable and do not show an age progression that could be related to closure temperatures, suggesting a rapid unroofing and cooling of the plutons (Pe-Piper et al., 2004). The sodic, calcic and biotite granites differ in their Nd-isotope composition (Table 3.2). The sodic granites show the highest values of $\varepsilon_{Nd(360 Ma)}$ (>+3), the calcic granites range from +1.1 to +2.7, whereas the biotite granites show the lowest ε_{Nd} values, from +0.2 to +2.0 (Pe-Piper and Piper, 1998; Papoutsa and Pe-Piper, 2014). Although there is a great overlap in ε_{Nd} between the calcic and biotite granites, the hypersolvus sodic granites have systematically higher ε_{Nd} and a distinct trace element signature (Fig. 3.5), which suggest that the sodic granites are products of a different magmatic source. With the exception of the mafic minerals and specific trace elements, the biotite and calcic granites present a rather similar mineralogy and whole-rock geochemistry, which could imply a comagmatic origin modified by fractionation or assimilation. If the biotite granites were fractionates of the calcic granites, they should be more enriched in incompatible elements and silica as more evolved melts. In Harker-type diagrams, however, the biotite and the calcic granites do not show a continuum, rather they present two parallel trends (Fig. 3.5), with no significant variation in highly incompatible elements, including Ga which can be used as a fractionation index in granitic systems (Breiter et al., 2013).

Alternatively, the variations between the calcic and biotite granites could be attributed to varying degrees of crustal assimilation of an originally similar melt. The granites show no evidence for substantial upper crustal assimilation. This conclusion is based on the following: a) lack of xenoliths and xenocrysts in the granites; b) absence of reaction textures such as corona rims; c) the complete absence of calcic and sodic granites in the homogeneous biotite granite plutons in the western Cobequid Highlands; d) the rapid emplacement and cooling of the plutons as indicated by geochronology and by the presence of petrologically similar granitic clasts in the sediments of the coeval Horton Group, and e) the chemical similarities with the coeval rhyolites, which represent the melts with the most limited interaction with the crust.

Assimilation at deeper crustal levels is more difficult to evaluate. If the Avalonian lower crust evolved in the late Neoproterozoic during complex subduction, then felsic differentiates of underplated mafic magma would be susceptible to assimilation in the late Paleozoic. Such a process is difficult to distinguish from partial melting of the same rocks and is evaluated further below.

Pluton	Туре	sample	Tzr (°C)	M value	P (GPa)	Water wt%	Viscosity (Pa*s)
ССР	biotite	2039	851	1.3	0.94	5.2	9.78
ССР	biotite	3920	856	1.2	1.08	5.0	9.91
NRP	biotite	1675	801	1.4	1.5	6.5	10.77
NRP	biotite	7071	797	1.5	1.3	6.0	10.79
NRP	biotite	7082	794	1.5	1.3	6.6	10.76
PHP	biotite	4672	861	1.3	1.36	5.2	9.59
PHP	biotite	4690	833	1.4	0.94	5.5	9.83
PHP	biotite	4704	772	1.6	0.31	7.4	11.82
PHP	biotite	7385	842	1.5	0.73	5.7	9.91
PHP	calcic	4670	889	1.4	1.11	4.8	8.9
PHP	calcic	4681	941	1.4	1.30	4.1	8.32
PHP	calcic	4689	856	1.1	0.83	5.0	9.76
PHP	calcic	7393	859	1.5	1.21	5.7	9.61
WP	biotite	6970	866	1.1	1.21	4.6	10.03
WP	biotite	7372	884	1.4	1.52	4.2	9.41
WP	calcic	6969	921	0.7	1.34	3.9	9.7
WP	calcic	7658	864	1.1	1.28	5.6	9.73
WP	sodic	6518	1045	1.6	-	_	7.08
WP	sodic	6419	920	1.4	-	_	9.03
WP	sodic	4636	891	1.5	_	_	9.42

Table 3. 3 Estimated magmatic parameters for the A-type granites of the Cobequid Highlands.

Notes: Abbreviations for pluton names as in Table 1. Magmatic temperatures are given within an error of \pm 30-50 °C and pressures in an error of \pm 0.18 GPa. Water estimates may be within an error of \pm 1.2 wt%. Pressures and water contents were not estimated for the sodic granites due to the lack of magmatic plagioclase. Viscosity estimated using the model of Giordano et al. (2008). Biotite and calcic granites are within an error of \pm 0.35 log units and viscosities for sodic granites have an associated error of \pm 0.25 log units.

Evidence of magma mixing between contemporaneous mafic and felsic melts is widespread in the Wentworth pluton, as seen from local development of mingling textures, and enclaves of gabbro with lobate contacts in the granite. Rocks of granodioritic composition, one end result of such mixing, are very rare. Samples from outcrops with any field evidence of mixing were not included in this study. However, evidence of interaction between felsic and mafic melts are potentially recorded in the geochemistry of the calcic granites. This is the only granite type that shows a positive trend between ε_{Nd} and CaO/Na₂O+K₂O, which could result from either mixing with gabbro or mantle-derived fluids (Fig. 3.7). The \mathcal{E}_{Nd} in the calcic granites correlates positively with the estimated



	CCP	NRP	PHP	WP
Biotite granite	\$			•
Calcic granite				•

Figure 3. 6 Pressure (estimated from plagioclase-melt barometer) vs. H₂O-in-melt concentrations, demonstrating the systematic differences of those parameters between the calcic and biotite A-type granites.

magmatic temperatures and negatively with pressure and viscosity (Fig. 3.8). The variations in the H_2O contents in the melt are too small to account for a systematic decrease in viscosity and do not correlate with \mathcal{E}_{Nd} , precluding substantial mixing with mantle-derived fluids. Ruprecht and Bachmann (2010) interpreted such correlations between viscosity and temperature as the result of heating by mafic melts. The negative trend that is observed for pressure suggests that such heating was more prominent at upper crustal levels. Textural evidence of substantial mixing (such as reverse chemical zoning of minerals and presence of mafic clots) in the studied granites is lacking; furthermore, the

only covariation with \mathcal{E}_{Nd} is that of CaO and the alkalies. Therefore substantial assimilation of mafic melts by the granitic melts is unlikely. It is possible that the observed isotopic and chemical variations reflect magma mixing at the source (such as a hybrid granodiorite), coupled with later heating by a coeval mafic melt after the calcic granitic melt was produced.



Figure 3. 7 ϵ_{Nd} vs. CaO/Na₂O+K₂O plot showing the possible mixing trend of the calcic granites. Upper Paleozoic gabbro and rhyolite are also plotted for comparison.



Figure 3. 8 ε_{Nd} vs. pressure (A), temperature (B), and estimated viscosity (C) for the calcic granites.

Taken together, the differences in Cr, Nb (Fig. 3.5), trends of \mathcal{E}_{Nd} (Fig. 3.7),

magmatic temperature (Table 3.3), and trends between pressure and magmatic H₂O contents (Fig. 3.6, Table 3.3) point to an origin from different magmatic sources. In addition, the distinct spatial clustering of the granitic types between the western plutons (only biotite granites in North River and Cape Chignecto) and the eastern plutons (dominant calcic granite, Wentworth and Pleasant Hills), suggest that the calcic and the biotite granites, even though are coeval, probably do not have the same source.

3.6.3 POTENTIAL MAGMATIC SOURCES

A-type granites are interpreted to be derived from magmas that have an exclusively igneous source (Whalen et al., 1987), whether by partial melting of granodiorite or tonalite, or extreme fractionation of mafic magma (Loiselle and Wones, 1979; Frost and Frost, 2010). In the Cobequid Highlands and nearby Avalonian Antigonish Highlands, the only older igneous rocks are the Neoproterozoic arc-related rocks and minor Ordovician A-type granites in the Jeffers Block and Antigonish Highlands, and the Neoproterozoic arc-related and minor ocean-crust rocks of the Bass River block, all with well documented geochemistry and radiogenic isotopes (Pe-Piper and Piper, 1998; Escarraga et al., 2011; MacHattie and White, 2012). The Ordovician A-type granites of the Antigonish Highlands are less REE-enriched than the late Paleozoic granites of the Cobequid Highlands, and contain calcic, sodic-calcic amphibole and hedenbergite, with only minor biotite and no sodic amphibole (Escarraga et al., 2011).

Both biotite and calcic granites are classified as A2-type granites in Eby's (1992) diagrams, suggesting that they are probably not the fractionates of mafic melts. Furthermore, a crustal source for the biotite and calcic granites may be indicated by their

metaluminous to peraluminous composition (Fig. 3.2B), and by the range of \mathcal{E}_{Nd} for these rocks which is similar to that of Neoproterozoic intermediate rocks (Table 3.2). Using the whole-rock composition of Neoproterozoic intermediate rocks, and applying batch partial melting (Shaw, 1970) geochemical models for specific highly incompatible trace elements (such as Nb, Rb, Th, and U), it is suggested that both biotite and calcic granites could have been formed by similar degrees of partial melting, ranging between 20% and 40% of extracted melt (Appendix B.1). Therefore, the observed systematic geochemical differences between the two granitic types are likely the result of source rocks of different composition, or differences in deep-crustal processes such as assimilation, since a significant role for upper crustal mixing or assimilation has been ruled out.

The sodic granite type is only present in the Wentworth pluton and is geochemically quite distinct from the other main granite types. Not only it is closer to the late Paleozoic and Neoproterozoic mafic rocks of the Jeffers Block in Nd isotopes compared to the other granite types (Table 3.2), but also shows strong depletion in Ba and Sr, and enrichment in Ga, which are indicative of strong fractionation. The sodic granites, however, show the highest amounts of Zn compared to the other granitic types (> 90 ppm, Table 3.1). Zinc is generally more concentrated in mafic melts, and many A-type granites interpreted as partial melts of intermediate feldspathic rocks have lower amounts of Zn (Jung et al., 1998; Wenner and Lloyd, 2006) similar to the biotite and calcic granites. In contrast A-type granites generated by fractionation of mafic melts appear more Znenriched (Vander Auwera et al., 2003; Shellnutt and Iizuka, 2011). Thus the trace element geochemistry along with the high \mathcal{E}_{Nd} values of Neoproterozoic and late Paleozoic mafic rocks in the Cobequid Highlands suggests that the sodic granite is a high-degree
fractionate of a mafic parent magma. Fractionation models for the sodic granites were performed for major and trace elements. Fractionation sequences for selected magmatic sources were thermodynamically simulated with the use of PELE (Boudreau, 1999) which is a minimization of the MELTS algorithm developed by Ghiorso and Sack (1995). The output data from PELE were used to model trace elements through Rayleigh fractionation as applied by Neuman et al. (1954). Detailed description of the models and data used are provided as supplementary material (Appendices B.2, B.3). Fractionation models for several trace elements (Fig. 3.9), suggest that the sodic granites could have been derived by extreme fractionation (90%) either of a source resembling the low-Ti mafic rocks of the Jeffers Block or a source chemically similar to the Fountain Lake basalt (Fig 3.9). Only an origin through fractionation of Late Paleozoic mafic magma is geologically feasible.

In the Cobequid Highlands, basalt in three main extrusion cycles (Dessureau et al., 2000) is dated by zircon in interbedded rhyolite between 358 ± 1 Ma and 355 ± 2 Ma (Dunning et al., 2002), but associated mafic plutonic rocks are older, with five amphibole ages between 367 and $358\pm<4$ Ma (Pe-Piper et al., 2004) and one zircon age of 370 ± 2 Ma (Gibbons et al., 1996). The top of the 1 km of basalt in western Cape Breton Island is dated at 371 ± 1 Ma (Dunning et al., 2002). There was thus mafic magma available several million years before emplacement of the sodic granite (362 ± 2 Ma, Dunning et al. 2002; 368 ± 4 Ma, Pe-Piper et al., 2004). Evolved rhyolitic melts can be produced from fractionation of mafic magma in the deep crust, with minor crustal assimilation, in less than 2 Ma (Turner et al., 1992; Annen and Sparks, 2002, Lowenstern et al., 2006). Stratigraphically, the rhyolites of the Fountain Lake Group are older than the main phase of the basalts, and chronologically, the granites of the eastern plutons are older or coeval

with some of the gabbros. These suggest that some late Paleozoic mafic magma may have been trapped deep in the crust or upper mantle and the timing of basaltic volcanism was related to faulting producing pathways through the crust rather than the availability of mafic magma from the upper mantle.



Figure 3. 9 Fractionation trace element modeling of selected isotopically suitable mafic sources for the sodic granites.

3.6.4 CONSTRAINTS ON PARTIAL MELTING FOR THE BIOTITE AND CALCIC A-TYPE GRANITES

The alkalinity (Fig. 3.2) and trace element classification (Fig. 3.3) of the late Paleozoic biotite and calcic A-type granites from the Cobequid Highlands suggests an origin through partial melting of crustal rocks as the dominant process. Therefore, in order to understand the origin of the observed trace element variations between these two granite types, four main factors need to be evaluated: a) the composition of the magmatic source, b) the control of various degrees of partial melting, c) the role of mixing with mafic magma in the deep crust and d) the role of crustal assimilation particularly in the deep crust. As previously mentioned, the biotite and calcic granites appear to have been derived





Figure 3. 10 Plots of the modeled fraction of extracted melt for the biotite and calcic granites vs (A) Cr, and (B) Zn, demonstrating the limited control of partial melting on the systematic variations of these elements.

Calcic granites are found only in the two larger complex plutons along the Rockland Brook Fault. The higher geothermal gradient implied by the higher magmatic temperatures (Table 3.3) may be related to the abundance of gabbros in this area, reflecting greater supply of mafic magma to the base of the crust and producing a large gravity anomaly and thick Fountain Lake Group volcanic successions (Pe-Piper and Piper, 2003). The higher Cr content of the calcic compared with biotite granite is unlikely to be due to mixing with gabbroic magma, since the Zn content of the two granite types is similar (Fig. 3.10).

Models of partial melting (Appendix B.2) of Neoproterozoic quartz diorites, granodiorites and tonalites from both the Jeffers and Bass River blocks (Appendix B.4) produce the best fit for chondrite-normalized REE patterns at 30-40% fraction of extracted melt (Fig. 3.11). The Bass River block tonalites, however, show systematically higher Cr, and depletion in Sr, in their whole-rock chemistry (Appendix B.5) and consequently produce Cr-enriched, Sr-depleted partial melts, compared to their equivalents from the Jeffers Block (Fig. 3.12A and 3.12B). The partial melts from the Bass River Block are more enriched in HFSE and depleted in LILE, matching broadly the characteristics of the calcic granites (Fig. 3.12C).



Figure 3. 11 Chondrite-normalized patterns of modeled REE, under (A) 20%, (B) 30%) and (C) 40% of partial melting of Neoproterozoic intermediate rocks. Shaded areas correspond to the REE-patterns of the granites.



Figure 3. 12 Plots of modeled (A) Cr, and (B) Sr vs. modeled Zr during 40% of partial melting of Neoproterozoic intermediate rocks. Panel (C) shows the modeled concentrations of total high field strength elements (HFSE) vs. total large ion lithophile elements (LILE). The ranges of these elements measured in the biotite (dashed lines) and calcic granite (solid lines) are plotted for comparison.

The intermediate plutonic rocks of the Jeffers and Bass River blocks differ in major oxides only in that the granodiorites and tonalites of the former appear to be more Al-rich (Appendix B.5). The peraluminous biotite granites show lower estimated magmatic temperature and higher water contents than the calcic granites. The estimated magmatic H₂O contents of the melt for the biotite granites correlate positively with increasing fraction of extracted melt (Fig. 3.13A), whereas magmatic temperatures show a negative correlation (Fig. 3.13B). The exact opposite trends are observed for the calcic granites. These data suggest that partial melting that produced the biotite granites was enhanced by the presence of H₂O (fluid-present melting), whereas the production of the calcic granitic melts was predominately controlled by high temperatures (dehydration melting). Holtz and Johannes (1994) argued that at moderate crustal depths, melts at ~750 °C require 6 wt.% H₂O, and for melt fractions of ~40% an additional 1–2 wt.% of H₂O is required. Although the temperature estimates for the biotite granites are higher than those for the "cold



Figure 3. 13 Plots of (A) estimated water-in-melt amounts and (B) zircon-saturation temperatures vs. the degree of partial melting for the calcic and the biotite granites, demonstrating the control of water availability and temperature on the fraction of produced melt.

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Calcic granite

granites" of Miller et al. (2003), the observed variations of temperature and H₂O with increasing fraction of extracted melt may favor a scenario similar to that of Miller et al. (2003), in which influx of H₂O is generated during dehydration of crustal lithologies. In hydrous alkaline melts, the stability of biotite is restricted to temperatures below 700°C (Scaillet and Macdonald, 2001). Calcic amphibole, on the other hand, is stable under higher magmatic temperatures, above 800 °C, and below this temperature reacts with the melt to form biotite and titanite (Dall'Agnol et al., 1999). The low magmatic temperatures

and the availability of Al and H_2O in the crustal rocks during partial melting might have favored a preferential crystallization of biotite over amphibole in the A-type granites of the central and western Cobequid Highlands.

The hypothesis that mafic magma contributed significantly to the calcic or biotite granite melts through magma mixing can be excluded, as there is no consistent pattern of trace elements associated with mafic melts (Fig. 3.10) and no mixing is apparent from variations in Nd isotopes with particular elements for the biotite granites (Fig. 3.7). It is likely that crustal assimilation played a role in the deep crust in modifying the partial melts of Neoproterozoic tonalite and granodiorite as they rose through the crust, but such a process would have enhanced the element and isotopic signatures already derived by partial melting from the different Neoproterozoic crustal blocks and is thus difficult to quantify.

Experimental work of Gardien et al. (2000) demonstrated that hornblende-bearing granitic rocks can be the partial melting products of plagioclase-biotite-quartz mineral assemblages. Mogk (1992) argued that formation of hornblende-bearing migmatites in tonalitic gneisses can be the result of incongruent melting of biotite-plagioclase assemblages according to the reaction

Bt+Pl+Qz+Ep+fluid = Hbl+Pl+Kfs-rich granitic melt. (1)

Mineral assemblages similar to the reactants of the above equation have been observed in the tonalites of the Bass River Block (Pe-Piper et al., 2010) although they are not migmatitic. Outcrop of the Bass River Block is limited to the area south of the Rockland Brook fault (Fig. 3.1B) and thus south of the Pleasant Hills and Wentworth plutons where the calcic granite is found.

3.6.5 GEOLOGICAL SYNTHESIS

The large volumes of gabbro in the Wentworth and Pleasant Hills plutons and the 1.5 km thick basalt in the Fountain Lake Group (Pe-Piper and Piper, 2003) suggest the presence of a major heat source in the eastern part of the Cobequid Highlands. The elemental and isotopic characteristics of the most primitive Fountain Lake basalts were interpreted by Dessureau et al. (2000) as indicative of an origin from asthenospheric and subcontinental mantle with elevated potential temperature suggesting mantle upwelling. It is therefore likely that in the late Paleozoic there was substantial interaction between the lower Avalonian crust and shallow upper mantle at the fault-bounded margin of the extending Magdalen Basin.

Magnetic data from the Cobequid Highlands (Piper et al., 1993) show that the eastern part of the Cobequid Highlands has high amplitude and long-wavelength magnetic anomalies, which was taken to suggest that the plutons in the area are underlain by gabbro. This interpretation is supported by a high gravity anomaly beneath the Rockland Brook fault in the southern Wentworth pluton (Koukouvelas et al., 2002). This gabbro may be related to the underplated mafic layer of high velocity identified from seismic refraction in the centre of the extensional Magdalen Basin (Marillier and Reid, 1990). Its surface expression may be the voluminous gabbroic body in the Wentworth pluton. The western part of the Cobequid Highlands, however, shows significantly lower magnetic and gravity anomalies, suggesting that this area was beyond the limits of significant mafic underplating (Piper et al., 1993). Published Pb isotopes for the late Paleozoic Avalonian rocks of the Cobequid Highlands (Pe-Piper and Piper, 1998b) show that the granites in the eastern part of the area (Pleasant Hills and Wentworth plutons) are generally more

enriched in ²⁰⁶Pb compared to the other plutons suggesting the presence of newly-formed crust (Rollinson, 1993).

It has been proposed that the mafic underplating of Magdalen Basin and the extrusion of the late Paleozoic continental tholeiites were related to a NW-migrating mantle plume (Dessureau et al., 2000). An origin for the continental tholeiites within a mantle plume was suggested on the basis of geochemistry (high FeO_T suggesting high potential temperature) although the isotopes and trace elements suggest significant contamination from the lithospheric mantle (Dessureau et al., 2000). Based on the timing of the similar tholeiitic igneous activity that followed from Visean to Permian in western Europe it was hypothesized that the plume was originally active in the Maritimes in the mid-Devonian (Keppie and Krogh, 1999) that later moved beneath Scotland and southern Norway (Dessureau et al., 2000; Wilson et al., 2004). The timing of these events was also synchronous with, and followed, the ongoing closure of the Rheic ocean by NW-directed intra-oceanic subduction beneath the Canadian Appalachians (Nance et al., 2012). The latter could result in mantle upwelling and mafic underplating by a number of processes, including slab detachment (Fig. 3.14A and B), perhaps aided by a pre-existing plume and the crustal extension in the Magdalen Basin. Whatever the mechanism, a deep crustal hot zone (DCHZ), such as that proposed by Annen et al. (2006) and Hyndman and Currie (2011), would have formed. In such zones, the felsic melts may be derived by two possible mechanisms: 1) partial melting of pre-existing crustal rocks, and 2) fractional crystallization of mafic sills emplaced in the lower crust.



Figure 3. 14 Suggested evolution of the late Paleozoic A-type granites of the Cobequid Highlands. Images (A) and (B) show the presence of a metasomatic reservoir in the upper mantle, and the mafic underplating following slab detachment during the closure of the Rheic Ocean. Image (C) shows the production and emplacement of the various A-type granite types from a deep crustal hot zone and the adjacent area (see text for details).

According to Annen et al. (2006), the repeated injection of hydrous basaltic melts into the

crust leads to continuous fractionation of emplaced mafic sills (Fig. 3.14C), associated

with release of H₂O and heat, promoting partial melting of the surrounding crust. The

felsic melts ascend from the hot zone to reservoirs in the upper crust, leaving behind the dense restites and cumulates. Crustal-scale strike-slip faults in the Cobequid Shear Zone may have provided effective pathways for magma ascent. This is indicated a) by the spatial relationship between the plutons with syn-magmatic deformation, and the major faults regionally (Koukouvelas et al., 1996), b) the earliest magmatic phases are segregated along distinct faults as reported from the Pleasant Hills pluton (Pe-Piper et al. 1998) and, c) prefull crystallization structures, such as lineation of amphiboles in the Wentworth pluton, show a systematic parallelism to the Rockland Brook fault suggesting an along-fault flow of magma (Koukouvelas et al., 2002). In the Cobequid Highlands, however, the mineralogy of the extruded late Paleozoic tholeiites suggests they were rather anhydrous melts. The coeval mid-crustal gabbros of the Wentworth pluton (3-5 kbar), on the other hand, appear substantially more hydrated (5-6 wt.% H₂O, Zhang personal communication, 2015). The DCHZ in the Cobequid Highlands thus deviates from the proposed model of Annen et al. (2006) in that the underplated mafic magma was originally anhydrous. However the repeated injection of mafic melts could increase the volatile contents through remelting of previously emplaced sills and continuous fractionation, and thus trigger partial melting of the crust. We suggest that such mechanisms are involved in the Cobequid Highlands. Furthermore, we propose that H₂O present in the hydrous phases of crustal rocks such as micas and amphiboles, was released upon heating and further induced partial melting.

In the eastern part of the Cobequid Highlands, the large variety in the granitic types present in the plutons of the area indicates derivation from such a hot zone, where the larger amounts of underplated mafic melts led to larger volumes of produced granitic

magmas. Crustal-derived biotite and calcic granites of the eastern Cobequids were produced at mean pressures between 9.6 and 13 kbar (Table 3.3, plagioclase-melt barometer), which corresponds to a 32–40 km depth in the continental crust. For comparison, the estimated crustal thickness, based on seismic refraction data (excluding sediment cover and underplated mafic rocks) in the central Magdalen Basin is 36 km (Marillier and Reid, 1990), probably rather thicker at the margins. The depth at which the granitic melts were produced, was therefore, at the base of the crust, and matches the depth proposed for deep crustal hot zones (Annen et al., 2006). The high temperatures in the area induced partial melting of the Avalonian crustal rocks, producing granitic melts that could then be emplaced at shallow crustal levels to form subsolvus granites. The moderate to low amounts of H₂O in the crustally-derived melts of the area could either be coming from lower crustal rocks, or released upon heating of mid-crustal lithologies as the melts migrated. Experimental work of Singh and Johannes (1996) demonstrated that at midcrustal pressures (~5 kbar) dehydration melting of tonalitic rocks starts at approximately 740 °C. The extreme fractionation of the underplated mafic sills, however, produced exclusively hypersolvus granites, due to their originally anhydrous nature. Even though all the studied subsolvus granites of the eastern Cobequid Highlands have a crustal trace element signature, they show a range of δ^{18} O from 5.5 to 5.8‰ (Table 3.2), which is lower than the majority of Avalonian granitoids (Whalen et al., 1996). This suggests that the rocks comprising the Avalonian lower crust and supplying oxygen to the melt were principally magmatic as suggested by seismic velocities (Marillier et al., 1994). The lower δ^{18} O values of these granites could also be the result of hydrothermal alteration (Simon,

1990). However the analyzed samples show very limited evidence of hydrothermal alteration and therefore the δ ¹⁸O values are taken to be representative of the parent melts.

Experimental work of Castro et al. (1999) on partial melting of crustal rocks has demonstrated the potential role of volatiles in the generation of peraluminous granitic melts and associated phase relations. At pressures of 15 kbar, the volume of extracted melt from anhydrous melting of a granodiorite gneiss and a biotite gneiss was generally less than 5 and 8%, respectively. Addition of 2 wt.% H₂O in the runs for the biotite gneiss increased the volume of extracted melt and stabilized plagioclase, absent in the anhydrous runs. Furthermore, plagioclase and K-feldspar in all runs were the earliest phases to be consumed in the starting materials. Their work may suggest that the higher amounts of Ba in the biotite and calcic granites could reflect the early melting of plagioclase and Kfeldspar in the source, and thus enrichment of the melt, in contrast to the fractionation process of a K-feldspar-poor source, such as the one proposed for the sodic granite. Furthermore, in order to attain the observed fractions of partial melts and crystallize early plagioclase, small amounts of H₂O must be added to the system.

The relative abundances of the produced crustal and residual fractionated granitic melts in deep crustal hot zones depend greatly on the emplacement rate of the intruded mafic magma (Annen and Sparks, 2002; Solano et al., 2012). In their models, Solano et al. (2012) demonstrated that at high emplacement rates of mafic magma, evolved melt segregation occurred in the overlying crust, and the produced granitic melts were predominantly derived from partial melting of the crustal rocks. The opposite was observed for low emplacement rates, where melt segregation occurred within the underplated layer and the granitic melts were produced predominantly through

fractionation. In the eastern Cobequid Highlands, the larger volumes of crustally-derived calcic and biotite granites, compared to the less abundant sodic granites, are consistent with high emplacement rates of mafic magma, and may, therefore, be related to the late Paleozoic mafic underplating. In these cases, the elevated temperature would affect a rather large volume of the crust, with the felsic melts forming initially as batches at various depths (Solano et al., 2012) (Fig. 3.14C).

If felsic melts are segregated through fractures, which can effectively drain the crustal hot zone, then the produced plutons would have resulted from the mixing of various individual batches of partial and residual melts, and would, therefore, have a polybaric geochemical signature (Solano et al., 2012). In the complex plutons of the eastern Cobequid Highlands the presence of different coeval granitic types of different sources indicates multiple batches of granitic melt. The systematically different pressures for the various granitic types (Fig. 3.6) within these plutons (Appendix B.1) suggest that these granitic batches probably originated from different depths, and segregated along the crustal-scale Rockland Brook Fault (Fig. 3.14C). The presence of multiple granite types in these plutons, particularly in the Wentworth pluton, is consistent with the concept of of crustally-derived and residual felsic melt batches that result from generation in deep crustal hot zones (Annen and Sparks, 2002).

In the central and western Cobequid Highlands, however, the H₂O in the granitic melt is in relatively higher amounts (Fig. 3.6) and has a different affinity, as indicated by the difference in oxygen isotopes (Table 3.2). Although some H₂O may have been derived from fractionating mafic melts, the systematically higher values of δ ¹⁸O (Table 3.2), if primary, suggest a difference supply from the crust of the western part of the Cobequid

Highlands, dominated by the sedimentary and volcaniclastic rocks of the Jeffers Group. Water released from these rocks would have a different isotopic signature from H₂O released from magmatic rocks (Rollinson, 1993). These relatively higher values of δ ¹⁸O are not consistent with hydrothermal alteration that would involve meteoric H₂O, since such post-magmatic processes result in the reduction of δ ¹⁸O (Simon, 1990).

The chemical affinities of A-type granites worldwide have been the subject of several detailed studies, and it is well established that these rocks may form by different geological processes. This study shows that it is possible for different geological processes to create mineralogically distinct A-type granites. The variations in the ferromagnesian minerals present in the A-type granites of the Cobequid Highlands result from source heterogeneity, mode of emplacement, and processes of melt generation. The granites with sodic amphibole are associated with fractionation of mafic melts, the granites with calcic amphibole were derived through dehydration melting of the crust at high temperatures, whereas the biotite granites were produced from H₂O-present melting of crustal rocks at lower temperatures. The mineralogical character of the A-type granites is thus a useful indicator of the magmatic conditions under which these rocks were formed.

3.7 CONCLUSIONS

The coeval A-type granites of the Cobequid Highlands, although emplaced under the same regional tectonic regime but not resulting from different sources, present the mineralogical variations that have been observed in the majority of A-type granites elsewhere. Both the biotite and calcic granites were derived by 20-40% partial melting of Neoproterozoic intermediate crustal rocks, whereas the sodic granites show characteristics

of extreme (90 %) fractionation from a mafic parent melt that resembles the coeval Fountain Lake basalts.

Asthenospheric mantle upwelling and large-scale tholeiitic magmatism in the late Devonian underplated the lower crust beneath the Magdalen Basin and presumably the eastern Cobequid Highlands. Repeated injections of mafic melts into the lower crust created, thus, a deep crustal hot zone. In this zone, the emplacement of mafic sills triggered partial melting of the surrounding hydrated crustal rocks which would, in turn, released water upon heating (dehydration melting), further inducing partial melting. Differences in crustal lithology between the eastern and western part of the Cobequid Highlands resulted in the production of variably hydrous granitic melts with systematic differences in oxygen isotopic signatures. Such melts ascended and were emplaced at shallow crustal levels, producing the subsolvus biotite and calcic granites. The extreme fractionates of anhydrous mafic sills would eventually produce the hypersolvus sodic granites. The late Paleozoic plutons of the area were thus constructed by several distinct A-type granitic magmatic batches, derived by specific geological processes such as extreme fractionation and partial melting of crustal rocks and magmatic conditions, such as temperature, pressure and water-in-melt, that resulted in systematic differences in mineralogy.

CHAPTER 4 VARIATION OF THE REE-HYDROTHERMAL CIRCULATION IN COMPLEX SHEAR ZONES: THE COBEQUID HIGHLANDS, NOVA SCOTIA

PREFACE

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4.1 ABSTRACT

The Cobequid Highlands in Nova Scotia comprise a series of 365–360 Ma, REEenriched A-type granite plutons, emplaced along an active shear zone. The purpose of the paper is to identify the sequence of REE-minerals produced during long-lived, regional, hydrothermal circulation and its evolution through time. The fractures hosted in these granites were investigated under the petrographic and scanning electron microscopes and microprobe. Epidote-rich, biotite-rich, chlorite-rich and Fe-oxide-rich fracture fillings have been observed. Mineral analyses revealed a variety of REEminerals in the fractures, including hydroxylbastnäsite-(Ce), parisite-(Ce), synchysite-(Ce), cerianite, thorite, Nb-minerals, hingganite-(Y) and chernovite-(Y). Cross-cutting relationships and mineral associations indicate that the REE-minerals have precipitated at different times and are associated with specific styles of hydrothermal alteration. Epidote was one of the earliest minerals to precipitate in the fractures, associated only with minor precipitation of thorite and hydroxylbastnäsite-(Ce) and removal of As from sedimentary rocks, under low temperatures and reducing conditions. Biotite and chlorite precipitated shortly after epidote under more oxidizing conditions during which comparatively larger amounts of hydroxylbastnäsite-(Ce), thorite, hingganite-(Y) and cerianite formed. The

Fe–oxide–filled fractures formed in the latest stages of hydrothermal circulation and are associated with the largest variety and highest amounts of hydrothermal REE–minerals, such as parisite–(Ce), synchysite–(Y), thorite, Nb–minerals and chernovite–(Y). The oxidizing conditions led to the precipitation of As and the formation of chernovite–(Y) after the alteration of hingganite–(Y). Thus the greatest remobilization and concentration of REE–minerals took place during a late, high–temperature, oxidizing hydrothermal circulation, apparently driven by minor late mafic intrusions.

4.2 INTRODUCTION

In igneous systems, the rare earth elements (REE) are concentrated as incompatible elements in the late stages of magmatic evolution. The majority of economically valuable REE–deposits are associated with either carbonatites or alkaline silicic intrusions (Chakhmouradian and Wall, 2012). The high amounts of REE hosted in such rocks cannot always be explained only by fractionation processes, suggesting their late–stage remobilization by fluids (Williams–Jones et al., 2012). The hydrothermal REE–minerals tend to be more variable and complex than those of magmatic origin due to variable conditions during hydrothermal circulation (Jiang, 2006). Chemistry of the fluid, pressure, temperature, the REE source, and the nature of the country rock are some of the factors that control the hydrothermal transportation and precipitation of REE (Smith et al., 2000).

Many hydrothermal REE deposits are spatially related to alkaline rocks, particularly A–type granites that are interpreted as the primary source of REE (Förster et al., 2011). Such rocks are usually enriched in REE and host magmatic REE–minerals (Whalen et al., 1989; Uher et al., 2009), whose fluid–induced breakdown can result in a

number of hydrothermal REE–phases (Meintzer and Mitchell, 1988; Jiang, 2006). The presence of F, Cl, S and CO₃ ions in the fluids promote the hydrothermal mobility of REE (Rolland et al., 2003; Uher et al., 2009). Based on their charge/radius ratio, the REE tend to form stable complexes principally with F at ambient temperatures, whose stability increases systematically from La to Lu (Williams–Jones et al., 2012). Hydrothermal mobilization of REE is common in areas with iron oxide–copper–gold (IOCG) mineralization (Kontak et al., 2008).

Latest Devonian post-collisional granite plutons, with varying amounts of gabbro, were emplaced along the active Cobequid Shear Zone in the Cobequid Highlands of the Canadian Appalachians (Pe–Piper and Piper, 2003). On the basis of their high alkalies, high concentrations of large ion lithophile elements (LILE) and Ga/Al ratio, these granites show A-type affinities, with within-plate chemical characteristics (Pe-Piper et al., 1991). These plutons, and particularly the mafic rocks, are thought to have provided the heat and possibly fluids for regional-scale hydrothermal activity that produced widespread IOCG mineralization (Pe-Piper et al., 2004; MacHattie and O'Reilly, 2008). The granitic rocks of these plutons are REE–enriched, as is commonly observed in A– type granites, and significant amounts of magmatic allanite–(Ce) and chevkinite–(Ce) are found in the largest pluton, the Wentworth plutonic complex, in the eastern part of the area (Papoutsa and Pe–Piper, 2013). Remobilization of REE has been inferred from the presence of hydrothermal minerals replacing primary REE-minerals in these rocks. There has been no previous attempt, however, to define the nature and the extent of the REEenrichment, either primary or hydrothermal, in the rest of the late Paleozoic plutons in the Cobequid Highlands. Whole–rock geochemical data reported in Pe–Piper (1991, 1995a,

1995b, 1998), and Pe–Piper and Piper (1998a) show that all A–type granites have REE concentrations comparable to those in the Wentworth pluton, with an average range between 100 and 400 ppm total REE. Preliminary petrographic examination of these rocks revealed that the Pleasant Hills, North River, and Cape Chignecto plutons have considerably lower amounts of magmatic REE–minerals than the Wentworth pluton. Only minor allanite–(Ce) is found in these plutons, and yet they contain similar bulk REE concentrations to the Wentworth granites. The motivation for this study came from the petrographic investigation of mm–scale, fractured granitic samples from the North River pluton, which contain more than 300 ppm of total REE but completely lack magmatic REE–minerals, suggesting possible hydrothermal enrichment. As the style of hydrothermal alteration varies along the Cobequid Highlands (Pe–Piper et al., 2004), this area provides a unique opportunity to assess the nature and controls on REE–hydrothermal enrichment, where present, along more than 100 km of the shear zone.

The purpose of this study, therefore, is to document the extent, mineral composition and relative age of post–magmatic REE–circulation in the late Paleozoic plutons in the Cobequid Highlands, and to correlate the changes in the abundance and type of REE–mineral assemblages with specific styles of hydrothermal alteration. Such a study will identify the regional sequence of hydrothermal processes associated with REE–mobility, and also provide information on the behavior of REE under different hydrothermal conditions.

4.3 **GEOLOGICAL SETTING**

The Cobequid Highlands in mainland Nova Scotia (Fig. 4.1A) consist of a basement of Neoproterozoic juvenile Avalonian crust (Pe–Piper and Piper, 2003). In the

late Devonian, the closure of the Rheic Ocean at promontories led to dextral strike slip faulting (Hibbard and Waldron, 2009). This resulted in a distributed extension of the Magdalen Basin as a stepover zone. Crustal thinning continued from latest Devonian until at least the early Carboniferous, accompanied by underplating of a thick gabbroic layer beneath the Gulf of St Lawrence (Marillier and Verhoef, 1989; Marillier and Reid, 1990). At the same time in the Cobequid Highlands, granite plutons with lesser amounts of gabbro were intruded along the faults of the Cobequid Shear Zone, which formed the southeastern margin of the extensional basin (Pe–Piper et al., 1998). Thick rhyolite pyroclastic rocks and voluminous flood basalts of the Fountain Lake Group were the volcanic equivalents of the plutons (Dunning et al., 2002).

The Cobequid Shear Zone consists of a series of subparallel dextral strike–slip faults, comprising the upper crustal expression of the Minas Geofracture between the Avalon and the Meguma terranes (Keppie, 1982; Koukouvelas et al., 2002; Pe–Piper and Piper, 2003). Major fault strands of the shear zone include the Kirkhill fault in the west, the Rockland Brook fault in the east, and the younger Cobequid fault that encompasses the entire area of the Cobequid Highlands (Fig. 4.1B). The Kirkhill and Rockland Brook faults were probably reactivated Neoproterozoic faults that acted as the pathways for magma during pluton emplacement between 365±4 Ma and 358±4 Ma (Murphy et al., 2011).

The faults of the Cobequid Shear Zone experienced ductile dextral strike–slip motion until at least 340 Ma (Visean) (Gibbons et al., 1996; Pe–Piper et al., 2004) with the intrusion of mafic dykes in the Cape Chignecto, West Moose River, and North River plutons during transtension (Piper et al., 1993). Later magmatism consists of <334 Ma



Figure 4. 1 (A) Map showing Appalachian terranes and Carboniferous faults in Atlantic Canada; (B) simplified geological map of the Cobequid Highlands in mainland Nova Scotia. lamprophyre dykes in the central Cobequid Highlands and a series of alkaline mafic stocks near the Cobequid fault southeast of the Cobequid Highlands, which are dated stratigraphically between 340 and 310 Ma (MacHattie and O'Reilly, 2008). The evolution of the regional fault system was accompanied by the development of the composite Maritimes basin (Piper et al., 1993; Murphy et al., 2011). In the Cobequid Highlands the major stratified units are represented by the latest Devonian–early Carboniferous clastic rocks of the Horton Group and the volcanic rocks of the Fountain Lake Group (Pe-Piper and Piper, 2003).

Several episodes of post-magmatic deformation are recorded in the Cobequid Highlands as indicated by the solid-state deformation recorded in the late Paleozoic

plutons (Waldron et al., 1989; Piper et al., 1993; Miller et al., 1995; Pe-Piper and Piper, 2003). The first phase of post-magmatic deformation that followed pluton emplacement is characterized by NW-vergent thrusting of late Tournaisian age, as indicated along the northern margin of the Wentworth pluton (Koukouvelas et al., 1996; Piper et al., 1996). North of the Cobequid fault, the sedimentary rocks of the Horton Group appear strongly deformed and cut by late granitic dykes which are no older than 355 Ma, (Pe-Piper et al., 1989; Pe-Piper and Piper, 2003) and therefore the deformation of the Horton Group predates Visean (Murphy et al., 2011). In the Cape Chignecto pluton, similar late granitic dykes were interpreted to occupy thrust planes (Waldron et al., 1989) and may thus indicate NW-vergent thrusting of similar age to that observed in the area of the Wentworth pluton (Piper et al., 1993). Close to the Serpukhovian–Bashkirian boundary, another phase of deformation is recorded along the northern and southern margins of the Cobequid Highlands. To the south, the Clarke Head megabreccia lies 4 km north of the Cobequid fault and contains mafic granulites whose metamorphism is dated at 370 ± 2 Ma (Gibbons et al., 1996). The Clarke Head area reflects the episodic nature of deformation along the faults of the Cobequid Shear Zone. The early stage of deformation that postdates metamorphism occurred at 338-346 ±3 Ma (corrected Ar-Ar age on hydrothermal amphibole in veins that cut the mylonitic fabric) (Gibbons et al., 1996). By the mid-Carboniferous (~320 Ma- Serpukhovian) a new E-W fault system developed as a result of continued Alleghanian convergence of Gondwana with Laurentia (Murphy et al., 2011). This fault system comprises the Cobequid–Chedabucto Fault Zone, which in the western part of the Cobequid Highlands reactivated the older Cobequid fault as a cataclastic dextral strike-slip fault, but in the east the newer part of the fault cuts lower

Carboniferous sedimentary rocks (Pe–Piper and Piper, 2003). Megabrecciation in Clarke Head is interpreted to have occurred in the late Carboniferous (ca. 315-310 Ma), since the megabrecciated rocks contain dispersed clasts of evaporites interpreted as Windsor Group (Gibbons et al., 1996) and may therefore be related to the Alleghanian deformation.

The Wentworth pluton (Fig. 4.1B), north of the Rockland Brook fault, has the most complex distribution of igneous rock types, mineralogy, and geochemistry of all the latest Devonian intrusions in the Cobequid Highlands. In the southwestern part of the pluton lies the largest gabbroic body in the area, whereas the northeastern part consists principally of biotite-amphibole granite (Koukouvelas et al., 2002). The Pleasant Hills pluton (Fig. 4.1B) lies north of the Cobequid and Rockland Brook faults in the eastern Cobequid Highlands. This composite pluton consists of gabbro, granodiorite, and hornblende–bearing granite of several generations (Pe–Piper et al., 1998). The North River pluton (Fig. 4.1B), in the central Cobequid Highlands, is a simple intrusion located in the stepover zone between the Rockland Brook and Kirkhill faults. It consists mainly of a homogeneous coarse–grained biotite–granite with minor gabbroic bodies near the margins and mafic dykes that cut the granite (Pe–Piper, 1991). The large Cape Chignecto pluton (Fig. 4.1B) is located immediately north of the Cobequid Fault in the western Cobequid Highlands, and consists of biotite-bearing A-type granite, diorite, and hybrid rocks that are interpreted to be products of mixing between granitic and mafic magmas (Pe–Piper, 1995b). In the Wentworth pluton Koukouvelas and Pe-Piper (1996) concluded that magma emplacement took place under shearing and E-W extension, as indicated by the shapes of deformed mafic enclaves in some of the granites. Magma emplacement during dextral shearing and NW-thrusting is inferred for the Pleasant Hills, North River

and Cape Chignecto plutons from field observations (Koukouvelas et al., 1996). The plutons have rim zones along their northern margin for which a sinistral sense of movement is interpreted, while dextral shearing more prominent of the southern margins of the plutons and closer to the main faults (Pe-Piper et al., 1998).

The latest Devonian granites of the Cobequid Highlands are metaluminous to peralkaline alkali–feldspar A–type granites (Koukouvelas et al., 2002). They are fine- to medium-grained rocks with granular texture that can be granophyric in places. The Wentworth pluton is the most alkaline, and is the only pluton that contains primary sodic amphibole (Pe–Piper, 2007). Calcic amphibole of ferro–edenitic composition is present in both the Wentworth and Pleasant Hills plutons (Pe–Piper et al., 1998; Pe–Piper, 2007), whereas the other granite plutons are biotite–bearing (Pe–Piper, 1991; 1995b). The Wentworth pluton granites contain many REE–minerals, including magmatic minerals like allanite–(Ce) and chevkinite–(Ce) as parts of the granite assemblage, late magmatic hingganite–(Y), and post–magmatic samarskite–(Y), fersmite, aeschynite–(Y) and hydroxylbastnäsite–(Ce) (Papoutsa and Pe–Piper, 2013). Even though the Wentworth granites host a variety of hydrothermal REE–minerals, most are dispersed within the body of the rock, rather than in fractures.

The granites of the late Paleozoic plutons in the Cobequid Highlands are commonly fractured, and the density of these fractures increases towards the major faults of the Cobequid Shear Zone (Nearing, 1991). The Cape Chignecto pluton is mylonitic, however mylonitic fabrics are absent in the North River pluton (Waldron et al., 1989; Pe–Piper, 1991). Hydrothermal alteration occured in places throughout most of the Carboniferous in the Cobequid Highlands. Sodic alteration is expressed by the formation of

hydrothermal albite and riebeckite, and was focused in the western part of the Cobequid Highlands along the Cobequid and Kirkhill faults (Pe–Piper et al., 2004), dated at 353±4 Ma in the West Moose River pluton (corrected Ar–Ar age on hydrothermal riebeckite, Table 4.1).

Description	Mineral	Method	Age (Ma)*	Reference
Albitized granite, West Moose River pluton	hydrothermal riebeckite	Ar-Ar	353±4	Pe-Piper et al. (2004)
Deformed diorite, Cape Chignecto pluton, metamorphic biotite	biotized hornblende	Ar-Ar	347±3	Pe-Piper et al. (2004)
Altered diorite, southwestern Cape Chignecto pluton	low-Ti biotite	Ar-Ar	343±3	Pe-Piper et al. (2004)
Gneissose granodiorite south of the Rockland Brook fault	mixed hornblende- biotite separates	Ar-Ar	336-343 ±4	Pe-Piper et al. (2004)
Medium-grained diorite, Cape Chignecto pluton	hydrothermal biotite	Ar-Ar	342±3	Pe-Piper et al. (2004)
Lamprophyre dyke in the central Cobequid Highlands	whole-rock	Ar-Ar	334±3	Pe-Piper et al. (2004)
Lamprophyre dyke in the central Cobequid Highlands	green biotite	Ar-Ar	326±2	Pe-Piper et al. (2004)
Carbonate-suphide vein in Copper Lake deposit	pyrite	Re-Os	323±8	Kontak et al. (2008)
Altered siltstone adjacent to a carbonate-sulphide vein with intergrowths of hydrothermal muscovite and carbonate, Copper Lake deposit	hydrothermal muscovite	Ar-Ar	309.5 ±1.6	Kontak et al. (2008)

Table 4. 1 Geochronological data from various styles of hydrothermal alteration.

Notes: *: All Ar-Ar ages published prior 2008 have been corrected from the original data according to Kuiper et al. (2008), due to new intercalibrations in Ar-Ar dating.

Among the plutons involved in this study, albitization of the granites is reported from only a few localities in the Cape Chignecto pluton (Nearing, 1991). In those cases, the association of albitization with ductile deformation fabrics is interpreted as evidence for an early, high-temperature hydrothermal event that was localized and took place in deep structural levels. Epidotization is observed along spaced fracture sets in the volcanic rocks of the Fountain Lake Group in the central and western parts of the Cobequid Highlands (Waldron et al., 1989; Nearing, 1991). In the Wentworth pluton, plastically folded epidote–rich dykes have been reported to cut ductile deformed granites (Koukouvelas et al., 2002).

Widespread biotization has been observed in the mafic rocks of the Cape Chignecto pluton and is dated at 343 ± 3 Ma (corrected Ar–Ar age on hydrothermal biotite), while hydrothermal green biotite in the central Cobequid Highlands yielded an age of 334 ± 2 Ma (corrected Ar–Ar age, Pe–Piper et al., 2004, Table 4.1), suggesting that potassic alteration in the Cobequid Highlands was long-lived. Biotization in the granites is macroscopically expressed as mafic stringers and clots up to 3 mm in width mostly containing secondary biotite, magnetite, and hematite (Nearing, 1991). In the most deformed rocks of the Cape Chignecto pluton, secondary biotite and magnetite are concentrated along brittle fractures at high angles to the foliation which generally has a NNE trend (Waldron et al., 1989). Brittle deformation in the North River pluton is more pronounced in the southern part of the intrusion closer to the Cobequid fault (Pe–Piper, 1995a). In the central part of the pluton, north-trending dykes, representing the last major structures, are dextrally offset by late fracture sets with a W–E shear trend, parallel to the Cobequid fault (Piper et al., 1993). There is limited information on the fracture systems in the eastern part of the Cobequid Shear Zone. Miller et al. (1995), however, reported that the majority of the fractures in the deformed granites of the Pleasant Hills and Wentworth plutons are parallel to the trace of the Rockland Brook fault and only secondary fractures occur as conjugate sets.

Widespread mineralization along the late Carboniferous Cobequid-Chedabucto Fault Zone occurs as veins or areas of intense alteration and includes barite, siderite, pyrite, ankerite and manganese deposits in brecciated rocks (Murphy et al., 2011). The most significant deposits are the Londonderry and Mt. Thom iron carbonate deposits, Copper Lake Cu–Au deposit, and the Brookfield barite deposit, all interpreted as IOCG mineralization (MacHattie and O'Reilly, 2008). The mineralization in many areas is demonstrably post–Visean (< 330 Ma). Mineralization has been dated directly at 323 Ma (Re–Os age on hydrothermal pyrite, Table 4.1) in the sulfide deposits of Copper Lake (Kontak et al., 2008), 100 km east of the Cobequid Highlands. The presence of monazite, thorite and xenotime in the fractures of Copper Lake indicates that REE-circulation accompanied this late hydrothermal event (Kontak et al., 2008). The presence of albitite dykes within the young 340–310 Ma gabbro stocks east of the Cobequid Highlands implies that a late stage, volatile-rich phase differentiated from the gabbro and may have played a role in generating the iron-rich hydrothermal fluids of the ~323 Ma IOCG mineralizing event in the Cobequid–Chedabucto fault zone (MacHattie and O'Reilly, 2008).

4.4 METHODS

The samples of this study were collected from the Wentworth, Pleasant Hills, North River and Cape Chignecto plutons using the maps of Pe–Piper and Piper (2005). Thin and polished thin sections were already available for these granites, and whole–rock analyses were previously done for most samples. For the purpose of this work, additional analyses of trace elements were done at the Activation Laboratories, using their code 4B1 for Lithoresearch. The concentrations of trace elements were determined by

metaborate/tetraborate fusion ICP-MS. For the investigation of the hydrothermal REEminerals, the samples were selected on the basis of high fracture density (>5% measured surface of fractures on thin section) and with microscopic evidence of alteration. Out of a total of 150 thin and polished thin sections from all plutons, 11 strongly fractured samples were eventually selected. These include 5 samples from the Wentworth pluton, 2 samples from the Pleasant Hills pluton, 2 samples from the North River and 2 samples from the Cape Chignecto pluton. The fracture fillings were first observed under a Nikon Eclipse E400 POL petrographic microscope, with a PixelINK PL-A686C camera, and then analyzed by electron dispersion spectroscopy for a qualitative mineral identification (EDS). A LEO 1450 VP scanning electron microscope (SEM) was used with a maximum resolution up to 3.5 nm at 30 kV and a detection limit > 0.1%, at the Regional Electron Microscopy Centre at Saint Mary's University. The operating conditions were at 15kV of accelerating voltage with a 25nA beam current and a beam diameter of 1 micron. Selected grains of all minerals identified with EDS were further analyzed for a more quantitative chemical description. All wavelength dispersive spectroscopy (WDS) mineral analyses were made at the Robert MacKay Electron Microprobe Laboratory at Dalhousie University, using a JEOL–8200 electron microprobe with five wavelength spectrometers and a Noran 133 eV energy dispersion detector. The operating conditions were at 15kV of accelerating voltage with a 20nA beam current, and a beam diameter of 1 micron. The standards, peaks and crystals used for the analyses of REE-minerals are: monazite (P, Ka, PETJ), columbite (Nb, La, PETJ), tantalite (Ta, La, LIFH), ThO₂ MAX (Th, M α , PETJ), UO₂ (U, M β , PETJ), zirconia (Zr and Hf, L α , PETJ and LIFH respectively), Y-Al-garnet (Y, La, TAP), LaPO₄ MAX (La, La, LIF),

CePO₄_MAX (Ce, L α , LIF). Drake and Weill (1972) REE glasses were used for: Pr (L β , LIFH), Nd (L β , LIFH), Sm (L β , LIFH), Eu (L α , LIFH), Gd (L β , LIFH), Dy (L α , LIF), Ho (L α , LIFH), Er (L α , LIF), and Yb (L α , LIF). Fluorine was calibrated using F-apatite (K α , TAPH). Certain pairs of elements (Ti-Hf, Zr-P, Hf-Ho, Er-Nb, and F-Ce), created peak overlaps. For those elements the standards were carefully selected as to not contain the overlapping element, and correction factors were calculated from the measured intensities of the peaks, in order to determine the real concentrations of the elements present.

In order to correlate the presence of specific hydrothermal REE-minerals with different styles of hydrothermal alteration, the modal and bulk compositions of the fracture fillings and the total amount of fractures present in selected samples were estimated by point-counting on thin sections. This was carried out on an automated stepping stage of Conwy Valley Systems and PETROG software for Windows. The polished thin sections were then scanned for a better relocation of the fractures of interest under the SEM. The EDS analyses and the backscattered electron (BSE) images were also used to estimate the modal composition of the fracture fillings. The area of each phase with a different brightness in BSE images was measured using ImageJ software for Windows. Each phase was identified from the EDS analyses and the final modal compositions were calculated in Excel. Bulk chemical compositions of the fracture fillings were estimated from the calculated modal compositions and the mineral analyses.

The temperature of the hydrothermal fluids was estimated by the chlorite thermometer of Kranidiotis and Maclean (1987), where applicable. The chemistry of chlorite can be affected by temperature and leads to an increase of Al^{iv} (Cathelineau and Nieva, 1985). The chlorite thermometer used in this work includes a correction for Fe–

Mg substitution in chlorites, which can also affect the amount of Al^{iv} present (Kranidiotis and Maclean, 1987). Using the correction of Zang and Fyfe (1995) for Al^{iv} in the equation of Kranidiotis and Maclean (1987), the estimated error is 25°C (Lana et al., 2003). This work follows the nomenclature schemes of Wiéwiora and Weiss (1990), Rieder et al. (1998), and Armbruster et al. (2006) for chlorite, biotite and epidote, respectively.

4.5 **RESULTS**

4.5.1 MAJOR MODAL COMPOSITION OF THE FRACTURE FILLINGS, DISTRIBUTION AND PARAGENETIC SEQUENCE

Epidote–rich fractures. The majority of these fractures are mm–scale and contain up to 94 % epidote, with titanite also being a major component in the fracture filling, up to 44 % (Table 4.2). The epidote approaches the composition of clinozoisite (Fig. 4.2A) with no more than 33% of the "pistacite" component, and locally coexists with andradite (Table 4.2 and Appendix C.1). In fractures where dominant epidote coexists with magnetite, the latter comprises less than 20% of the modal composition of the fracture filling. In some fractures chlorite appears in varying amounts (up to 41%) as mixture with fine–grained epidote or calcite. Pyrite (Figs. 4.3a and b) has been found in a few epidote– titanite fractures.

Biotite–rich fractures. Secondary biotite of annite composition (Fig. 4.2B) is abundant in these fractures (up to 96%, Table 4.2, Appendix C.2). The biotite is commonly mixed with varying amounts of epidote (Figs. 4.3C). Magnetite (less than 40%) is also present.

Chlorite–rich fractures. The main component of these fractures is chlorite with a composition ranging from ripidolite to chamosite (Figs. 4.2C and D) and magnetite (Fig.

4.3D, Table 4.2, Appendix C.3). Titanite, ilmenite, TiO_2 mineral, epidote and zircon are also present in these fractures (< 2 mm width). From chemical analyses alone it was not possible to accurately determine which TiO_2 polymorph (rutile, anatase or brookite) is present in the fracture fillings, and therefore, in this work this composition will be



Figure 4. 2 Chemical discrimination diagrams for the major hydrothermal minerals found in the granites of the Cobequid Highlands. Variations in the chemistry of hydrothermal epidote are shown in plot (A), biotites are classified in plot (B), and plot (C) is a discrimination diagram of Fe-rich chlorites (after Wiewióra and Weiss, 1990). Plot (D) shows variations in Fe-number (Fe#) for the hydrothermal chlorites from all plutons.

referred to as TiO₂ mineral. Epidote and titanite occur as fine-grained patches in rare

fractures, whereas hydrothermal zircon occurs as subhedral to anhedral grains associated

with chlorite.

Fe–rich fractures. These fractures are characterized by dominant magnetite (up to 88 %) coexisting with lesser amounts of epidote, biotite and chlorite. Pyrite and barite inclusions rarely occur in magnetite–rich fractures (Fig. 4.3E).

Calcite–rich fractures. This is the least common type of fracture–filling in the studied samples. This type comprises high amounts of calcite coexisting with lesser chlorite and titanite, while epidote is absent (Table 4.2). In the Wentworth granites, the main type of fracture, in terms of abundance and size, observed in the samples is epidote–rich. In a total of 102 studied fractures from the Wentworth granites, 60% are epidote–rich, 28% are calcite–rich, 10% are magnetite–rich and 2% are titanite–rich.

In rare cases, magnetite–rich fractures cross–cut epidote–filled fractures at a high angle, forming conjugate sets (Fig. 4.3B), and the former appear more continuous, suggesting a probable younger age. Irregular patches of titanite, andradite and chlorite in the epidote suggest that these minerals precipitated at the same time from the fluids in the epidote–filled fractures. Pyrite inclusions in magnetite indicate pyrite probably formed first.

In the Pleasant Hills pluton, the fractures show a larger variation in the distribution of hydrothermal minerals than in the Wentworth pluton and epidote occurs in lower amounts (Table 4.2). The texturally older fractures comprise biotite mixed with epidote, cut by younger magnetite–bearing biotite fractures that contain no epidote, at high angles as conjugate sets (Fig. 4.3C). These younger fractures offset the old ones, and have the highest concentration of biotite close to the intersection with the older biotite fractures. Conversely, the amount of magnetite increases away from the intersection point, until the fractures are filled by almost monomineralic magnetite. There is no clear relationship

								Mo	dal comp	ositio	(%) I						REE minerals
Pluton	Sample (No of studied fractures)	% fractures	Type of fracture	andradite	barite	biotite	ealcite	chlorite	ətobiqə	ətinəmli	ətitəngam	pyrite	вее	TiO ₂ mineral	ətinstit	noəriz	
	30-9-1 (18)	11	ep-tnt (77%) tnt-rich (15%) mag-rich (8%)					10-20 20 12	30-50 30 20		5-8 43	0	4	5)-30 44 23	0-4 2	thorite
	6484 (13)	12	ep-tnt (43%) ep-rich (50%) mag-rich (7%)		-				47-60 60-94 20		16-36 2-20 55	0-12 <1 11		8 10	-20 -14 10		
υτωοιτμ	6497 (24)	17	cal-rich (95%) chl-rich (5%)	0-5.5			30-85 15	14-30 41			1-23 7		0-5 <1		-26 37	$\overline{\vee}$	hydroxylbastnäsite-(Ce), thorite hydroxylbastnäsite-(Ce), hingganite-(Y)
ıэW	6591 (23)	16	ep-tnt (40%) ep-rich (30%) mag-rich (26%) tnt-rich (4%)						36-50 52-82 20-22 26		16-35 7-16 50-58 15	~ ~	0.5	11 8 11 12 1	-20 -10 -26 40	$\begin{array}{ccc} 0-6\\ -& -& -\\ 0+& -& -\\ \end{array}$	hydroxylbastnäsite-(Ce) hydroxylbastnäsite-(Ce), thorite
-	6686 (24)	12	ep-tnt (38%) ep-rich (38%) mag-rich (24%)	0-46 17 0-17					30-60 60		[4-27 0-9 43-88	V	0.5	$\begin{array}{c} 15\\0\\8\end{array}$	-35 -15 -33	0.4	hydroxylbastnäsite-(Ce)
elliH ture	4106 (32)	40	ep-bt (48%) bt-rich (26%) ep-rich (26%)			30-50 55-85 18-27			37-60 60-75		0-40	V C	.0.5)-2 <	× 7	-20		hydroxylbastnäsite-(Ce) hydroxylbastnäsite-(Ce), cerianite thorite, hydroxylbastnäsite- (Ce)
Pleas	4672 (31)	∞	bt-chl (41%) bt-rich (59%)		90	40-50 3-96		18-30 3-30			0-32)-2				hydroxylbastnäsite-(Ce), Nb-minerals

Table 4. 2 Petrographic description of the studied samples.

								Mod	al com	positio	(%) u					REE minerals
Pluton	Sample (No of studied fractures)	~ fractures	Type of fracture	andradite	barite	biotite	ealeite	chlorite	ətobiqə	ətinəmli	ətitənyam	pyrite	вее	TiO2 mineral titanite	nooriz	
		c	chl-rich (38%)					70-92			6-24		1-3			synchysite-(Ce), hydroxylbastnäsite-(Ce)
	1 / /9 (34)	ע	mag-rich (52%)					5-11		7-12	60-88	C)-5		$\overline{\vee}$	synchysite-(Ce), parisite- (Ce), hydroxylbastnäsite- (Ce), thorite
J			ep-rich (10%)					6	0-85		6-30	•)-2		\leq	hydroxylbastnäsite-(Ce)
əvi X			chl-rich (18%)					68-90			6-30	2	-12			hingganite-(Y), chernovite- (Y)
North	4824 (39)	L	chl-mag-rich (38%)					20-50		0-39	31-50	1	-25		0-5	hingganite-(Y), chernovite- (Y), parisite-(Ce), Nb- minerals
			mag-rich (44%)					6-13		0-21	58-84	0	-18		0-5	thorite. synchysite-(Ce), parisite-(Ce), hydroxylbastnäsite-(Ce), chernovite-(Y)
013	2039 (27)	28	bt-rich (79%)			60-91				2-11	2-30	-	9 - 0		$\overline{\nabla}$	hydroxylbastnäsite-(Ce), parisite-(Ce), thorite
oəuā			chl-ilm (16%)					37-48		4-40	5-20		4		0-2	hydroxylbastnäsite-(ce)
₿id(mag-rich (5%)			30					45	V	0.5 2	0	4	hydroxylbastnäsite-(ce)
) 9((22) 84 (22)	40	bt-rich (18%)			62-90		-)-20				0-2	0-4	0.5	chernovite-(Y), thorite
qeJ		F	ep-rich (11%) chl-rich (71%)			0-24		0-6 6 60-92	0-92	9-0		0	-18 0	ŝ	0 1−4	cerianite. chernovite-(Y)
												>	> +	,	- >	

between hydrothermal biotite and chlorite, but only the biotite fractures contain epidote. In the chlorite fractures of the Pleasant Hills pluton, magnetite appears to either rim corroded chlorite, or fill empty spaces in fractured and deformed chlorite veins, and therefore appears to post–date them (Fig. 4.3F). In rare cases, magnetite grains show interlocking textures with chlorite, suggesting that the formation of those two minerals was at one time contemporaneous.

In the North River pluton, the main variation in the modal composition of the fracture fillings is in the distribution of chlorite and magnetite. From the 73 studied fractures of this pluton, 61% are magnetite–rich, 33% are chlorite–rich, 4% are epidote–rich and only 1% are biotite–rich (Table 4.2).

Magnetite may occur as grains intergrown with chlorite suggesting co–precipitation from the fluids. However, some magnetite fills pores or penetrates corroded chlorite, indicating a later formation. Rare chlorite–rich fractures appear to have been deformed and fractured. As in the other plutons, some secondary magnetite–rich fractures that contain chernovite–(Y) appear to form conjugate sets with pre–existing chlorite–rich fractures.

The major hydrothermal minerals common in the mm–scale fractures of Cape Chignecto pluton include chlorite, biotite, magnetite and smaller amounts of epidote, ilmenite, TiO₂ mineral, zircon, and titanite (Table 4.2 and 4.3). The fractures of the Cape Chignecto granites do not show clear textural relationships between biotite and chlorite. Both minerals appear corroded and are replaced by magnetite associated with TiO₂ mineral, rare pyrochroite (Table 4.3) and REE–carbonate minerals. Some chlorite is associated with chernovite–(Y), which appears to be a late mineral. Locally TiO₂



Figure 4. 3 Back–scattered electron (BSE) images of selected fractures from the granites of the Cobequid Highlands. Images (A) and (B) show offset epidote–rich fractures with presence of pyrite from the Wentworth pluton. Image (C) shows a pre–existing biotite–epidote fracture that is offset by a younger biotite–magnetite fracture in Pleasant Hills pluton. Image (D) shows a chlorite–filled fracture that is irregularly cut by a later magnetite veinlet in North River pluton. Image (E) shows grains of barite found in a rare magnetite vein in the Wentworth granites. Image (F) shows late magnetite filling a dismembered chlorite fracture in the Pleasant Hills pluton. Mineral abbreviations: Ep: epidote, Py: Pyrite, Brt: Barite, Bt: Biotite, Mag: Magnetite, Chl: Chlorite.

mineral-bearing chlorite fractures are cut by younger, almost barren veinlets that only in

places are filled with magnetite. Some chlorite shows ilmenite forming along the
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Pluton	WP	WP	WP	PHP	ЧНР	NRP	WP	CCP	CCP	CCP	ΡΗΡ	NRP	NRP	WP	ЧНР	CCP	NRP	NRP	CCP
Sample	6497	6497	6498	4106	4672	1779	6497	6284	6284	6284	4106	4829	1779	6497	4106	6284	4824	1779	6284
Mineral	eb	and	tnt	bt	ilm	chl	cal	Ti-min	pyr	aln	H-bas	par	syn	thr	cer	cher	hing-Y	zrn	cher-thr
P_2O_5	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.37	n.d	0.07	0.08	2.04	2.64	0.06	0.12	3.03
Nb ₂ O ₅	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.07	b.d.	0.05	0.03	b.d.	0.04	b.d.	n.d	b.d.	b.d.
Ta_2O_5	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.01	b.d.	n.d	n.d	b.d.	n.d	b.d.	n.d	b.d.	b.d.
As205	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.06	n.d	n.d	0.01	3.36	33.73	n.d	0.02	25.83
SiO ₂	38.86	34.39	31.50	37.98	0.11	26.63	1.77	2.49	0.56	30.26	3.13	1.23	0.61	22.26	13.24	3.02	29.43	32.11	3.02
TiO ₂	0.10	1.20	34.78	2.05	49.25	b.d.	0.03	88.97	0.13	1.39	b.d.	n.d	n.d	0.56	0.03	b.d.	0.06	b.d.	0.14
ZrO ₂	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.07	n.d	n.d	1.25	n.d	0.07	0.01	59.67	0.21
HfO_2	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.07	n.d	n.d	b.d.	n.d	b.d.	n.d	2.50	b.d.
ThO ₂	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.04	3.11	0.33	0.16	42.14	0.69	2.00	0.25	0.38	23.52
\mathbf{UO}_2	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.23	0.04	n.d	1.46	0.03	0.30	0.01	0.65	0.76
Al_2O_3	24.61	3.26	3.47	19.28	0.02	16.65	0.15	0.45	0.97	11.46	0.65	0.47	0.23	4.82	2.00	b.d.	0.18	0.03	0.04
Y_2O_3	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.30	1.31	2.06	4.92	0.93	8.74	30.91	27.84	1.64	19.02
La2O3	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	6.51	15.40	10.55	8.63	1.29	0.10	0.06	n.d	b.d.	0.07
Ce2O3	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	12.78	25.11	18.15	19.99	3.57	47.90	4.55	0.34	0.01	5.67
Pr ₂ O ₃	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	1.16	2.17	2.26	2.01	0.41	0.13	0.18	0.17	b.d.	0.16
Nd ₂ O ₃	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	4.31	9.77	10.05	9.15	1.71	0.68	1.26	0.84	b.d.	1.11
Eu ₂ O ₃	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.31	0.76	1.88	1.88	0.13	0.30	0.06	1.08	0.02	0.09
Sm ₂ O ₃	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.56	1.15	0.82	0.58	0.52	n.d	1.57	0.12	b.d.	1.35
Dy ₂ O ₃	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.31	0.36	1.27	1.79	0.29	0.92	5.49	4.69	0.24	3.31
Gd ₂ O ₃	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.42	0.68	0.58	0.84	0.28	1.85	5.17	5.11	0.03	3.43
Er_2O_3	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.10	0.05	n.d	0.12	0.12	0.44	2.75	2.28	0.41	1.39
$H_{02}O_3$	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.02	0.20	b.d.	0.71	2.44	3.01	0.13	1.56
Yb ₂ O ₃	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	n.d	n.d	0.01	0.42	1.67	1.78	0.48	0.95
FeO	11.55	25.11	0.88	19.66	40.93	31.72	0.89	2.18	0.29	17.94	2.24	9.10	0.91	7.05	4.73	0.96	7.73	0.70	0.38
MgO	0.15	0.69	0.05	7.36	6.76	10.43	0.03	0.00	0.09	0.37	0.01	n.d	0.25	0.20	n.d	b.d.	0.00	0.01	b.d.
MnO	b.d.	0.06	0.02	0.06	0.03	0.54	0.20	0.02	74.11	0.41	b.d.	n.d	n.d	b.d.	1.75	b.d.	0.08	0.07	0.16
CaO	23.77	32.90	29.31	0.03	0.07	b.d.	52.25	2.46	0.13	9.43	4.41	12.20	17.85	6.63	0.24	0.39	5.34	0.22	0.53
Na ₂ O	b.d.	b.d.	0.02	0.10	0.06	0.02	0.03	0.01	0.03	0.00	b.d.	n.d	n.d	0.08	n.d	b.d.	n.d	0.06	b.d.
K_2O	b.d.	b.d.	0.14	9.83	0.06	0.05	0.07	0.03	0.08	0.02	0.22	0.11	0.07	0.06	0.03	0.02	0.03	0.07	0.03
Ч	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.04	3.81	3.14	2.68	0.08	n.d	b.d.	0.14	0.08	b.d.
CO2 calc	ı	ı	ı	ı	ı	ı	22.95	ı	ı		23.08	27.26	29.67	ı	·	·	ı	ı	ı
0- F	'	·	·	ı	ı		·	,	·	-0.02	-1.60	-1.32	-1.13	ı	'	'		·	'
Total	99.04	97.61	100.16	96.35	97.29	86.03	78.36	96.59	76.39	98.12	96.58	100.23	101.52	95.92	90.39	99.22	90.58	99.66	95.76

Distan	U/M	M/N	M/M	arra	unu	UDN	M/M	200		000	nın	adiv	a an	U/M	nın	a	aan	a an	
LIULUI	ML	A F		rnr	THT	NKL	WF	CCF	CCF	CCF	THE	INN	NKF	A L	THE	CCF	NKL	NKF	CCF
Sample	6497	6497	6498	4106	4672	1779	6497	6284	6284	6284	4106	4829	1779	6497	4106	6284	4824	1779	6284
Mineral	eb	and	tnt	bt	ilm	chl	cal	Ti-min	pyr	aln	H-bas	par	uńs	thr	cer	cher	hing-Y	zrn	cher-thr
Norm.	0=12.5	0=12	0=5	0=10, 0H=2	O=3	0=10, 0H=8	C=1	C=1	C=1	C=8	C=1	C=3	C=2	0=4	C=1	0=4	0=10	0=4	0=4
Ь	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.009	0.000	0.003	0.002	0.035	0.082	0.003	0.003	0.061
qN	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.003	0.000	0.002	0.001	0.000	0.000	0.000	0.000	0.000	0.000
Ta	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000
\mathbf{As}	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.001	0.000	0.000	0.000	0.036	0.649	0.000	0.000	0.323
Si	3.041	2.929	1.024	2.842	0.003	2.860	0.030	0.034	0.008	3.003	0.095	0.092	0.030	0.795	0.270	0.222	1.862	0.993	0.144
Ï	0.000	0.077	0.850	0.115	0.935	0.000	0.000	0.929	0.001	0.104	0.000	0.000	0.000	0.015	0.001	0.000	0.003	0.000	0.005
Zr	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.001	0.000	0.000	0.022	0.000	0.003	0.000	0.900	0.005
Hf	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.001	0.000	0.000	0.000	0.000	0.000	0.000	0.022	0.000
Тћ	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.001	0.022	0.006	0.002	0.343	0.003	0.034	0.004	0.003	0.256
n	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.002	0.001	0.000	0.012	0.000	0.005	0.000	0.004	0.008
AI	2.268	0.327	0.066	1.699	0.000	2.107	0.003	0.004	0.009	1.341	0.023	0.041	0.013	0.203	0.048	0.000	0.013	0.001	0.001
Υ	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.016	0.021	0.082	0.130	0.018	0.095	0.606	0.937	0.027	0.242
La	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.238	0.173	0.290	0.158	0.017	0.001	0.001	0.000	0.000	0.001
Ce	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.462	0.280	0.495	0.364	0.047	0.358	0.061	0.008	0.000	0.050
Pr	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.042	0.024	0.061	0.036	0.005	0.001	0.002	0.004	0.000	0.001
Nd	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.153	0.106	0.267	0.162	0.022	0.005	0.017	0.019	0.000	0.010
Sm	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.010	0.012	0.048	0.032	0.006	0.002	0.020	0.024	0.000	0.001
Eu	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.019	0.008	0.021	0.010	0.002	0.000	0.001	0.003	0.000	0.011
Gd	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.010	0.007	0.031	0.029	0.003	0.006	0.063	0.098	0.002	0.026
Dy	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.013	0.004	0.014	0.013	0.003	0.012	0.065	0.104	0.000	0.026
Er	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.003	0.000	0.000	0.002	0.000	0.003	0.029	0.046	0.004	0.011
Ho	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.001	0.003	0.001	0.005	0.032	0.060	0.001	0.012
Υb	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.003	0.019	0.034	0.004	0.007
Fe	0.680	1.608	0.024	1.230	0.864	3.309	0.013	0.025	0.004	1.489	0.057	0.566	0.038	0.211	0.081	0.059	0.409	0.018	0.015
Mg	0.000	0.008	0.002	0.821	0.254	1.670	0.001	0.000	0.002	0.054	0.000	0.000	0.018	0.010	0.000	0.000	0.000	0.000	0.000
Mn	0.000	0.050	0.001	0.004	0.001	0.049	0.003	0.000	0.946	0.034	0.000	0.000	0.000	0.000	0.030	0.000	0.004	0.002	0.007
Ca	1.993	3.002	1.021	0.002	0.002	0.005	0.948	0.037	0.002	1.002	0.144	0.973	0.950	0.254	0.005	0.031	0.362	0.007	0.027
Na	0.000	0.000	0.001	0.015	0.001	0.000	0.001	0.000	0.000	0.000	0.000	0.000	0.000	0.006	0.000	0.000	0.000	0.004	0.000
K	0.000	0.000	0.003	0.938	0.001	0.000	0.002	0.000	0.001	0.003	0.008	0.011	0.004	0.003	0.001	0.001	0.002	0.003	0.001
F	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.006	0.090	0.123	0.105	0.005	0.000	0.000	0.028	0.004	0.000
Total	7.982	8.001	2.992	7.666	2.061	10.000	1.000	1.029	0.973	8.006	1.090	3.123	2.11	2.005	1.000	2.000	4.000	2.000	1.250
Notes: el	p: epidote	, and: and	dradite, tı	nt: titanite	e, bt: biot	ite, ilm: i	lmenite,	chl: chlor	ite, cal: c	alcite, Ti	-min: TiC)2-miner	al, pyr: py	rochroite	e, aln: alli	anite-(Ce	e), H-bas:		
hydroxyl	lbastnäsit	э-(Ce), pí	ar: parisit	te-(Ce), s.	yn: synch	ysite-(Ce), thr: the	orite, cer:	cerianite	, cher: ch	ernovite-	(Y), hing	r-Y: hing	ganite-(Y), zrn: zii	con, che	r-thr: mix	tture of th	norite
and cher.	novite-(Y). CO ₂ w.	as calcul.	ated base	d on char	ge-balanc	ce calcula	ations, Nc	orm. indie	cates the 1	number o	f oxygen	s used fo	the norn	nalizatior	n of the s	tructural	formula,	b,d=
below de	stection li	mit.																	

cleavage that also fills pores in the corroded parts of the same grain, suggesting a later origin.

4.5.2 WHOLE-ROCK REE CONTENTS

In terms of whole–rock composition, the REE concentrations of unaltered granites of the area range from 120 to 409 ppm, with an average concentration of 280 ppm (Pe– Piper 1991, 1995a, 1995b, 1998, and Pe–Piper and Piper 1998a). The bulk REE concentrations of the studied fractured samples range from 216 to 413 ppm in the Wentworth granites, 403 ppm in the Pleasant Hills pluton, 313–356 ppm in the North River granites, and from 199 to 303 ppm in the Cape Chignecto pluton (Table 4.4).

The only magmatic REE–phases found in the studied samples were grains of allanite–(Ce) and chevkinite–(Ce) in the Wentworth granites, and extremely rare relics of dissolved allanite–(Ce) in the granites of the Cape Chignecto. Therefore, a magmatic REE contribution may be recorded in the bulk REE concentrations of these samples. The rest of the REE–minerals reported in this work (Table 4.4), however, are found only in fracture–fillings, showing a clearly hydrothermal origin. Although the size of most of the studied samples is rather small (< 1mm width), fractured granites of the North River and Pleasant Hills plutons with ~8% fractures and no magmatic REE–minerals, contain up to 350 ppm of total REE in their whole–rock composition (Table 4.5).

4.5.3 HYDROTHERMAL REE-MINERALS

Hydroxylbastnäsite–(Ce). This mineral is found principally in epidote–rich fractures of the Wentworth pluton, and in lesser amounts in the biotite fractures of the Pleasant Hills pluton (Table 4.3, Appendix C.4). This carbonate replaces epidote and/or TiO₂ minerals,

Mineral name	Ideal chemical formula
<u>Silicates</u>	
Allanite-(Ce)	$CaCe \{Al2Fe^{2+}\}(Si_2O_7)(SiO_4)O(OH)$
Hingganite-(Y)	$(Y,REE,Ca)_2(\Box,Fe^{2+})Be_2[SiO_4]_2(OH)_2$
Zircon	ZrSiO ₄
Thorite	(Th,U)SiO4
<u>Oxides</u>	
Cerianite	$(Ce^{4+},Th)O_2$
Samarskite-(Y)	(Y,Fe ³⁺ ,Fe ²⁺ ,U,Th,Ca) ₂ (Nb,Ta) ₂ O ₈
Chernovite-(Y)	YAsO ₄
<u>Carbonates</u> Hydroxylbastnäsite-	
(Ce)	$(Ce,La)(CO_3)(OH,F)$
Parisite-(Ce)	$Ca(Ce,La)_2(CO_3)_3F_2$
Synchysite-(Ce)	Ca(Ce,La)(CO ₃) ₂ F

Table 4. 4 Ideal chemical formulae of the studied REE-minerals.

either as anhedral crystallites $< 10 \ \mu m$ in diameter or as 10–30 μm slender prisms (Fig. 4.4A), and is highly enriched in LREE, with a positive Eu anomaly (Fig. 4.5A).

Allanite–(Ce). Allanite–(Ce) is rarely found in fracture fillings but is always associated with epidote–rich fractures in Wentworth and Cape Chignecto plutons (Table 4.3). It occurs as irregular patches replacing hydrothermal epidote in 300 µm sized vugs (Fig. 4.4B). Allanite–(Ce) presents a REE pattern very similar to that of the REE– carbonate and also has a positive Eu anomaly (Fig. 4.5B).

Thorite. Hydrothermal thorite is found in small amounts but is widespread throughout the plutons and associated with all fracture types (Table 4.3, Appendix C.5). The most common mode of occurrence is as inclusions in allanite–(Ce), biotite, REE– carbonates and zircon. These inclusions range in diameter from 30–60 μ m (Fig. 4.4C) although some discrete thorite grains less than 10 μ m have also been found. This mineral shows variable REE–patterns showing a HREE enrichment in North River and Cape Table 4. 5 Whole-rock analyses of the studied granites.

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Pluton Sample	CCP 2039	CCP 6284	NRP 1779	NRP 4824	PHP 4672	WP 6686	WP 6591	WP 6484
Major o	2039 video (wt	0204	1//9	4024	4072	0000	0391	0404
SiO.	72 64	70) 77 10	74.02	75 85	75 16	73 50	74.02	77 24
5102 T:O:	0.33	0.16	0 10	0.15	0.25	0.20	0.22	0.10
	14.02	12 20	12.01	12.51	12.64	12 40	12 27	0.19
	14.02	12.20	12.01	12.31	2 42	2 00	104	1 2 1
re203T	2.75	1.15	1.97	1.47	2.42	2.00	1.94	1.21
MnO	0.02	0.10	0.01	0.05	0.02	0.07	0.05	0.02
MgO	0.39	0.22	0.19	0.43	0.85	0.02	0.09	0.07
	0.54	0.08	0.10	0.20	0.20	0.05	0.50	0.79
Na ₂ O	5.08	5.10	5.59	5.59	5.74 4.70	5.24 2.75	5.90	3.32
K_2O	5.38	2.92	5.45	5.43	4.79	3.75	5.37	4.85
P2O5	0.11	0.04	0.02	0.01	0.04	0.02	0.03	0.03
	0.46	0.30	0.31	0.40	0.40	0.29	0.30	0.40
Total	100.10	99.95	99.62	100.09	100.55	100.27	99.73	100.07
Trace el	ements (p	pm)	102	12.6	226	505	22.4	0.6
Ba	504	529	403	436	226	535	224	96
Rb	180	32	195	137	195	133	182	180
Sr	54	66	33	54	27	27	32	44
Y	39.8	63.4	71.5	54.2	102	95.5	67.9	56.4
Zr	273	162	162	162	342	469	274	227
Nb	18.7	19.7	20.9	20.1	37.9	37.7	36.3	46.4
Pb	59	13	15	10	16	13	19	71
Ga	21	18	21	16	28	27	22	18
As	b.d.	b.d.	b.d.	7	b.d.	b.d	b.d	21
Zn	b.d.	60	b.d.	b.d	b.d.	60	60	b.d.
Cu	b.d.	b.d.	b.d.	30	b.d.	10	b.d	10
V	8	8	b.d.	b.d	5	5	b.d	8
Cr	b.d.	b.d.	20	b.d	30	b.d	b.d	b.d.
La	68.4	28.9	72.3	63.5	108	83.5	65.9	39
Ce	122	97.1	142	122	111	150	128	82.1
Pr	15.8	7.26	17.4	16.4	22.9	20.3	15.7	10.2
Nd	56.5	25.3	62.7	60.1	80.1	77.9	56.2	39.5
Sm	10.9	6.1	13.5	12.2	16.5	17.4	11.8	9.1
Eu	0.9	0.34	0.75	0.73	1.27	1.51	0.5	0.45
Gd	8.45	6.65	11.2	9.77	15	14.8	9.8	9.1
Tb	1.52	1.56	2.13	1.75	2.84	2.73	1.85	1.61
Dy	8.19	9.95	13	10.9	17.2	17.2	11.6	9.85
Ho	1.56	2.09	2.59	2.12	3.61	3.57	2.46	2.04
Er	4.07	6.18	8.06	5.76	11.2	10.8	7.75	5.84
Tm	0.59	0.98	1.27	0.9	1.76	1.73	1.23	0.88
Yb	3.87	6.56	7.97	6.07	11	10.8	7.96	5.82
Lu	0.63	1	1.17	0.93	1.55	1.64	1.12	0.86
Со	1	84	b.d.	49	b.d.	84	73	75
Cs	4.5	0.9	1.2	0.4	2.5	1.3	1.6	1.9
Hf	7.4	5.5	5.6	5.6	10	12.8	8.8	8
Sb	b.d.	b.d.	b.d.	0.5	0.2	b.d	b.d	b.d.
Ta	1.82	3.2	2.61	2.72	4.42	5.32	4.62	4.18
Th	20.4	24.5	23.2	21.1	28.7	22.8	25.9	18.2
U	3.15	4.23	5.41	4.27	4.73	6.36	6.6	4.45
ΣREE	303.38	199.97	356.04	313.13	403.93	413.88	321.87	216.35

Chignecto plutons whereas appears LREE–enriched in Wentworth pluton (Fig. 4.5C). Low analytical totals of this phase also indicate that the thorite is hydrated.

Parisite–(Ce) and synchysite–(Ce). Parisite–(Ce) and synchysite–(Y) are both members of the bastnäsite group and they are mostly found in chlorite–rich and Fe–rich fractures in North River and Cape Chignecto plutons (Table 4.3, Appendix C.4). Parisite and synchysite appear as small (10 μ m) inclusions in chlorite or as well–formed 60–100 μ m crystals with a fibrous appearance with thorite inclusions and associated with magnetite (Fig. 4.4D). Parisite–(Ce) (Fig. 4.5D) shows weaker positive Eu anomalies than hydroxylbastnäsite–(Ce), and synchysite–(Ce) (Fig. 4.5E) shows no Eu anomalies with reference to Sm and Gd. Both parisite–(Ce) and synchysite–(Ce) show deviations from the ideal chemical formula by having smaller amounts of F (less than 0.5 a.p.f.u.) suggesting, presumably, the presence of OH[–].

Cerianite. Cerianite was found in Pleasant Hills and Cape Chignecto plutons (Table 4.3, Appendix C.5). Cerianite is found as anhedral crystallites of few µm in diameter, filling some micro–fractures adjacent to larger biotite–filled fractures (Fig. 4.4E). Cerianite although systematically LREE–enriched shows a large scatter in its REE patterns for HREE and no Eu anomalies are observed (Fig. 4.5F).

Hingganite–(Y). This mineral occurs predominantly in North River pluton associated mostly with Fe–rich fractures, and to a lesser degree with chlorite–rich fracture fillings. Rare hingganite is also found in rare Fe–rich fractures in Wentworth pluton (Table 4.3, Appendix C.6). Texturally it appears as irregular patches on magnetite clots (Fig. 4.4F) or as 20 μ m inclusions in corroded hydrothermal chlorite. This mineral is HREE–enriched and shows a well–defined negative Eu anomaly in the REE patterns (Fig. 4.5G).



Figure 4. 4 BSE images of selected REE-bearing fractures from the granites of the Cobequid Highlands. Image (A) shows needles of hydroxylbastnäsite–(Ce) forming on a mixture of TiO_2 mineral and magnetite. Image (B) shows repeated alteration of an epidote-filled vug. Epidote has been replaced initially by allanite-(Ce) while chernovite-(Y) was the last mineral to precipitate. Image (C) shows a thorite as an inclusion in a zoned zircon that fills a void in quartz. Image (D) shows needle-like synchysite-(Ce) growing on magnetite. Image (E) shows a veinlet partially filled by cerianite, adjacent to a biotite vein. Image (F) shows the association of hingganite-(Y) with Fe-rich minerals such as magnetite and chlorite, in the rare calcite-bearing fractures from the Wentworth pluton. Image (G) shows a chernovite-(Y)-bearing chlorite-magnetite fracture associated with pyrochroite. Image (H) shows grains of hingganite-(Y) in corroded chlorite, being replaced by chernovite–(Y). Image (I) shows an older chlorite–rich fracture cut by a younger chernovite–(Y)–bearing magnetite fracture. Image (J) shows crystals of zircon both as inclusions and as overgrowths on hydrothermal epidote in the Wentworth granites. Mineral abbreviations: Ab: Albite, Aln: Allanite-(Ce), Ca: Calcite, Cer: Cerianite, Cher: Chernovite-(Y), Hing-(Y): Hingganite-(Y), Oz: Quartz, Syn: Synchysite-(Ce), Ti-min: TiO2 mineral, Tit: Titanite, Fsp: Feldspar, Thr: Thorite, Hf–Zrn: hafnium-bearing zircon, H-bas: Hydroxylbastnäsite-(Ce), Pyr: Pyrochroite.

Chernovite–(Y). Chernovite–(Y) is found only in the North River and Cape Chignecto plutons in late Fe–rich and chlorite–rich fractures (Table 4.3, Appendix C.5). It occurs as discrete 10 μ m grains or filling veinlets in altered titanite, and in places associated with pyrochroite (Fig. 4.4G). In North River pluton, this mineral is found replacing a Y–rich silicate that resembles hingganite–(Y) or appears in late magnetite– filled fractures that cut older fractures filled with chlorite (Figs. 4.4H and I). This is the only REE–mineral with a systematic association with other REE–phases; it is associated with cerianite and thorite, most commonly with the latter. Chernovite–(Y) is a HREE– enriched mineral with a strong negative Eu and a positive Ce anomaly in its REE pattern (Fig. 4.5H) and contains high amounts of As (Table 4.3).

Nb–rich phases. Rare crystals of Nb–minerals are found in Pleasant Hills pluton associated with chlorite–rich fractures. They occur as inclusions in corroded chlorite, filling surfaces of crystallographic weakness such as cleavage planes and altered parts of the host grain. These grains have subequal amounts of Nb and Y and thus resemble samarskite–(Y), but due to the small size of the grains and the low analytical totals in WDS analyses, these minerals remain unidentified.

Hydrothermal zircon is present in small amounts in Wentworth pluton and occurs both as subhedral to anhedral discrete grains and as overgrowths on epidote (Fig. 4.4J). The REE–carbonates in the Pleasant Hills pluton appear more F–rich than the ones found in Wentworth pluton. The North River pluton hosts several types of hydrothermal REE– minerals found in varying amounts in the fracture fillings (0–25%). Most fractures with >3 % REE– minerals are either magnetite–rich or chlorite–magnetite fractures. The rare hydrothermal allanite–(Ce) from the Cape Chignecto pluton is chemically similar to



Figure 4. 5 Chondrite-normalized REE patterns for the hydrothermal hydroxylbastnäsite-(Ce) (A), allanite-(Ce) (B), hingganite-(Y) (C), thorite (D), cerianite (E), synchysite-(Ce) (F), parisite-(Ce) (G), and chernovite-(Y) (H), found in the granites of the Cobequid Highlands. Chondrite values after Sun and McDonough (1989).

that in Wentworth pluton, however there is a larger scatter of HREE in its REE patterns (Fig. 4.5B). In the same pluton, many of the thorite and cerianite analyses show a negative Eu anomaly (Fig. 4.5C and F). The total amount of REE minerals in the fractures of the Cape Chignecto granites is in rare cases as high as 18%. In most fractures the REE minerals vary from 0 to 6% of the modal composition of the fracture filling. There is no clear relationship between the type of the fracture filling (e.g. biotite–rich, chlorite–rich) and the type of hydrothermal REE–minerals present. It is noteworthy that the REE–carbonates not only show geographic variations in relative abundance but also in the distribution of the type of carbonate. The Wentworth and Pleasant Hills plutons host principally hydroxylbastnäsite–(Ce) in their fractures. The second most abundant REE–carbonate after hydroxylbastnäsite–(Ce) regionally is synchysite–(Ce), which together with parisite–(Ce), is mostly found in the North River pluton (Fig. 4.6).

4.5.4 TEMPERATURE ESTIMATES

The chlorite thermometer estimates show a range of temperature between 134°C and 278°C for the hydrothermal fluids circulated through the A–type granites of the Cobequid Highlands. The epidote–rich fractures of the Wentworth pluton, where patches of chlorite coexist with epidote, yield low temperatures of 157°C. The biotite–rich fractures of the Pleasant Hills pluton show temperatures from 132°C to 171°C. The chlorite–magnetite rich fractures of the North River pluton show the highest temperature estimates that range from 117°C to 236°C. The chlorites in the Cape Chignecto pluton yield hydrothermal temperatures between 152°C and 168°C, close to the range for the chlorites of the Pleasant Hills pluton.



Figure 4. 6 Ternary diagram demonstrating the distribution of the different types of REE– carbonates found in the granite plutons of the Cobequid Highlands.

4.6 **DISCUSSION**

A large variety of hydrothermal REE–minerals are found in mm–scale fractures hosted in the A–type granites of the Cobequid Highlands. The majority of the granitic samples reported in this work do not contain magmatic REE–minerals but have moderate amount of REE in whole–rock analyses, suggesting hydrothermal enrichment. The mode of occurrence of the post–magmatic REE minerals in the Pleasant Hills, North River and Cape Chignecto plutons is significantly different than that of the hydrothermal REE–Nb– rich minerals reported from the Wentworth pluton (Papoutsa and Pe–Piper, 2013). In the Wentworth pluton, most of the post–magmatic REE–minerals are dispersed in the body of the rock, whereas in the rest of the plutons they are found mostly in fractures, suggesting a more efficient circulation and transport of these elements through hydrothermal fluids.

The data show that the style of hydrothermal alteration recorded in fractures in granite plutons in the Cobequid Highlands varies not only in time but also

geographically. The fundamental question this study aims to answer is how these variations in hydrothermal alteration are related to the variations of the post-magmatic REE-circulation in the Cobequid Shear Zone, as recorded by the A-type granites in the area.

4.6.1 STYLES OF HYDROTHERMAL ALTERATION IN THE COBEQUID HIGHLANDS

Based on field observations reported in the literature from all plutons, it is evident the earliest stages of hydrothermal alteration involved Na–rich (albitization) and Ca–rich (epidotization) styles (Waldron et al., 1989; Nearing, 1991, Koukouvelas et al., 2002; Pe– Piper et al., 2004). The localized character of albitization and epidotization regionally, and also the fact that evidence of albitization was not found in the studied samples, prevents establishing their relative ages in this study. The geochemical affinity of later hydrothermal alteration is progressively more Fe–rich and is associated at first with potassic alteration (precipitation of hydrothermal biotite), and then precipitation of hydrothermal magnetite with or without coeval chlorite (Fe–rich alteration).

In the Cobequid Highlands, the early circulating fluids associated with albitization were of magmatic origin, as suggested by oxygen isotope data on mineral separates of hydrothermal albite (Pe–Piper, unpublished data). Epidotization in A–type granites can be closely associated with albitization, and may occur when the country rock is hydrated by the circulating fluids (Kaur et al., 2012). It has been observed that such metasomatic events can be associated with leaching of REE from the country rock and with the development of micro–porosity during dissolution–reprecipitation processes (Kaur et al., 2012; Putnis, 2002). The A–type granites of the Cobequid Highlands contain primary plagioclase (An₆) (Pe–Piper, 1991; 1995a, 1995b; 1998; Pe–Piper and Piper, 1998). Dissolution of feldspars has been observed in the albitized granites of the central

Cobequid Highlands and that process could be associated with the formation of sodic hydrothermal fluids. Iron–rich epidote tends to be more stable under oxidizing conditions, close to the hematite–magnetite buffer (Bird and Helgeson, 1981). The Al– Ca–rich composition of epidote (clinozoisite) that is present in the fractures of the granites indicates that epidotization in the Cobequid Highlands most likely took place under relatively reducing conditions.

Biotization in the Cobequid Highlands is observed in plutons that contain hornblende-bearing granites, whereas chloritization is observed only in biotite-bearing granites. The Cape Chignecto pluton shows evidence for both biotization and chloritization, even though it contains biotite-bearing granites, probably because hornblende occurs in the abundant mafic rocks (Waldron et al., 1989). Biotization most commonly occurs during the hydrothermal alteration of magmatic hornblende to hydrothermal biotite (Brimhal et al., 1985). Chloritization, on the other hand, usually occurs after the breakdown of magmatic biotite (Parry and Downey, 1982; Parneix et al., 1985) at low temperatures (Morad et al., 2011). The distribution of hydrothermal chlorite and biotite in the fractures of the granites in the Cobequid Highlands appears to depend on the availability of magmatic biotite versus calcic amphibole, whether in the granites or gabbro.

In the Cobequid Highlands, both hydrothermal biotite and chlorite approach iron–rich end–members (Fig. 4.2B and C), suggesting either that fluids or bulk rock, or both, were iron–rich. The A–type granites of the Cobequid Highlands are in general ferroan rocks and therefore may have provided some iron for the formation of hydrothermal biotite and chlorite. Other possible sources of iron for the fluids are the mafic rocks in these plutons

which are ferro–gabbros (Pe–Piper and Piper, 2003) or the late lamprophyre dykes and gabbro stocks (MacHattie and O'Reilly, 2008).

The ~323 Ma iron–rich hydrothermal event along the Cobequid Shear Zone is associated with carbon–rich fluids, as indicated by the presence of iron–carbonates such as ankerite and siderite in fractures, for example in the large Londonderry iron deposit (Kontak et al., 2008). Iron is the most abundant element in high–temperature hydrothermal fluids that evolve from high–level magmas (Simon et al., 2004). Precipitation of Fe–oxides such as magnetite, hematite and martite has been interpreted to result from evolving, progressively more oxidizing, hydrothermal fluids (Figueiredo e Silva et al., 2011). Fluids related to this type of mineralization are usually characterized by very high temperatures, high salinity, and coexist with carbon–rich fluids (Borrok et al., 1998). Iron–enrichment of hydrothermal fluids has been observed to be associated with circulation of REE and as a result, hydrothermal REE–minerals have been found in magnetite deposits (Pršek et al., 2010; Figueiredo e Silva et al., 2011).

Although there is no direct evidence (i.e. measurements from fluid inclusions) for increased salinity in the hydrothermal fluids in the Cobequid Highlands, observations have been made that may support that hypothesis. The early Na-rich alteration in the central Cobequid Highlands was associated with fluids of increased halogen component, on the basis of high Rb, Cs, Hf, Nb and Zn concentrations in areas of pervasive alteration (Pe-Piper et al., 1991). In the same area, biotite from late minette dykes (>80% modal biotite) shows elevated concentrations of Cl (up to 1.6 wt.%) and F (up to 1.3 wt.%) (Pe-Piper, unpublished data), and high concentrations of synchysite-(Ce) in fractures (Wisen, 2015). Finally, Owen and Greenough (1999) reported the presence of scapolite-bearing pegmatite in the western Cobequid Highlands, whose marialitic composition was

interpreted to reflect the presence of fluids with at least 30-40 mol% NaCl. Smith and Henderson (2000) demonstrated, from their fluid inclusion study in the Bayan Obo REE deposit, that REE can be efficiently mobilized in hydrothermal fluids with 6-10 wt% NaCl equivalent at temperatures from 240 to 340 °C. The hydrothermal temperatures necessary for the mobility of REE decreased with increasing salinity. The West Moose River granites, adjacent to the North River pluton, are strongly fractured and albitized (Pe-Piper et al., 1991). These granites also have an impressive variety of hydrothermal REE-minerals, but no primary REE-minerals were recognized except for zircon, and a strongly altered grain that resembles a relict chevkinite-(Ce) (Pe-Piper et al., 2015). Therefore it is possible that hydrothermal fluids of significant salinity were circulating through the faults and associated fracture systems of the Cobequid Zhear Zone from early stages (Na-alteration). Such fluids could therefore sufficiently mobilize the REE from the fractured host granites, which could explain the small amounts of magmatic REE minerals that were preserved in the plutons of the central Cobequid Highlands.

4.6.2 SPATIAL VARIATION IN HYDROTHERMAL ALTERATION

In the eastern part of the Cobequid Highlands, the dominant style of hydrothermal alteration is epidotization, as indicated by the widespread epidote–rich fractures in the Wentworth pluton. The decreased amounts of epidote in the biotite–rich fractures of the nearby Pleasant Hills pluton shows a westward transition in the style of hydrothermal alteration, from epidotization to biotization, which is dominant mostly in the western Cobequid Highlands.

A gradual increase of magnetite in the fractures is observed mainly from the Pleasant Hills pluton towards the central part of the Cobequid Shear Zone, with the highest concentration in the fractures of the North River pluton. However, the largest Fe deposits

of the Cobequid-Chedabucto Fault Zone – the Bass River magnetite deposit and the giant Londonderry deposits – developed along faults south of the Pleasant Hills and Wentworth plutons (Fig. 4.7). This suggests that in the Cobequid Highlands during the late Carboniferous, Fe–rich hydrothermal alteration was focused mainly along the Cobequid fault.

4.6.3 CORRELATION OF THE OBSERVED HYDROTHERMAL STYLES WITH DATED HYDROTHERMAL EVENTS

The sequence of changes in the hydrothermal circulation through the faults and fracture systems across the Cobequid Shear Zone can be extrapolated from geochronological data and textural evidence of minerals. Epidotization probably took place shortly after the emplacement of the plutons, while these faults were still active. The epidotization of the Wentworth granites represents, therefore, one of the earliest stages of hydrothermal circulation. Epidotization occurred until the first stages of biotization, in order for epidote to precipitate in the oldest biotite–filled fractures of the Pleasant Hills pluton (Fig. 4.7). The formation of biotite–filled fractures, during potassic alteration, probably postdated the 353 ± 4 Ma riebeckite in the West Moose River pluton and continued for ~20 Ma after pluton emplacement (Fig. 4.7).

The age of the magnetite-rich fractures found in the granites cannot be precisely determined by textural data alone. Formation of hydrothermal magnetite in the fractures of the A-type granites in the Cobequid Highlands seems to have been active for a long time, as indicated by textural data (Fig. 4.3C to F). This suggests precipitation of hydrothermal magnetite from, at least the later stages of biotization (Fig. 4.7) until the latest hydrothermal stages, when almost monomineralic magnetite fractures formed.

Magnetite is absent from the largest S– and Fe–rich deposits in mainland Nova Scotia related to the ~323 Ma IOCG mineralization, such as the Copper Lake and the Londonderry deposits (Murphy et al. 2011). However, the Bass River Co–bearing





magnetite deposit (Belperio, 2010), close to the Londonderry fault at the southern edge of

the Pleasant Hills pluton, has abundant magnetite in breccias of the 321-324 Ma

Londonderry Formation (ages from Murphy et al., 2011) that include clasts of the Pleasant Hills granite. The high gravity anomaly that characterizes the Bass River magnetite deposit coincides with an area of high gravity anomaly between the North River and Pleasant Hills plutons and is represented by hematite–rich fractures (Belperio, 2010). Furthermore, the high linear magnetic anomaly that follows the nearby Londonderry fault is interpreted as the result of discontinuous fault–controlled emplacement of hydrothermal magnetite (Belperio, 2010). This suggest that during the ~323 Ma IOCG mineralizing event magnetite precipitated in the area and that these iron– rich fluids which circulated along the Cobequid and Londonderry faults reached the A– type granites in the central part of the Cobequid Highlands. The late magnetite–rich fractures in the Pleasant Hills and North River pluton are probably of the same age.

4.6.4 CHANGES IN THE HYDROTHERMAL FLUIDS

In the eastern part of the Cobequid Highlands, the epidote–rich fractures indicate Ca– rich fluids. The presence of minor chlorite, in places mixed with the epidote, suggests that at some stage during epidotization, the fluids became relatively more enriched in Fe. This led to co–precipitation of chlorite with epidote in the Wentworth pluton. The hydrothermal fluids appear to be even more enriched in Fe in the Pleasant Hills pluton, where Fe–rich minerals like biotite dominate over epidote in the fracture fillings.

The sequence of the hydrothermal types of fluids, as recorded by fractures in the late Devonian granites in the Cobequid Highlands, suggest that the chemistry of the fluids, in general, became progressively more Fe–rich. Pyrite is stable under extremely reducing conditions (Hutcheon, 1998) and its presence in the fracture fillings of the Wentworth pluton further suggests that epidotization took place under reducing conditions. Grains of pyrite surrounded by colloidal hematite have been found dispersed in the Wentworth

granites (Papoutsa and Pe–Piper, 2013), which is interpreted as a product of pyrite oxidation at very low pressure (Bhargava et al., 2010). Pyrite is also present in small inclusions in the magnetite–rich fractures of the Wentworth pluton. These observations suggest that the oxidation state of the fluids changed from reducing during epidotization to more oxidizing during biotization.

The bulk Fe content of fracture fillings, estimated from abundances of different minerals, shows a positive correlation with temperature estimated from chlorite composition, suggesting the influence of temperature on the solubility and circulation of Fe in aqueous solutions (Fig. 4.8). In some cases, the chlorite appears in patches, coexisting with other minerals, and it could have formed later as a product of alteration. Therefore the chlorite–derived temperature estimates for those fractures should be used with caution.

The mineral assemblages in the fracture fillings of each style of hydrothermal alteration suggest that there were changes in the chemistry of the fluids not only on a regional scale, but also locally and over a short time scale. The patches of andradite in the epidote–rich fractures of the Wentworth pluton (Table 4.2) suggest that the water fugacity was probably variable, so as to form a less hydrated, but otherwise Ca-Al-rich, mineral phase such as andradite during epidotization. The limited presence of calcite, fluorite, and pyrite in the studied epidote–rich fractures suggest that although Ca, C, F, and S were present in the fluids, their concentrations were probably moderate to low, at the time.

4.6.5 CORRELATION OF REE-RICH SECONDARY MINERALS WITH STYLES OF HYDROTHERMAL ALTERATION

The epidote fractures of the Wentworth pluton have the smallest amounts of hydrothermal REE–minerals (hydroxylbastnäsite–(Ce) and thorite), compared to biotization and Fe–rich hydrothermal alteration (Table 4.2). These fractures are hosted in the intrusions with the highest amounts of magmatic REE–minerals, regionally.



Figure 4. 8 Variation of the FeOt bulk composition from the fracture fillings of the A–type granites of the Cobequid Highlands, in relation to the observed styles of hydrothermal alteration and the observed REE–mineral assemblages. Increase in oxygen fugacity is qualitatively implied based on the relative abundance of magnetite.

Replacement of primary allanite–(Ce) by hydrothermal REE–carbonates in the Wentworth (Papoutsa and Pe–Piper, 2013) and Cape Chignecto granites suggest that the magmatic REE–minerals were the primary source of REE in hydrothermal stages. The low concentrations of hydrothermal REE–minerals in the fractures of the Wentworth granites can therefore be attributed to limited mobility and transport of REE in the hydrothermal fluids during epidotization.

Secondary fluorite is scarce in these fractures, although it is generally present as rare dispersed grains in the Wentworth granites, most commonly associated with amphiboles. The release of F into the hydrothermal system is taken to be the result of the alteration of amphibole in the early granites of the Wentworth (Pe–Piper, 2007), and resulted in the mobilization of REE (Papoutsa and Pe–Piper, 2013). The epidote fractures, however, lack hydrothermal biotite, which is abundant in the nearby Pleasant Hills pluton and is associated with the amphibole breakdown. Therefore, the epidotization in the Wentworth pluton was probably an early event that predated the breakdown of the primary amphiboles and the major introduction of REE in the hydrothermal system.

The amount of hydrothermal REE–minerals in the plutons of the Cobequid Highlands appears to increase from epidotization to biotization (Fig. 4.8), with a dominance of LREE minerals and particularly REE–carbonates. The solubility of carbonate minerals decreases from reducing to oxidizing conditions (Tabelin and Igarashi, 2009). Therefore the increase of the oxidation state of the fluids from epidotization to biotization was an important factor that enhanced the precipitation of REE–carbonate minerals.

The Fe–rich style of hydrothermal alteration presents the largest variety and highest amounts of hydrothermal REE–minerals, such as hingganite–(Y), hydroxylbastnäsite–(Ce), parisite–(Ce), synchysite–(Ce), cerianite and chernovite–(Y). The precipitation of chernovite–(Y) is also enhanced by strongly oxidizing fluids

(Tabelin and Igarashi, 2009) and therefore, the increasing oxidation state of the fluids during the Fe–rich hydrothermal alteration resulted in the formation of that mineral.

4.6.6 CONTRIBUTION OF OTHER SOURCES TO THE FORMATION OF THE HYDROTHERMAL REE-MINERALS

Unaltered A–type granitic rocks of the Cobequid Highlands (on the basis of low LOI values and absence of fractures) show lower amounts of bulk REE (average of 280 ppm, Pe–Piper, 1991; 1993a; 1995b; 1998; Pe–Piper and Piper, 1998a), and particularly the LREE, compared to the fractured granites in this study (Fig. 4.9A), which correlates with the presence of the strongly LREE–rich hydrothermal minerals. Furthermore, the positive correlation between the bulk Y–Nb–Th–U–REE and Fe₂O₃ corresponds with an increase in magnetite–rich fractures (Fig. 4.9B) in the studied samples for each pluton, confirming a hydrothermal REE overprint on the whole–rock composition of these rocks.

The presence of minerals like parisite–(Ce) and synchysite–(Ce) (Table 4.3) suggest elevated amounts of Ca and C in the hydrothermal fluids. The late Paleozoic granites of the Cobequid Highlands, however, are alkali feldspar granites with small amounts of primary calcic minerals like plagioclase, less than 1 wt. % CaO and no primary carbonate minerals. Therefore, even though the REE were released from the granite during hydrothermal alteration, the amount of Ca and C released from the granites alone was probably not sufficient as to form Ca–rich REE–carbonates.

A potential source of Ca would be the mafic rocks associated with the granites, however, the North River pluton consists predominantly of granite with only minor marginal gabbro (Pe–Piper, 1991). A larger potential source of Ca and especially C for the formation of the REE–carbonates is the succession of carbonate rocks and



Figure 4. 9 Comparison between the whole–rock composition of the studied fractured granites and unaltered granites of the area (data for unaltered granites compiled from Pe–Piper, 1991; 1995a; 1995b; 1998; and Pe–Piper and Piper 1998a). Figure (A) shows the distinct increase in LREE in the fractured granites as a result of hydrothermal LREE–carbonates and cerianite precipitation. Image (B) shows the positive correlation of rare metals, the abundance of magnetite–rich fractures and the whole–rock Fe₂O₃ of the studied granites.

evaporites of the Visean Windsor Group (south of the central and eastern Cobequid Highlands), or carbonate rocks of the Horton Group south of the North River pluton (Pe– Piper, 1995a; Piper, 1996). Mineralization related to fluids derived from the Windsor Group is characterized by the presence of Pb, Zn and Cu (Kontak et al., 2006), but no minerals with these elements has been found in the REE–carbonate-bearing fractures. It is therefore more likely that the host granites together with the carbonates of the Horton Group were responsible for locally elevated amounts of Ca and C in the hydrothermal fluids, leading to preferential precipitation of parisite–(Ce) and synchysite–(Ce) in the fractures of the central Cobequid Shear Zone.

The presence of the As–rich mineral chernovite–(Y) in the fractures of the North River and Cape Chignecto plutons suggests the circulation of As in the late REE–bearing hydrothermal fluids. The only granitic samples that show As above detection limit are those with chernovite–(Y)–bearing fractures (Table 4.4). Therefore, the host granites were unlikely to have been the source of As in the hydrothermal system. The most probable sources of As in the area are the shales of the Horton Group which contain moderate amounts of As (13 ppm, Pe–Piper, unpublished data). The release of As into hydrothermal fluids is believed to be associated with the breakdown of Fe–Mn oxides (Tabelin and Igarashi, 2009). The presence of pyrochroite in the chlorite fractures of Cape Chignecto could be related to the formation of chernovite–(Y).

4.6.7 GEOLOGICAL CONTROLS ON THE HYDROTHERMAL REE-ENRICHMENT IN THE COBEQUID HIGHLANDS

It is evident from the spatial variations in the style of hydrothermal alteration and abundance of hydrothermal REE–rich minerals (Fig. 4.7) that specific geological factors have favored the concentration of these elements in certain areas of the Cobequid Highlands. The behavior of REE is greatly influenced not only by chemical parameters, such as the composition of the hydrothermal fluid and the country rock, but also by physical parameters like pressure and temperature that will be evaluated in this section.

The importance of large scale crustal structures in focusing hydrothermal fluids and controlling mineralization has been underlined in several studies (McCaig et al., 1990; Abraham and Spooner, 1995; Kreiner and Barton, 2009; Murphy et al., 2011). During movement along the faults of the Cobequid Shear Zone, the plutons in the central part of the Cobequid Highlands were located in the stepover zone between the Rockland Brook fault to the east and the Kirkhill fault to the west and became strongly fractured (Pe–Piper and Piper, 2003). Therefore that area provided more pathways for an effective circulation of the hydrothermal fluids, compared to the rest of the shear zone. The Cobequid–Chedabucto Fault Zone saw its main activity at the Serpukhovian–Bashkirian boundary at ~318 Ma, but as shown by the Copper Lake ~323 Ma date the fault zone was

active before that time and motion continued at least until the Moscovian (~307 Ma) (Murphy et al., 2011). Plutons such as the North River and Cape Chignecto plutons, closer to the late cataclastic motion on the Cobequid fault, were able to interact with hydrothermal fluids for longer periods of time, compared to the Wentworth pluton which is bounded by the Rockland Brook fault. Furthermore, this suggests that the later REE– Fe–rich, oxidizing fluids, which were concentrated along the Cobequid and Londonderry faults in late Carboniferous, did not significantly affect the Wentworth pluton.

Besides the creation of pathways for the hydrothermal fluids by faulting, the tectonic setting plays an equally important role in influencing hydrothermal pressure that in turn affects the stability and solubility of hydrothermal minerals. High pressures increase the solubility of Fe and Mg in hydrothermal stages, permitting the fluids to reach oversaturation (Kalczynski et al., 2012). A decrease of pressure, therefore, can control the precipitation of Fe-rich minerals, such as magnetite and siderite. Such a decrease may occur in shear zones when the pressure changes from lithostatic to hydrostatic in the extensional segments of the faults (Kalczynski et al., 2012). Fluctuation of the pressure of fluids during hydrothermal circulation is a phenomenon that has been correlated with vein systems associated with shear zones. In such settings, the pressure of fluid circulation may reach suprahydrostatic values when fluids migrate into shallow, heterogeneous, and stressed crust during shear (Sibson, 1996). The faults through which the fluids circulate may control the hydrostatic pressures through what is known as faultvalve behavior (Sibson, 1992). Even though fault-valve behavior is influenced mainly from the prevailing stress, and the existing faults of the area, the pressurized fluids that result from such behavior may, in turn, have an impact on the permeability of the area by creating fracture meshes. These structures are best developed in areas of transfer faults,

dilational jogs, and may precede the formation of large fault zones (Sibson, 1996). Sibson et al. (1988) associated the creation of these vein sets with shear zones where pre-existing high angle faults were reactivated during horizontal compression from crustal shortening and strike-slip faulting with a sense of thrusting. Fractures in these vein systems through which pressurized fluids have passed tend to form conjugate sets (Sibson, 1996).

In the central part of the Cobequid Highlands (North River and West Moose River plutons) Piper et al. (1993) described similar deformation to that associated with pressurized fluids as mentioned above. This area is characterized by high angle faults at the margins of the intrusions, while the brittle deformation in the plutons is manifested by joint sets of fractures, some of which with dilational features. Furthermore, the late Carboniferous deformation in the Cobequid Shear Zone which is associated with the IOCG mineralization, involved dextral strike-slip motion and with a sense of thrusting (Murphy et al., 2011), resembling thus the tectonic setting required for high hydrostatic pressures and high Fe solubility in the fluids. Therefore intense fracturing of the plutons in the central part of the Cobequid Highlands could be taken as evidence of the presence of Fe-rich pressurized fluids and local release of stress that enhanced the precipitation of Fe oxides, particularly in the fractures of that area.

Temperature is another important factor that controls the stability of minerals from magmatic to hydrothermal stages. Hydroxylbastnäsite–(Ce) is stable at low hydrothermal temperatures (Fig. 4.8), whereas parisite–(Ce) and synchysite–(Ce) form under relatively higher temperatures (Williams–Jones and Wood, 1992). Positive Eu anomalies in hydrothermal minerals are indicative of low (below 250 °C) hydrothermal temperatures (Sverjensky, 1984). The association of calcite with synchysite–(Ce), which is abundant only in the fractures of the central Cobequid Shear Zone, requires higher

temperature (Fig. 4.8) and pressure in hydrothermal stages, compared to the other REE– carbonates. In the east, lower temperatures led to the formation of hydroxylbastnäsite– (Ce) with positive Eu anomaly (Fig. 4.5A). High–temperature metasomatism (>400 °C) in the central Cobequid Highlands has been inferred by Pe–Piper (1995a) from the chemistry of garnets found in meta–carbonates south of the North River pluton. It is, therefore, evident that there is a correlation between temperature and the type of the dominant REE–carbonate in the fractures, which suggests a significant control as well of temperature on the REE–mineral species that precipitated from the fluids.

The present work shows that the hydrothermal REE–enrichment of the A–type granites of the Cobequid Highlands shows a systematic variation that is highly dependent on the timing and style of regional hydrothermal alteration. The abundance of hydrothermal REE–minerals as well as REE mineral speciation increased with time as the regional hydrothermal system became more Fe–rich, progressively oxidizing and of higher temperature. High temperatures may have been driven by the minor late mafic intrusions: lamprophyre dykes in the west and small alkaline gabbro stocks in the east. Other sources besides the granitic rocks have contributed to the chemistry of some hydrothermal REE–minerals, such as chernovite–(Y).

The observed correlation between the abundance of hydrothermal REE–minerals, timing and specific types of hydrothermal alteration may have implications for REE mineral exploration. Hydrothermal systems elsewhere, with high amounts of REE and fluids with similar chemical evolution, may form significant mineral deposits at the final stages of circulation.

4.7 CONCLUSIONS

The late Devonian A–type granites of the Cobequid Highlands in Nova Scotia are located in an area where different styles of hydrothermal alteration persisted over a period of several tens of million years after magmatic emplacement. The nature of the hydrothermal fluids varies both geographically and in time and has affected the behavior and distribution of hydrothermal REE–minerals that precipitated in fractures.

Epidotization in the Cobequid Highlands was accompanied by the circulation of minor S, resulting in the formation of hydrothermal pyrite under reducing conditions and low temperatures. These conditions enhanced the release of As from the sedimentary rocks of the area and along with the presence of C and F, probably initiated the mobilization of the REE hosted in primary minerals in the granites, preventing, however, the precipitation of hydrothermal REE–carbonates in these fractures.

As the hydrothermal fluids became progressively more Fe–rich and relatively more oxidizing, REE–fluorocarbonates precipitated, principally in the biotite–filled fractures of the Pleasant Hills pluton and the chlorite–rich fractures of the North River pluton. Early hydrothermal biotite co–precipitated with epidote, whereas in the later stages of biotization the formation of epidote ceased and magnetite started to precipitate with biotite. There is no clear textural relationship between hydrothermal biotite and chlorite that would indicate a paragenetic sequence between these two minerals, however the latter is mixed with higher amounts of magnetite and is concentrated mostly in the North River pluton. These fractures contain the highest amounts and the largest variety of hydrothermal REE–minerals including hingganite–(Y) and minerals of the bastnäsite group. Localized increase of Ca and C in the fluids, as well as higher hydrothermal temperature, resulted in preferential concentration of synchysite–(Ce) and parisite–(Ce)

in the fractures of the central part of the Cobequid Shear Zone over hydroxylbastnäsite– (Ce) which dominates in the east.

By Bashkirian time, most of the hydrothermal circulation was focused along the Cobequid–Chedabucto Fault Zone, resulting in the widespread iron alteration of the adjacent plutonic bodies in the central and western Cobequid Highlands. The younger almost monomineralic magnetite fractures that contain chernovite–(Y) and cut the chlorite–rich fractures in the North River pluton are probably related to that Fe–alteration event. Chernovite–(Y) formed in these fractures from the alteration of hingganite–(Y), which was enhanced by the change of the oxidation state of the hydrothermal fluids and precipitation of As.

CHAPTER 5 AN EVOLUTIONARY HISTORY OF AVALONIA: NEW INSIGHTS FROM THE COBEQUID HIGHLANDS

5.1 **INTRODUCTION**

In chapter 3 it was shown that the mantle and lower crust have played an important role in the petrogenesis of the A-type granites in the Cobequid Highlands. It is therefore important to understand the geological evolution of the Avalonian crust in order to evaluate the changes in the Avalonian lithosphere since it provided a) the necessary lithologies as magmatic sources, and b) the geodynamic conditions for the late Paleozoic A-type granites to form throughout west Avalonia. This chapter shows how geochemical evidence from dated igneous rocks can provide essential data pertinent to paleogeographic reconstructions since they provide a snapshot of mantle and crustal conditions at different times.

The scope of this chapter is not only to review the geological evolution of Avalonia from Neoproterozoic to late Paleozoic, as presented in the literature, but most importantly to correlate specific evolutionary stages of the paleocontinent with relevant geological evidence from the Cobequid Highlands and elsewhere in west Avalonia. The geodynamic evolution of the Neoproterozoic Cobequid Highlands is approached through stratigraphic, structural, geochemical, and isotopic comparisons with Avalonian basement elsewhere in order to evaluate the crustal conditions that gave rise to A-type granitic magmas. As demonstrated in chapter 3, the isotopic character of the late Paleozoic Atype granites may be variable and thus reflect a significant lower crustal heterogeneity. This heterogeneity is further explored in this chapter, and the above correlations are combined with available paleomagnetic data in order to evaluate the ultimate origin of the Neoproterozoic crustal blocks in the Cobequid Highlands and other parts of the nowdismembered Avalonia. Geochemical data, predominantly from mafic rocks from different parts of west Avalonia, are compared since they can provide evidence for the lithospheric mantle from different crustal blocks. These findings are then compared with previous and new paleoreconstructions in an attempt to explain the evolution of Avalonia from the Neoproterozoic to late Paleozoic.

In this chapter, the nature and timing of Ordovician magmatism in west Avalonia is reviewed and discussed in order to evaluate the paleogeography of several parts of the microcontinent during the closure of Iapetus and the opening of the Rheic oceans. The last part of the chapter deals with the late Paleozoic evolution of west Avalonia during the closure of the Rheic Ocean. A brief comparison of known coeval A-type granites in west Avalonia is presented with evidence from the literature and from the findings of this thesis, in order to evaluate any petrogenetic similarities between these rocks, on a broader scale. Finally, the late Carboniferous fluid circulation and mineralization observed in the Cobequid Highlands is discussed with an emphasis on the structural controls along the Cobequid-Chedabucto Fault Zone during the amalgamation of Pangea.

Avalonia is a terrane with a rather complex history (Keppie and Dallmeyer, 1989), which is subdivided into west Avalonia (Atlantic Canada and US), and east Avalonia (British Isles and Belgium). West Avalonia, as part of the Appalachian Orogen, is considered to be a peri-Gondwanan fragment that was accreted to the Laurentian margin after Ganderia and before the docking of Meguma in late Devonian (van Staal et al., 2009). Although the peri-Gondwanan origin of Avalonia is now well established, the original position and orientation of this terrane in the Neoproterozoic are still

controversial. To make things more complicated, different parts of the Avalonian basement show contrasting isotopic and chemical characteristics that have led researchers to interpret Avalonia as a collage of different crustal fragments, and therefore a composite terrane (Nance and Murphy, 1996). Willner et al. (2013), however, argued that Avalonian fragments originated from an originally coherent microcontinent along the continental margin of Gondwana. Isotopic evidence (Sm-Nd) from crustally derived rocks in west Avalonia indicates the presence of a basement formed at ca 1 Ga as juvenile crust, whereas east Avalonia shows isotopic signatures of an older (1.3-1.8 Ga) basement (Nance and Murphy, 1996). It is argued thus that west Avalonia was adjacent to the Amazonian craton in the Neoproterozoic, whereas east Avalonia occupied positions adjacent to the West African craton (Nance and Murphy, 1996).

5.2 PRECAMBRIAN AVALONIA - AN EARLY AVALONIAN ASSEMBLY

The Neoproterozoic eon is an important interval in Earth's history that was manifested by widespread rifting and orogenic activity accompanying the breakup of Rodinia and the assembly of Gondwana (Trompette, 1997). During mid-Neoproterozoic (800-700 Ma), extensive rifting within Rodinia initiated its breakup (Evans, 2009). The final stages of the evolution of Rodinia are critical to most studies concerning the subsequent assembly of Gondwana, since that period is the starting point for many paleoreconstruction models for the latter (Faleiros et al., 2011; Saraiva dos Santos et al., 2014).

Although understanding the arrangement of the numerous cratons within Rodinia (i.e. the paleolatitudes for Laurentia and Baltica) has suffered from the lack of reliable paleomagnetic data, reconstruction models were used in the past decades, based on

isotopic data and assumptions, about the location of paleopoles (Nance and Murphy, 1996, Thompson et al., 2012). Nonetheless there is always a degree of uncertainty when modelling plate reconstructions, and particularly in the Proterozoic. The fact that the ancient oceans have been consumed by subduction and that, by definition, all paleomagnetic poles can only provide constraints for latitude and not longitude cause major uncertainties. Rodinia itself has several different paleogeographical versions as summarized in Evans (2009). The applied model of Rodinia used in this work is the one, which in the author's opinion, reconciles best the published data reported in this chapter. Such uncertainties may influence not only the modelled paleogeography of Gondwana, but have further implications on the early evolution of Neoproterozoic proto-Avalonia as a peri-Gondwanan terrane. As Murphy et al. (2004) argued, there is a need for more paleomagnetic data to document the movement histories and connections of West Gondwana and peri-Gondwanan terranes.

In the past decade, more paleomagnetic data became available for Laurentia (Weil et al., 2004), Baltica (Brown and McEnroe, 2004; Pisarevsky and Bylund, 2006), Amazonia (D'Agrella-Filho et al. 2008) and the African cratons (D'Agrella-Filho et al., 2004). For the evolution of Avalonia, the histories of these cratons are critical, since it is argued that they all have been proximal to parts of Avalonia at different times (Nance and Murphy, 1996; Murphy et al., 2004). The updated paleomagnetic database (such as the PALEOMAGIA archive), as well as more precise geochronological controls, resulted in the most up-to-date reconstruction models for Rodinia (Evans, 2009; Johansson, 2014) and Gondwana (Torsvik and Cocks, 2013). For the purpose of this thesis, this section will

focus mostly on the evolution of West Gondwana, along the periphery of which Avalonia is suggested to have evolved (Murphy et al., 2001).

In order to understand the Neoproterozoic history of Avalonia, it is necessary to evaluate the configuration of Rodinia during its breakup at ca 750 Ma. For that purpose, paleoreconstructions have been made using G-Plates software for Windows (www.gplates.org). The model for Rodinia adopted here includes the refined version of the SAMBA (South AMerica-BAltica) configuration (Gower et al., 1990; Hoffman, 1991; Moores, 1991; Dalziel, 1997; Li et al., 2008) in which Amazonia, West Africa and Baltica formed a single coherent landmass until at least 800 Ma (Johansson, 2014). In this configuration, the Rodinia supercontinent is all-inclusive, as proposed by Evans (2009), and surrounded by a Panthalassan-type ocean, the Brasiliano Ocean (Fig. 5.1A). Johansson (2014), however, did not present any paleolatitudes for the various cratons. Therefore the Neoproterozoic paleolatitudes, in the models herein, were constrained with reference to the position of Laurentia, which itself has been determined by published paleopoles (Table 5.1).

At the onset of Rodinia breakup at ca 750 Ma, Baltica, according to the SAMBA configuration, is considered to have occupied the southern margin of Rodinia. Pisarevsky et al. (2003) and Li et al. (2008) showed Amazonia south of Laurentia on the western flank of Baltica and that position is adopted here as well. Amazonia at 750 Ma, in the models here, is separated from the Congo craton and eastern South America (Parana, Rio de la Plata cratons) by the Brasiliano Ocean. Amazonia is separated from the Laurentian margin by the Grenville-Sunsas orogenic belts (Johansson, 2014).

The position of Avalonia in the Neoproterozoic is still controversial. Based on isotopic data from crustally derived rocks, it is interpreted that the Avalonian juvenile basement developed in the peri-Rodinian ocean (Murphy et al., 2001). There are no



Figure 5. 1 Reconstruction models produced by G-plates illustrating (A) the break-up of Rodinia at ca 750 Ma, (B) and (C) the transition to a Gondwanan configuration from 650 to 600 Ma based on the SAMBA model (Johansson, 2014), and (D) the separation of Gondwana, Laurentia and Baltica during the opening of Iapetus Ocean at 540 Ma (as in Torsvik et al., 2014). Paleolatitudes for west Avalonia at ca 570 Ma (Pisarevsky et al., 2012) are indicated for comparison. Lines of latitude and longitude are at 30 ° intervals. Abbreviations: Av= Avalonia, Gr-Sns= Grenville-Sunsas orogenic belts, P= Pampean craton, R= Rio de la Plata, Pn= Parana.

paleomagnetic data from Avalonian rocks of this age, however Murphy et al. (2004) placed the 1 Ga proto-Avalonian crust at high southern latitudes (~90°S). Even though that is not paleomagnetically constrained, it was the preferred position in order that Avalonia would occupy a position along the western flank of Amazonia at the time. The configuration of Rodinia used in Murphy et al. (2004), however, was different from the one adopted here. Therefore, although the position of Avalonia relative to Amazonia remains the same, the paleolatitudes in the configuration presented here are significantly different.

The published paleomagnetic studies for the Neoproterozoic Avalonia (McNamara et al. 2001, Vizan et al. 2003, Thompson et al. 2007, Murphy et al. 2012) are all on Ediacaran rocks (ca 620 Ma), and they place west and east Avalonia at a paleolatitude between 12 to 34°S, with the exception of New England Avalon that shows a higher paleolatitude of 37-55°S (Pisarevsky et al., 2012). Prior to Ediacaran, the Amazonian affinity of Neoproterozoic detrital zircons (Barr et al., 2012) favors a position within the Brasiliano Ocean along the margin of the Amazonian craton (Fig. 5.1a). This, however, precludes an association of West Africa with east Avalonia at that time, a view that is further supported by the presence of early Neoproterozoic zircons in east Avalonian rocks that are absent in West African sources (Pollock et al., 2012; Willner et al., 2013). Given the configuration of Rodinia at the time, proto-Avalonia is placed in the reconstructions herein at southerly paleolatitudes of ~20° (Fig. 5.1A).

5.2.1 BREAKUP OF RODINIA

The Rodinia breakup began at ca 750 Ma, with rifting west of Laurentia (Fig. 5.1A) (Johansson, 2014). Counterclockwise rotation of eastern South American and the
African cratons initiated the change from a Rodinian to Gondwanan configuration (Li et al., 2008). These rotations resulted in the closure of the Brasiliano Ocean, while the separation of the South American cratons (Parana, Rio de la Plata) from Congo led to the opening of the intervening Adamastor Ocean (Fig. 5.1B). The closing of Brasiliano and subsequently of Adamastor oceans would have created long-lived subduction systems adjacent to the Amazonian margin until the final collision of Amazonia with the eastern South American cratons and Africa in the Ediacaran (Fig. 5.1C) (Johansson, 2014). This is indicated by the Brasiliano Orogenic Belts which lie between the Amazonian and South American cratons and record 670-550 Ma arc-related magmatism (Bento dos Santos et al., 2015). This is when the Avalonian terranes also record a major phase of arcrelated magmatism (Murphy et al., 2001). Avalonia is thought to have accreted to Amazonia before ca 600 Ma (Fig. 5.1C), since it records intracontinental extension by 590 Ma (Balintoni and Balica, 2013). During that period, the Brasiliano fold belts within Amazonia record substantial reworking of the crust, in a branched system of mobile belts, with dextral shear zones extending to the African cratons (Fuck et al., 2008). There was, therefore, a large amount of strike-slip motion accommodating the assembly of West Gondwana (Evans, 2009).

5.2.2 AVALONIA DURING THE ASSEMBLY OF GONDWANA

By 550 Ma West Gondwana was assembled, or nearly so (Johansson, 2014). By that time, Avalonia in the models herein is predicted to have dextrally transferred along the margin of Gondwana during counterclockwise rotation of Amazonia (Fig. 5.1C). Laurentia, on the other hand, maintained an equatorial paleolatitude at ca 550 Ma (Hodych et al., 2004), while Gondwana drifted at high southern paleolatitudes during the opening of Iapetus Ocean (Fig. 5.1D). Faunal and stratigraphic data from Avalonia also favor a position at high southern paleolatitudes during the late Neoproterozoic (Pollock et al., 2012). This lateral transport of Avalonia along the margin of Gondwana is documented by syn-igneous shearing of the Neoproterozoic Frog Lake and Debert River plutons in the Cobequid Highlands (Pe-Piper et al., 2010; Pe-Piper et al., 1996). Synmagmatic deformation is indicated by pre-full-crystallization orientation of feldspar in granodioritic rocks, plastically deformed veins, and presence of C-S structures. Dextral strike-slip displacement has also been interpreted in late Neoproterozoic- early Cambrian sedimentary rocks of New Brunswick, where the detrital zircons reflect a change from purely Amazonian sources in the Neoproterozoic to Amazonian-West African sources by early Cambrian (Satkoski et al., 2010).

There is, however, evidence suggesting that Avalonia in the early Neoproterozoic might not have included all the present-day Avalonian terranes, and that some crustal blocks may have originated elsewhere. In west Avalonia, in particular, there are areas that remain problematic and present important differences in their Neoproterozoic geological histories. Such an area is the Avalon Zone of SE New England. Revised geochronology of the granitoid rocks of SE New England revealed that the main Avalonian arc phase in the area was rather short-lived and spanned between 610 and 590 Ma (high precision U-Pb dating on zircons, Thompson et al., 2010). This episode of arc-related magmatism broadly resembles the timing of intermediate and felsic magmatism in the Bass River Block of the Cobequid Highlands (609 ± 4 Ma Debert River, and 603 Ma McCallum Settlement plutons, MacHattie and White, 2012). The absence of older felsic rocks in SE New England, such the ca 630 Ma rhyolites of the Jeffers Group and the ca 620 Ma

Age (Ma)	Slat	Slon	D	Ι	(°) 300	Plat	Plon	(°) db	(₀) mp	Rock unit	Reference
552	50.2	293.5	105.4	82.1	5.1	-44	134.5	9.7	6.6	Sept-Iles dykes and baked contacts	Tanczyc et al., 1987
565	50.2	293.5	333	-29	8	20	141	5	6	Sept-Iles layered intrusion	Tanczyc et al., 1987
571	54	300.5	110.8	50.1	11.7	-13.7	176.2	10.5	15.7	Double Mer formation	Murthy et al., 1992
575	46.2	280.6	82.2	82.7	3.1	-46.3	121.4	5.9	6.1	Callander alkaline complex	Symons and Chiasson, 1991
581	46	282	316	75	6	60	240	14	14	Grenville diabases	Murthy, 1971
600	54.5	301.5	331	49	6	7	146	8	11	Pre-Grenvillian mafic	Park and Gower, 1996
615	45.6	283.4	135	30	8.4	16	148	5.2	9.3	intrusions Cheneaux metagabbro	Seguin and Brun, 1984
615	53.7	303.4	279.1	28.1	7.3	17.4	215.2	4.4	8	Long Range dykes	Murthy et al., 1992
670	45.6	284.6	288.7	-49.3	7.3	-9.5	160.8	6.5	9.7	Buckingham volcanics	Dankers and Lapointe, 1981
725	44.3	283.6	315	-15	5	24	154	2.6	5.2	Kingston dyke 4	Irving, 1972
750	46.1	285.6	114	38	10	0	164	6	12	Morin anorthosite complex	Irving et al., 1974
Notes:	Slat an	d Slon ar	e presen	lt-day g€ mameti	sographic c inclinati	coordina	ttes of sar	mple locat	tion, Plat an fidence elli	d Plon are present-day coordina	ttes of the paleopoles, D= mean

Laurentian paleopoles used for the Neoproterozoic reconstructions presented in this work. Table 5.1

3 n 1 ipac, up 1, u95⁻ τ<u>ό</u> magnetic declinauou, i confidence ellipse, respectively. Georgeville Group (Murphy et al., 1997) was taken by Thompson et al. (2010) to imply either different histories between these areas or different crustal levels. Thus, the chronological similarities in arc-sequences between SE New England, and parts of the Cobequid Highlands, set them apart as distinct domains within the Avalonian collage (Thompson et al., 2010).

5.2.3 Comparison between oldest Neoproterozoic units in west Avalonia

Knowledge of the Avalonian basement is rather limited due to its sparse exposure (Murphy and Nance, 1991). Indirect evidence for the nature of this basement has been obtained through isotopic characteristics of igneous rocks derived either from crustal anatexis or the lithospheric mantle. In the following sections, published stratigraphic, isotopic, and whole-rock geochemical data for Neoproterozoic west Avalonia will be further evaluated and compared in order to understand the early stages of crustal accretion.

In the Cobequid Highlands, the Neoproterozoic Bass River and Jeffers blocks are separated by a crustal-scale fault. The Rockland Brook fault (RBF) is interpreted to be a reactivated Neoproterozoic fault (Miller, 1991; Pe-Piper and Piper, 2003) and consists of two parts: a W-E trending section and a NE-trending segment. The NE-trending western part is younger and records mostly late Paleozoic motion, whereas the eastern part of the fault is older (ancestral Rockland Brook fault of Miller, 1991), and considered to represent a steep, parallel-walled ductile shear zone (Miller, 1991).

Stratigraphically, the Bass River and Jeffers blocks show important differences in their Precambrian basements (Fig. 5.2). The oldest unit in the Bass River Block is the metasedimentary ca 1 Ga Gamble Brook Formation (Barr et al., 2012). Although the

depositional age of the Gamble Brook Formation is placed around 1 Ga, Archean zircons (ca. 2.8 Ga) have also been reported (Barr et al., 2003). In the absence of volcanic beds, however, the age of the Gamble Brook Formation is based on youngest detrital zircon



Figure 5. 2 Distribution of mapped geological units with time between the Bass River and the Jeffers blocks of the eastern Cobequid Highlands (data compiled from MacHattie and White, 2012). Abbreviations: CARB.= Carboniferous, DEV.= Devonian, SIL.= Silurian, ORD.= Ordovician, CAM.= Cambrian, EDI.= Ediacaran, PM= Polson Mountain pluton, GB= Gain Brook pluton, MCS= McCallum Settlement pluton, DR= Debert River pluton, FGL= Frog Lake pluton, WP= Wentworth pluton, HG= Horton Group, FLG= Fountain Lake Group, EMB= Eight Mile Brook pluton, GSB= Gunshot Brook pluton, SMB= Six Mile Brook pluton, ME= Mount Ephraim pluton.

dated, and therefore must be treated with caution. The Gamble Brook Formation, which

has been metamorphosed under greenschist conditions, is interpreted to have been

originally deposited within a rifted island-arc setting based on the geochemistry of the sediments and the presence of contrasting sources (Murphy, 2002). The upper Gamble Brook Formation contains pelitic layers which based on their Sm-Nd isotope composition were assumed to have derived from a juvenile mafic source, whereas the structurally lower quartities were derived from an older cratonic basement. Murphy (2002) associated the juvenile source of the upper pelites with the proto-Avalonian crust. The Economy River Gneiss (734 ± 2 Ma, Doig et al., 1993) is a granodioritic orthogneiss that was interpreted by Murphy et al. (2001) to represent an earlier phase of arc-related magmatism. Even though the Economy River Gneiss was assigned by Murphy et al. (2001) and Pe-Piper and Piper (2005) to the Bass River Block, the unit is located just south of the Pleasant Hills pluton, north of the Cobequid Fault, and in the stepover zone between the Rockland Brook and Kirkhill faults. Therefore its affinity based on its location is unclear (Pe-Piper and Piper, 2003), since there is no obvious spatial relationship with the faults that separate the two crustal blocks, and it is not considered further in this chapter. The Folly River Formation comprises metasedimentary rocks interlayered with volcanogenic sediments and pillow lavas. According to their geochemistry, the lavas were deposited on an ocean floor as within-plate continental tholeiites related to rifting (Pe-Piper and Murphy, 1989; Murphy et al., 2001).

The sedimentary rocks of the Gamble Brook Formation show the development and subsequent folding of a mylonitic fabric, which was interpreted by Nance and Murphy (1990) as the result of a single deformational event related to shearing. The age of this shearing was estimated at ca 640-625 Ma based on synmagmatic deformation in dated granites of the area. This deformational event was not identified in the Folly River

Formation, which in contrast shows polyphase deformation, and thus the rocks were interpreted to be much younger than that of the Gamble Brook Formation. This was based on the observed truncation of the mylonitic fabric in the Gamble Brook Formation by a mafic dyke assigned to the Folly River Formation (Nance and Murphy, 1990). The changes in the mylonitic fabric between the Folly River and the underlying Gamble Brook formations were taken by Murphy et al. (2001) to result from contrasting structural styles, and the contact between the two formations was suggested to be an unconformity. Therefore, the Folly River Formation was thought to be much younger than the Gamble Brook Formation. MacHattie and White (2012), however, argued that the observed change in deformation between the two units is because the Folly River Formation resides at the core of the Bass River Block, away from mylonitic bounding faults. They state that the two formations are not in contact and based on petrographic similarities between the two (metamorphic grade and presence of quartzite beds) these units are lateral stratigraphic facies equivalents.

The oldest unit in the Jeffers Block (Fig. 5.2) is the polymetamorphic 810 ±12 Ma Mount Thom Complex (Donohoe and Wallace, 1982; zircon U-Pb age from MacHattie et al., 2013). The Mount Thom Complex consists of paragneiss, granite gneiss and amphibolite (MacHattie and White, 2012). Even though the tectonic environment of Mount Thom Complex remains controversial, Meagher (1995) concluded that the amphibolites have subalkaline and tholeiitic protoliths, whereas the granite gneisses resemble volcanic arc granites (Pe-Piper et al., 1998). The Dalhousie Mountain Formation of volcanic and volcaniclastic rocks has no geochronological data available. However it is suggested to be of Neoproterozoic age, based on the similarities with

Neoproterozoic formations in the Antigonish Highlands and the fact that is intruded by the ca 605 Ma Gunshot Brook pluton (Donohoe and Wallace, 1982; Murphy et al., 1991; Pe-Piper and Piper, 2003; MacHattie and White, 2012). A Neoproterozoic age for the Dalhousie Mountain Formation is consistent with the presence of the lithologically similar Neoproterozoic Jeffers Group throughout the northern and western Cobequid Highlands (Pe-Piper and Piper, 2003).

Lithostratigraphic correlations have been suggested between the Neoproterozoic fault-bounded blocks in the Antigonish and the Cobequid Highlands. The Jeffers Block of the Cobequid Highlands was correlated with the Keppoch Block, whereas the Bass River Block was correlated with Clydesdale block (Murphy et al., 1991). Therefore, the correlations made between the Neoproterozoic units of Antigonish and the Cobequid Highlands may indicate the presence of two Avalonian fault-bound blocks, the Jeffers-Keppoch and the Bass River-Clydesdale that can be traced throughout mainland Nova Scotia (Fig. 5.3).

Murphy and Dostal (2007) argued, on the basis of consistent isotopic characteristics in Avalonian igneous rocks through time, that Avalonian crustal fragments remained attached to their lithospheric mantle from Neoproterozoic to Devonian. In this case, the isotopic signature of the Neoproterozoic basalts should record systematic differences in mantle sources if they originated in entirely different microplates. Published Nd-Sm and Pb isotope data (Pe-Piper and Piper, 1998b) indicate that there are significant and systematic differences between the Jeffers and Bass River blocks. The $\epsilon_{Nd(t)}$ values of late Neoproterozoic (ca 650 Ma) mafic rocks from these blocks show strongly positive values for the Jeffers Block (+ 4.5 to + 5.7), whereas their equivalents in the Bass River Block, range from +2.5 to -0.1 (Pe-Piper and Piper, 1998b). These values, when plotted on the ε_{Nd} evolution diagrams for Avalonia (Nance and Murphy, 1996), show that the Neoproterozoic gabbroic rocks of the Jeffers Block have a definite west Avalonian signature, whereas those from the Bass River Block plot in the overlapping area between west Avalonia and the envelope defined for the Tocantins Province of the Amazonian craton (Fig. 5.4), a part of Gondwana that has been interpreted to be proximal to Avalonia in the Neoproterozoic, and source of detritus (Murphy et al., 2001; Barr et al., 2012). This isotopic diversity could be interpreted either as a) reflecting the presence of Tocantins basement rocks, or b) derivation from a source similar to that of Tocantins Province. The Neoproterozoic gabbros of the Bass River Block have a Pb isotopic signature chemically resembling the Kerguelen oceanic basalts, for which an enriched mantle component is interpreted (Zindler and Hart, 1986) (Fig. 5.5). Hawkesworth et al. (1986) correlated the EM II reservoir with the continental lithosphere of South America and Africa cratons in Gondwana.

Pb isotope data are not available for the Neoproterozoic gabbros from the Jeffers Block. Systematic differences in whole-rock geochemistry between the mafic rocks of the two crustal blocks, however, may indicate contrasting mantle sources. The Bass River Block Neoproterozoic gabbros show systematically higher concentrations of REE (Fig. 5.6A) and Nb (Fig. 5.6B) than the gabbros of the Jeffers Block, suggesting a larger EM II component (Zindler and Hart, 1986). The Bass River Block gabbros have lower Sr (Fig. 5.6C) and higher Zn (Fig. 5.6D) and TiO₂ (Fig. 5.6E) concentrations than those from Jeffers Block. In the Mn-Ti-P ternary diagram, the basalts of the Folly River Formation plot clearly within the MORB field (Fig. 5.7A), and in the Ti-Zr-Sr diagram have a



lapetan arcs are after Hibbard et al. (2006). Abbreviations: J= Jeffers Block, B= Bass River Block, K= Keppoch Block, C= Clydesdale Block, D= Figure 5.3 Simplified geological map of the Northern Appalachians (modified after Pe-Piper and Piper, 2003 and Barr et al., 2012). Positions of Dunn Point Formation.



Figure 5. 4 Neoproterozoic mafic rocks from the Bass River and Jeffers blocks (data from Pe-Piper and Piper, 1998b, ages modified after MacHattie and White, 2012) plotted on ε_{Nd} evolution curve diagrams for west Avalonia, Grenvillem and Tocantins Province (fields after Nance and Murphy, 1996).



Figure 5. 5 ¹⁴³Nd/¹⁴⁴Nd vs ²⁰⁶Pb/²⁰⁴Pb variation diagram for oceanic basalt suites (modified from Zindler and Hart, 1986) with positions for the Neoproterozoic gabbros from the Bass River Block. DMM= depleted mantle, PREMA= prevalent mantle, HIMU= high- μ mantle, FOZO= undegassed (volatile-rich) mantle. Red lines represent the modelled trends for isotopic mixing between selected end-members. Numbers along lines refer to the relative proportions of one end-member.

definite ocean floor affinity (Fig. 5.7B) which is different from the ca 620 Ma Neoproterozoic arc-related basalts from the same crustal block, suggesting probably an origin in a different tectonic environment.

The systematic differences in trace elements between the ca 620 Ma arc-related gabbros from the Bass River and Jeffers blocks may reflect substantial differences in their lithospheric mantle sources that could indicate derivation from different lithospheric fragments. Furthermore, the rift affinity of the Folly River basalts correlates more with the tectonic setting of the Gamble Brook Formation, rather than the arc-related ca 620 Ma gabbros from the same block.

Compared to west Avalonian rock units elsewhere, the basement of the Bass River Block (Gamble Brook and Folly River formations) correlates in lithologic character and metamorphic grade only with the oldest Avalonian rocks of SE New England, namely the ca 1 Ga Westboro Formation and the 1170 Ma Blackstone Group (Barr et al., 2012; Thompson et al., 2012). The Blackstone Group contains metapelites, quartzites and metavolcanic rocks of greenschist facies, notably pillow basalts, resembling the lithological character of the Folly River Formation. The Westboro Formation is overlain by the Middlesex Fells complex which contains basaltic flows and tuffs with transitional to alkalic compositions (Cardoza et al., 1990). Comparison between the basalts of the Folly River Formation and the Middlesex Fells complex reveal that they are chemically similar and quite distinct from the Jeffers basalts (Fig. 5.8A to E). The Mesoproterozoic Westboro and Blackstone formations of SE New England have been interpreted as crustal fragments of a pre-Avalonian continental margin (Bailey et al., 1989). The stratigraphic, chronological and chemical similarities between the Avalonian Folly River and the

Westboro and Middlesex formations could imply not only that a) the Neoproterozoic basements of SE New England and Bass River Block are probably related, but also that b) the Folly River Formation, the age of which is still controversial, resembles known early Neoproterozoic Avalonian basement elsewhere, rather than late Neoproterozoic. Comparison of U-Pb ages from detrital zircons from the Westboro and Gamble Brook formations reveals that the two formations share similar distributions in zircon age populations (Table 5.2), resembling ages from Amazonian source belts (Thompson et al., 2012).

5.2.4 LITHOSTRATIGRAPHIC CORRELATIONS BETWEEN THE BASEMENT UNITS OF BASS RIVER AND SE NEW ENGLAND WITH AMAZONIA

The Bass River Block has a distinct isotopic signature that resembles the Tocantins structural province (Fig. 5.4), and contains detrital zircons interpreted to come from Amazonian sources. The understanding of the Precambrian evolution of South America is, therefore, important if any correlation with the Avalonian blocks of the Cobequid Highlands is to be attempted.

Structurally, the present-day South American platform has been divided into cratons that represent older crustal fragments of pre-existing microplates and structural provinces that surround the cratons (de Brito Neves and Fuck, 2013). The structural provinces closest to the Amazonian craton are the Borborema, Tocantins, and Mantiqueira provinces, which are orogenic belts. The cratonic fragments were initially brought together during the assembly of Rodinia by subduction systems known as Tonian orogenic belts which are restricted to the cratonic basements (850-1000 Ma) (de Brito Neves and Fuck, 2013). In the late Neoproterozoic, the Brasiliano orogenies, which developed along a branching system of subduction zones, are characterized by 650-600



Figure 5. 6 Chondrite-normalized REE patterns (normalizing values after Sun and McDonough, 1989) (A) and selected geochemical diagrams (B to E) that demonstrate best the chemical differences between the Neoproterozoic mafic rocks of the Bass River and Jeffers blocks.



Figure 5. 7 Trace element classification diagrams after (a) Mullen (1983) and (b) Pearce and Cann (1973) for the Neoproterozoic mafic rocks of the Cobequid Highlands.

Ma arc sequences in all provinces and record substantial crustal reworking (de Brito Neves and Fuck, 2013). The distal parts of the provinces are compositionally diverse, and present evolved volcano-sedimentary arc sequences, whereas the marginal belts that surround the cratons are dominantly quartzite-pelite-carbonate assemblages (de Brito Neves and Fuck, 2013). The basement of the Bass River Block (Gamble Brook Formation) is broadly similar in age and stratigraphy to the Tocantins Group (lower quartzite unit and upper chlorite schist) of the Araguaia belt present only in the Tocantins province (de Brito Neves and Fuck, 2013; Goodwin, 2013).

The stratigraphic and chemical similarities discussed above between the Neoproterozoic basements of the Bass River Block and the Avalon Zone of SE New England (Table 5.2) suggest that these parts of the Avalon composite terrane either a) reflect different crustal levels of the Avalonian crust; or b) comprise a distinct unit of Avalonia that did not originate within the same microcontinent of proto-Avalonia, but



Figure 5. 8 Harker-type geochemical plots for the Neoproterozoic mafic rocks of the Cobequid Highlands and SE New England (A to E). AFM classification diagram (F) after Irving and Baragar (1971). Chemical data for SE New England compiled from Cardoza et al. (1990). have an Amazonian affinity. Given the substantial variations in isotopic signatures and

paleomagnetic data between these areas and the rest of Avalonia, the preferred

interpretation followed in this chapter is that of a different origin. For that reason, these

two areas are plotted as separate fragments along the Amazonian margin in the Neoproterozoic reconstructions presented here. The Jeffers Block, on the other hand, shows isotopic and stratigraphic similarities with the Antigonish Highlands, which are taken as representative of typical west Avalonian crust (Fig. 5.3) (Murphy et al., 1989). The Jeffers Block, along with the Antigonish Highlands, records substantial arc activity in the late Neoproterozoic, which could have been related to the closure of the Brasiliano Ocean. These areas may have been, therefore, parts of a proto-Avalonia drifting in the Brasiliano Ocean towards the Amazonian margin of Gondwana in the Neoproterozoic.

Proto-Avalonia and other peri-Amazonian fragments were probably brought together in the late Neoproterozoic. Paleomagnetic data from the ca 590 Ma Lynn-Mattapan Formation in SE New England indicate that this part of Avalonia occupied a paleolatitude of ~ 30°S (Thompson et al., 2007), which according to the reconstructions presented here, corresponds to the Amazonian margin (Fig. 5.1C) and not to the West African craton as previously proposed. Similar paleolatitudes have been reported from Newfoundland, at ca 580 Ma (McNamara et al., 2001), suggesting that proto-Avalonia approached the Amazonian craton at the time. The significantly shorter phase of arcrelated magmatism in SE New England may record the final collision of the Amazonian margin with proto-Avalonia at 610-590 Ma. The early convergence between proto-Avalonia and these peri-Amazonian fragments is indicated by differences between these Avalonian fragments, which are restricted only to their earliest Neoproterozoic units. Whether the fragments of SE New England and Bass River Block joined proto-Avalonia as one coherent piece and separated later, however, cannot be determined.

5.3 EARLY PALEOZOIC AVALONIA: DRIFTING BETWEEN TWO OCEANS

Paleozoic reconstructions that involve the assembly of Gondwana and later Pangaea follow the most up-to-date models of Torsvik et al. (2014), which relate the movements of the plates to deep mantle structure. The lowermost mantle is dominated by two large low shear-wave velocity provinces beneath present-day Africa and the Pacific Ocean. It is argued that in the last 300 Ma, large igneous provinces and kimberlites have been emplaced directly above the margins of these mantle provinces, in plume generation zones (Torsvik et al., 2014). Correlation between the large igneous provinces and kimberlites with the plume generation zones, at the time of their formation, was used by the authors as a reference frame for their reconstruction models. Their published Euler rotation parameters are followed, with the exception of Avalonia and Ganderia, whose movements have been modified based on published paleomagnetic data and other geological constraints (Appendix D.1).

By early Cambrian (ca 540 Ma) West Gondwana had been assembled and separated from Laurentia, Greenland and Baltica by the Iapetus Ocean (Cawood et al., 2001). Although a high southern paleolatitude is accepted for the Amazonian margin of Gondwana (Murphy et al., 2010), the position of the peri-Gondwanan terranes proposed here, is different from that proposed by these authors. Following the late Neoproterozoic reconstructions, the early Paleozoic position of Avalonia is predicted here along the NW margin of the Amazonian craton, with west Avalonia being the most proximal while east Avalonia is relatively closer to the West African craton, both facing the Iapetus Ocean (Fig. 5.1D). Ganderia is placed outboard of west Avalonia along the Amazonian margin.

				Bass River Bl	ock- Cobequid	Jeffers Block- Cobequid
Affinity	Amazonia	Avalon SE New	v England	High	lands	Highlands
				Gamble	Folly River	
Unit Name	Tocantins Group	Westboro	Blackstone	Brook Fm.	Fm.	Mount Thom Complex
Age	ca 1.3-0.9 Ga	ca 1 Ga (inferred age)	ca 1.1 Ga	ca 1 Ga	not	ca 810 Ma (U-Pb)
			(inferred age)	(inferred age)	determined	
Petrographic description	upper meta-pelite,	pelite, quartzite,	pelite, quartzite,	Upper pelite-	pelite,	paragneiss, granite gneiss,
	lower quartzite,	overlain by basaltic	pillow lavas	lower	quartzite,	amphibolite
	pillow basalts	flows (Middlesex		quartzite	pillow lavas	
		Fells Fm.)				
Metamorphic grade	Greenschist facies	Greenschist facies	Greenschist	Greenschist	Greenschist	Amphibolite facies
1			facies	facies	facies	
Tectonic environment	Passive margin	continental rifting	continental	rifted island	Ocean floor-	volcanic arc
			rifting	arc	rifting	
Description of igneous	not determined	Transitional-alkali	transitional-		MORB basalt	calc-alkaline, tholeitic
rocks		basalt	alkali basalt			
Detrital zircon populations	not determined	1200, 1500, 2100 Ma	not determined	1200, 1500,	not	not determined
(peaks)				2100, 2800	determined	
				Ma		

Table 5.2 Summary of the main characteristics for the discussed Neoproterozoic geological units in west Avalonia.

Murphy et al. (2010) favored placing Avalonia along the SE margin of Amazonia, which rotated clockwise by 460 Ma, bringing Avalonia opposite the Laurentian margin. The reconstructions presented here place Avalonia facing the Laurentian margin since Cambrian times, although that is not paleomagnetically constrained. This position is used, however, since a clockwise rotation of Amazonia was not predicted in the Neoproterozoic-Early Paleozoic models. This position nonetheless favors a later drift of Avalonia that can accommodate an oblique collision with Laurentia before the formation of Pangaea.

Current paleotectonic models suggest that after the formation of Avalonia along the margin of West Gondwana, the terrane was separated from the latter during the opening of the Rheic Ocean (Fig. 5.9A and B) (Murphy et al., 2001). Although there is considerable debate with regards to the time of rifting, it is generally accepted that it took place between the late Neoproterozoic and Silurian. Landing (2005) supported a late Neoproterozoic separation of Avalonia, based on the differences in Cambrian fauna and stratigraphic sequences between Newfoundland and Morocco. Isotopic and paleomagnetic constraints, however, suggest that Avalonia occupied a high southerly latitude close to Gondwana from late Neoproterozoic to early Ordovician (van Staal et al., 1998; Pisarevsky et al., 2012).

Early Ordovician rifting of Avalonia is indicated by a major change in the tectonic regime, reflected in the detrital zircons of early Paleozoic Avalonian platformal sequences and the presence of Ordovician rift-related volcanic rocks in England (van Staal et al., 1998), and in southern Mexico, along the southern margin of the Rheic Ocean (Keppie et al., 2008). Pollock et al. (2009) reported that the sedimentary sequences of the

main Avalonian arc phase (650-600 Ma) in Newfoundland contain large populations of coeval zircons suggesting their deposition as first-cycle sediments, locally derived from



Figure 5. 9 Early Paleozoic reconstructions showing (A) rifting of Ganderia in early Cambrian followed by the separation of Avalonia from west Gondwana in early Ordovician (B). Figure (C) shows the possible arrangement of Avalonia and Ganderia as they drifted together during the closure of Iapetus Ocean.

igneous rocks in the surrounding arc. The younger platform sedimentary sequences, on

the other hand, contain large populations of Mesoproterozoic and Paleoproterozoic

zircons (1.6-2.3 Ga). The presence of Mesoproterozoic zircon was considered consistent

with the presence of a Mesoproterozoic Avalonian basement, although no rocks of such age are exposed but the age is inferred from isotope data of crustal rocks. Based on the absence of a Paleoproterozoic Avalonian crust, Pollock et al. (2009) argued that the source of such zircons was from an older crustal source external to Avalonia. An alternative hypothesis for the origin of these zircons would be the derivation from Avalonian basement rocks with xenocrystic zircons such as those reported from Mira terrane (Bevier and Barr, 1990).

Avalonia is suggested to have amalgamated to a composite Laurentian margin as a distinct microcontinent, separated from Ganderia, in the Silurian (Murphy et al., 2008). The two terranes separated from West Gondwana as distinct domains in early Paleozoic (Pollock et al., 2012). The identical estimated drift rates between Avalonia and Ganderia between 490 and 460 Ma suggest that the two were on the same microplate after they separated from Gondwana (Thompson et al., 2010; van Staal et al., 2012). The contrasting geological histories between the two terranes are attributed to a narrow intervening oceanic seaway in the Ordovician (Fig. 5.9C) (van Staal et al., 2012).

MacNiocaill et al. (1997), on the basis of paleomagnetic and faunal data from the Appalachian Orogen, identified several early to middle Ordovician arcs within Iapetus Ocean: a) a peri-Laurentian arc at 10-20°S, b) an intra-oceanic arc at 30°S, and c) a peri-Avalonian arc at ~50-60°S. Hibbard et al. (2006), however, placed the peri-Gondwanan arc systems outboard of the northern Gander margin. Ordovician magmatism in west Avalonia is almost exclusively alkaline and extension-related, with no reports of arcrelated activity. Back-arc sequences, however, are well preserved in the eastern part of Avalonia (van Staal et al., 1998), while of limited extent in its western portion (Hamilton and Murphy, 2004). An extensional tectonic environment for Ordovician magmatism could be equally related to either back-arc extension, since Iapetus is thought to have been subducted beneath the Laurentian margin (Torsvik and Rehnström, 2003), or rifting associated with the opening of Rheic Ocean, south of Avalonia (Fig. 5.10).

5.3.1 CORRELATION BETWEEN ORDOVICIAN MAGMATISM AND PLATE TECTONICS

In order to evaluate the relative positions of the Avalonian blocks in the early Paleozoic it is important to compare the nature of magmatism throughout the microcontinent, since it may be related to geodynamics. While Avalonia occupied a position between the closing Iapetus and the spreading Rheic oceans from Ordovician to Silurian, this time interval is characterized by limited magmatism in west Avalonia (Pollock et al., 2009). Ganderia, on the other hand, records substantial arc-related magmatism during middle Ordovician (Moench and Aleinikoff, 2003). The Ordovician marks an important change in the nature of Avalonian magmatism. It is during that period that the first alkaline magmatic events are recorded in west Avalonia, with A-type affinities, which set them apart from the Precambrian arc-related magmatism. Ordovician alkaline magmatism, however, is not synchronous throughout west Avalonia. Early Paleozoic A-type magmatism is observed in the Jeffers Block of the Cobequid Highlands (481.6 ±4.1 Ma, U-Pb, Eight Mile pluton, MacHattie et al., 2013), in the Antigonish Highlands (469.4 ± 0.5 Ma- Brora Lake Haggarts Lake, West Barneys River, McGraths Mountain, and Leadbetter Road plutons, 460 ± 3.4 Ma- Dunn Point Formation, 454.5 ± 0.7 Ma- McGillivray Formation), and in northeastern mainland Nova Scotia (478± 3 Ma, Cape Porcupine Complex) (White et al., 2002; Hamilton and Murphy, 2004; MacHattie and White, 2012; Escarraga et al., 2012; Barr et al., 2012).

Middle Ordovician



Figure 5. 10 Simplified cross-section of the Ordovician rifting of Avalonia during the closure of Iapetus Ocean and the opening of the Rheic Ocean (modified from van Staal et al., 1998).

Ordovician arc-related magmatism has been reported in the fault-bounded Putnam-Nashoba terrane of SE New England (Fig. 5.3), where it has been interpreted to reflect the closing of an ocean separating Avalonia and Laurentia (Acaster and Bickford, 1999). This block, however, has also been described as the trailing edge of Ganderia (Wintsch et al., 2014). The middle Ordovician (460 ± 3.4 Ma) volcanic rocks of the Dunn Point Formation in the Antigonish Highlands (Fig. 5.3), have a paleolatitude of ~40°S (Fig. 5.11A) (Johnson and van der Voo, 1990). Hamilton and Murphy (2004) interpreted these rocks, based on their geochemistry and bimodal character, to reflect a rifted arc tectonic setting and placed this rifted arc on the Avalonian microcontinent, outboard from Laurentia, within Iapetus.

In the Avalon zone of SE New England, Ordovician alkaline intrusions include the Cape Ann, Peabody, and Quincy granites, while the bimodal Nahant alkaline suite was intruded between ca 490 and 488 Ma (Buma et al, 1971; Thompson et al., 2010). The early Paleozoic alkaline magmatic events, where present in west Avalonia, were broadly synchronous with rifting of Avalonia from West Gondwana. A-type granites are related to extensional tectonic regimes (Loiselle and Wones, 1979), and therefore, could be related to the opening of the Rheic Ocean in the early Ordovician. The coeval mafic rocks show some distinct variations in their trace element characteristics. Mafic rocks from Cape Porcupine (Fig. 5.3) have negative Nb and Ti anomalies relative to MORB (Fig. 5.12A) (Barr et al., 2012), which could indicate derivation close to a volcanic arc (Xu et al., 2000). Mafic rocks from both the Antigonish Highlands and SE New England show positive Ti and Nb anomalies relative to MORB (Fig. 5.12B and C). The early Ordovician Nahant Gabbro of SE New England, however, is geochemically distinct, showing an additional positive Th anomaly (Fig. 5.12C).



Figure 5. 11 Ordovician-Silurian reconstructions showing (A) the position of Avalonia in mid-Ordovician, and (B) the approach of the microcontinent with Laurentia and Baltica in late Ordovician as indicated by the paleomagnetism in Newfoundland (Hodych and Buchan, 1998). Figure (C) shows the amalgamation of Ganderia and Avalonia by Silurian and the formation of Laurussia following the closure of Iapetus Ocean.

Even though the Ordovician felsic rocks in Avalonia all present A-type alkaline affinities, the earlier initiation of alkaline magmatism in SE New England compared to the rest of west Avalonia, together with a paleolatitude of $65^{\circ}S \pm 7^{\circ}$ (Thompson et al., 2010), could indicate a position in the southern part of Avalonia adjacent to the northern margin of the spreading Rheic Ocean.



Figure 5. 12 MORB-normalized spider diagrams for the Ordovician mafic rocks of west Avalonia as recorded (a) in Cape Porcupine, (b) Antigonish Highlands and (c) SE New England (Nahant Gabbro). Geochemical data compiled from Murphy et al. (1991), Thompson et al. (2010), and Barr et al. (2012).

The above geochemical differences and paleomagnetic data in igneous rocks could imply that in the Ordovician the western part of the Avalonian microcontinent could be divided into three zones with respect to the southern Iapetan margin : a) a proximal zone, either as trailing edge of a Ganderia connected to Avalonia, or as the leading edge of Avalonia, as exposed in Putnam-Nashoba terrane with volcanic arc, calcalkaline characteristics; b) a back-arc zone (~50-60°S), well within the main Avalonian

domain of the Ordovician Avalo-Ganderian connection, characterized by alkaline magmatism with A-type affinities in felsic rocks, as exposed in Jeffers Block, Antigonish Highlands, and Cape Porcupine; and c) the part of the microcontinent adjacent to the trailing edge facing the Rheic Ocean, which is characterized by southerly Ordovician paleolatitudes (>60°S) and alkaline rift-related rocks as exposed in SE New England.

The 490 Ma Avalonia is placed at the northern margin of Gondwana at a paleolatitude of ~60°S between the Amazonian and the African cratons, in these reconstructions (Fig. 5.9B). However, there are some issues to be addressed. A paleolatitude of 65°S for SE New England at the time, as reported for the Nahant Gabbro, should position Avalonia along the West African margin and adjacent to east Avalonia for which a paleolatitude of $62^{\circ}S \pm 9^{\circ}$ has been reported (Trench et al., 1992). Thompson et al. (2010) argued on that basis that Avalonia was adjacent to the West African craton, and the Nahant paleopole could be used without rotation corrections. In the reconstructions presented here, west Avalonia, including SE New England, is positioned at $\sim 60^{\circ}$ S, still within error from the published paleolatitude. A 65°S paleolatitude for SE New England in this configuration could be explained only by oblique movements bringing that part of the terrane adjacent to east Avalonia. To the author's knowledge no such fault kinematics have been reported for Avalonia so far, however this hypothesis should be tested. The alternative would be that Avalonia at 490 Ma had drifted away from Gondwana and was closer to Baltica; however a change to a Baltic fauna is reported only after the Tremadoc (Pollock et al., 2012). Therefore, SE New England is positioned at the trailing edge of west Avalonia, on the basis of its distinct and longer-lived alkaline magmatism in the early Paleozoic.

5.3.2 VARIATIONS IN THE ORDOVICIAN GEOLOGICAL HISTORIES BETWEEN PARTS OF THE AVALONIAN CANADIAN MARITIMES

The lack of Ordovician igneous rocks in parts of west Avalonia such as the Bass River Block and in Newfoundland makes them difficult to be assigned into zones within the Avalonian microcontinent during that period. Early Silurian (441 ± 2 Ma, U-Pb on baddeleyite) mafic sills of the Avalon Peninsula in Newfoundland, however, show a paleolatitude of $32^{\circ}S \pm 8^{\circ}$ (Fig. 5.11B) (Hodych and Buchan, 1998), suggesting that either: a) this part of Avalonia was closer to Ganderia and the Iapetan margin than SE New England and the Antigonish Highlands; or b) that Avalonia had drifted to that paleolatitude by early Silurian. Based on the differences in paleolatitudes of geological units of different ages within a paleocontinent, the drift rates may be estimated. Avalonia drifted from Gondwana towards Laurentia between Cambrian and middle Ordovician at a rate of 9.2 cm/a (Thompson et al., 2010), based on paleomagnetic data from Nahant. These rapid drift rates were interpreted as the effect of slab rollback beneath the leading edge of Ganderia during the closure of Iapetus Ocean (van Staal et al., 2012). Taking into consideration a Laurentian paleolatitude of ~20 °S between 460 and 420 Ma and the paleolatitude of the coeval Dunn Point Formation, Hamilton and Murphy (2004) argued that the approach to of Avalonia to Laurentia from middle Ordovician to early Silurian was characterized by slower drift rates of 5.5 cm/a. The decrease in drift rates close to the Silurian was considered, the result of the opening of the Tetagouche-Exploits back-arc basin and the docking of Ganderia to Laurentia (van Staal et al., 2012). Taking into account these drift rates and back-calculating the distance covered during the drift, a 441 Ma paleolatitude of 32° in the Avalon Peninsula would correspond to a paleolatitude of 41°S at 460 Ma, which is similar to that of the middle Ordovician Dunn Point Formation

of the Antigonish Highlands. Therefore, the Avalon Zone of Newfoundland should have been located further north than SE New England, within the Avalonian microcontinent, but possibly at the same latitude as the Antigonish Highlands (Fig. 5.11), as originally proposed by Hodych and Buchan (1998).

5.3.3 SILURIAN ACCRETION OF AVALONIA TO LAURENTIA, RELATED MAGMATISM AND MANTLE METASOMATISM

Silurian magmatism in west Avalonia reflects subduction during the final stages of the closure of Iapetus Ocean. The Pass Island granite in SW Newfoundland, just south of the Hermitage fault (Fig. 5.3), has yielded a Silurian age of 423 ± 4 Ma (SHRIMP U-Pb on zircon, Kellett et al., 2014), with trace-element characteristics of volcanic-arc and syn-collisional granites (Kerr et al., 1995), which could be linked to the collision of Avalonia with the Laurentian margin (Fig. 5.11C). In SE New England, the Silurian Franklin pluton (ca 417 Ma) presents alkaline post-orogenic characteristics (Hermes and Zartman, 1992), whereas in Cape Breton, Silurian arc magmatism in the Creignish Hills (Fig. 5.3) was interpreted to represent subduction along the northern margin of the Rheic Ocean (Keppie et al., 2000). In the Cobequid and Antigonish Highlands, however, no Silurian igneous rocks have been reported, although sedimentary clastic sequences of that period are present. In the Cobequid Highlands, the Silurian shales, siltstones and sandstones of the Wilson Brook and Portapique River formations are known only within the Jeffers Block, and are considered identical with the sequences of the Arisaig area in the Antigonish Highlands (Murphy et al., 1996; Pe-Piper and Piper, 2003). These sedimentary successions are quite different from the Devonian-Carboniferous Horton Group, which commonly contains conglomerate and lacustrine deposits, suggesting terrestrial basins (Pe-Piper and Piper, 2003). On the basis of whole-rock geochemistry,

isotopic data and paleocontinental reconstructions, Murphy et al. (1996) argued that the source of the Silurian strata in the Antigonish Highlands is probably eastern Laurentia or Baltica.

Murphy et al. (2004) argued that the late Ordovician-early Silurian sediments of the Arisaig Group in the Antigonish Highlands do not contain detrital zircons of Avalonian origin. All studied sedimentary rocks of that rock unit are characterized by strongly negative ε_{Nd} values (-4.8 to -9.3) and T_{DM} ages greater than 1.5 Ga. In the nearby Cobequid Highlands, the comparative study of Pe-Piper and Piper (1998b) demonstrated that this isotopic signature is not observed in the younger igneous rocks of the area, while Pb isotopes in the late Paleozoic felsic rocks show a strong Neoproterozoic component. This is consistent with subduction of terrigenous sediment in Neoproterozoic, but predominantly ocean crust in the early Paleozoic. Furthermore, Murphy et al. (2008) concluded on the basis of trace element geochemistry that the Ordovician mafic rocks in the Antigonish Highlands show only a minor influence from a crustal component. Thus, there is no evidence across mainland Nova Scotia of mantle metasomatism in the Ordovician, despite the subduction that was presumably required for the closure of Iapetus Ocean.

5.4 LATE PALEOZOIC WEST AVALONIA: SHEARING AND MAGMATISM

After the collision of Ganderia with the Laurentian margin during the Salinic orogeny in Silurian, the subsequent collision of Avalonia and Meguma terranes marked the onset of the closure of the Rheic Ocean during the late Silurian-early Devonian Acadian orogeny (van Staal et al., 2009). The 423-416 Ma arc sequences along the trailing edge of Ganderia and coeval HP-LT metamorphism east of that arc, as well as the

development of early Devonian NW-directed shear zones and fold belts in Ganderia all suggest that the collision of Avalonia was oblique and had a westward polarity (Keppie, 1993, Zagorevski et al., 2007; Dolmeier and Torsvik, 2014). Hibbard and Waldron (2009) hypothesized late Paleozoic strike-slip motion along the length of the Appalachian Orogen on the basis of misalignment of the Hermitage flexure in Newfoundland with the Laurentian margin and crustal thinning beneath the Gulf of St. Lawrence. The scale of this orogen-parallel displacement, from Devonian to early Mississippian, was estimated to be \sim 220-250 km, and was related to the accretion of the peri-Gondwanan Meguma terrane (Hibbard and Waldron, 2009). Intra-continental deformation can be interpreted as far-field response to plate tectonics and is associated with the reactivation of pre-existing crustal lineaments (Murphy et al., 1999). The oblique collision of Avalonia during the Acadian orogeny after the oblique convergence with Gondwana in the Neoproterozoic, could be related to the reactivation of pre-existing paths of weakness such as W-E trending faults like the Kirkhill and the Rockland Brook faults in the Cobequid Highlands.

The synchronous initiation of igneous activity in west Avalonia, and especially the voluminous mafic volcanism in mainland Nova Scotia (Dessureau et al., 2000) with the late Paleozoic strike-slip tectonics, could be evidence of edge-driven mantle convection (Anderson, 1998). Edge-driven convection theory states that in mature cooling planets the mantle-core boundary is a highly conductive cooling layer. According to Archimedes principle parts of the cooling layer will sink into the interior and displace older material underneath. As a result, mantle updrafts are created and are expressed as hotspots in the interior of plates (Anderson and Natland, 2014). A similar scenario is

proposed for the upper mantle underneath drifting plates. Colder mantle that has interacted with the lithosphere in subduction zones will sink into the deeper mantle. The hot mantle updrafts in this case will be expressed as voluminous mafic volcanism with asthenospheric chemical characteristics (Anderson and Natland, 2014). Such excess volcanism is focused at major lithospheric discontinuities, such as cratonic boundaries and continental margins, during rapid lithospheric extension. Regionally, the thickest late Paleozoic basalt succession, with primitive asthenospheric chemical signature (Dessureau et al., 2000), is found along the Rockland Brook in the Cobequid Highlands (Diamond Brook Formation). This volcanism may provide further evidence that this tectonic segment could reflect a deep crustal-scale discontinuity, probably an early Neoproterozoic suture, as discussed earlier in this chapter.

Pre-existing crustal heterogeneities can profoundly influence the lithosphere's response to subsequent deformation. This tectonic inheritance is well documented in cratons, where long-lived lineaments originated as Archean shear-zones (Butler et al., 1997). Repeated episodes of displacements along these faults generate highly heterogeneous kinematics in their sub-structure, the rheology of which will try to adapt to different conditions of temperature, strain and stress of each deformational phase (Butler et al., 1997). Experimental work of Henza et al. (2011) demonstrated that in a sequence of deformational events: a) reactivation of pre-existing faults will occur only if the first phase of deformation is well-developed; and b) the pre-existing faults act as obstacles for the propagation of younger faults. The longevity of the Rockland Brook fault (and possibly Kirkhill fault as its westward continuation) could therefore be indicative of the

presence of a significant crustal scale lineament in the Neoproterozoic that affected the Avalonian continental lithosphere until 300 Ma later.

5.4.1 COMPARISON BETWEEN THE LATE PALEOZOIC A-TYPE GRANITES IN WEST AVALONIA

Devonian peralkaline granites with either sodic or sodic-calcic amphibole, are also found in SE New England (Quincy Granite) and in the Antigonish Highlands (West Barneys River and Brora Lake plutons) (Sayer, 1974; Escarraga et al., 2012). The Antigonish peralkaline granites have the strongest negative Eu anomalies and show the highest REE-enrichment compared to the other A-type granites of the area. Furthermore, they present similar isotopic characteristics with the coeval mafic rocks, and a comagmatic relationship between the two has been proposed (Escarraga et al., 2012), consistent with the character of the Wentworth sodic granites. A similar pattern has been described for the early Devonian Quincy Granite in SE New England (Sayer, 1974). The peralkaline granites have been described to have more pronounced negative Eu anomalies and enrichment in Zr, Yb, Hf and Ta relative to other coeval granites of the area (Sayer, 1974). Therefore, throughout west Avalonia, the late Paleozoic peralkaline granitic melts point to a common petrogenetic origin associated with fractionation of coeval mafic melts.

5.4.2 LATE PALEOZOIC HYDROTHERMAL CIRCULATION AND MINERALIZATION

Closure of the Rheic Ocean continued through the Mississippian to Pennsylvanian and culminated in collision between Laurussia and Gondwana during the late Carboniferous Alleghanian orogeny, resulting in the formation of Pangaea (Domeier and Torsvik, 2014). In the Cobequid Highlands, the latest phase of ductile deformation recorded in the plutons occurred at ca 339 Ma (Murphy et al., 2011). During that time,

the master faults of the Cobequid Shear Zone record dextral shear associated with the precipitation of hydrothermal biotite (Pe-Piper et al., 2004). In the early Pennsylvanian (~320 Ma), an intense phase of deformation is evident along the southern and northern margins of the Cobequid Highlands. Structures in these areas indicate transpressional strain and shortening during dextral shear (MacInnes and White, 2004). From that period until late Pennsylvanian, the Cobequid-Chedabucto Fault Zone records mylonitization and focussed mineralization (Murphy et al., 2011).

Although the regional character of Pennsylvanian mineralization in the Cobequid Highlands is predominantly IOCG (MacHattie and O'Reilly, 2009), this thesis demonstrated that it was also associated with hydrothermal circulation of rare metals through fractures. The circulation of these metals was long-lived, and the association of minerals such as thorite and hydroxylbastnäsite-(Ce) with hydrothermal epidote and biotite indicates that the precipitation of these minerals started probably in the Visean, as demonstrated in chapter 4. The sources for REE in these fluids were the A-type granites of the area, with primary enrichment by the presence of various magmatic REE-minerals such as allanite-(Ce) and chevkinite-(Ce) (Papoutsa and Pe-Piper, 2013). The mobilization of REE was probably initiated by complexing with F during the breakdown of amphiboles (Pe-Piper, 2007; Papoutsa and Pe-Piper, 2013). Although mobilization of these metals can be facilitated by the presence of halogens and S, their precipitation can be controlled by the oxidation state of the fluids.

The hydrothermal behavior of the REE metals throughout the Cobequid Highlands revealed two major characteristics of the regional system: a) the chemistry of the fluids and consequently the hydrothermal REE-minerals was greatly influenced by

lithologies in both source and depositional areas such as carbonates and shales of the Horton Group that provided C and As and possibly lamprophyres that provided Cl, F and S, and b) as the chemistry of the fluids changed over time and so did the abundance and variety of the REE-minerals. The presence of carbonates and As-rich REE-minerals points to additional sources for C and As from the sedimentary rocks of the Horton Group. On the other hand, precipitation most REE-minerals, as reported in this thesis, was associated with Fe-rich hydrothermal assemblages that are probably related to the Pennsylvanian IOCG mineralization. Important variations in the distribution of the hydrothermal REE-minerals reveal structural changes related to the evolution of the major faults of the shear zone. The lack of substantial hydrothermal REE-minerals in fractures in the Wentworth pluton indicates that the Rockland Brook fault was probably not active during late Carboniferous REE circulation.

Murphy et al. (2011) argued that the Cobequid-Chedabucto Fault Zone was a metallotect, a large-scale crustal structure that controls mineralization. Micklethwaite and Cox (2004) demonstrated that in ancient fault systems, hydrothermal mineralization is focused is areas of rupture arrest. In this process, when an earthquake occurs on a fault the aftershocks are mainly distributed in areas of stress transfer around the seismogenic fault rupture. Such areas form where older faults are linked with younger segments. This process results in the formation of areas with transient permeability and localized fluid flow (Micklethwaite and Cox, 2004). Such relationships may be present in the Cobequid-Chedabucto Fault Zone. The Bass River magnetite deposit lies south of the Pleasant Hills pluton and adjacent to the area where the older Rockland Brook fault merges with the younger Cobequid fault (Pe-Piper and Piper, 2003; MacHattie and O'Reilly, 2008). The
Londonderry iron deposit is located between the Londonderry and Cobequid faults. The Mount Thom (Cu-Co-Au), Brookfield (Ba), and Bridgeville (Fe) deposits are located along the part of the fault zone where the Cobequid fault merges with the younger Chedabucto fault (MacHattie and O'Reilly, 2008). Thus the correlation between the distribution of known IOCG deposits and areas of potential rupture release, along the Minas Fault Zone, may reveal a genetic relationship between the two. Such a correlation, along with the hydrothermal distribution of REE reported in this thesis, could provide guidelines for REE mineral exploration.

CHAPTER 6 DISCUSSION AND CONCLUSIONS

6.1 POST-OROGENIC VERSUS ANOROGENIC A-TYPE GRANITE SUITES: VARIATIONS IN MAGMA SOURCES AND PETROGRAPHY OF GRANITES

A-type granites belong to a distinct group of granitoids which is characterized by alkaline geochemical character and elevated concentrations of incompatible elements (Loiselle and Wones, 1979). These granitoids are reported from both anorogenic and post-orogenic tectonic environments (Whalen et al., 1986). Examples of anorogenic granites include the Younger Granites in Nigeria (Ogunleye et al., 2005), south India (Ambalavayal granite, Rajesh, 2000), Canada (Sept Iles granites in Quebec, Namur et al., 2011), Brazil (Redenção granite, Oliveira et al., 2009), and Yemen Republic (El-Gharbawy, 2010). Post-orogenic A-type granites, besides from the areas described in this thesis, are reported from several parts of the world such as southern Brazil (Serra da Graciosa Province, Gualda and Vlach, 2006), several intrusions in SE China (Xie et al., 2006), NW China (Karamaili granites, Su et al., 2007), Australia (Lachlan fold belt and New England belt, as reviewed in Sylvester, 1989) and Newfoundland (Topsails suite, Whalen et al., 1996).

Anorogenic A-type magmatism follows a major orogenic event after more than 100 Ma and is not directly related to the after effects of orogeny (Condie, 1991). The emplacement of post-orogenic A-type granites, on the other hand, is mostly observed during the final orogenic stages during continent-continent collision dominated by transcurrent-fault tectonics during accretion of continental blocks or terranes (Bonin et al., 1998), as in the Appalachians and the Central Asian Orogenic Belt (Keppie and Dallmeyer, 1987; Windley et al., 2007). Anorogenic A-type granites are mostly found in the interior of stable areas such as the Amazonian craton and the Arabian Shield (Oliveira et al., 2009; El-Gharbawy, 2010).

In anorogenic environments the dominant variety of A-type granite is the peralkaline type with sodic amphibole or pyroxene. Metaluminous A-type granites with calcic amphibole and biotite are much less common (Rajesh, 2000; Ogunleye et al., 2005; El-Gharbawy, 2010). The higher magmatic temperatures and primitive isotopic signatures of the peralkaline granites are evidence of fractionation from mafic melts, whereas the metaluminous varieties have been interpreted to reflect either less evolved melts or products of crustal contamination (Oliveira et al., 2009; Namur et al., 2011).

A-type granites emplaced in post-orogenic settings include peralkaline, metaluminous and peraluminous rocks (King et al., 1997; Gualda and Vlach, 2006). The metaluminous and peraluminous A-type granites generally dominate over the peralkaline varieties. The metaluminous and peraluminous A-type granites form by melting of crustal rocks (Creaser, 1991) and consequently tend to be more K and LILErich (Bonin et al., 1998; Li, 1998).

In the Cobequid Highlands, the emplacement of mainly metaluminous and peraluminous rather than peralkaline A-type granites in the latest Devonian-earliest Carboniferous constitutes a typical example of post-orogenic A-type magmatism that followed the early Devonian Acadian orogeny. These late Paleozoic granites present distinct geochemical and mineralogical differences which can be observed in A-type granites in similar settings elsewhere, such as the variations in ferromagnesian minerals and trace elements.

6.2 IMPACT OF THE CRUSTAL STRUCTURE ON THE POST-OROGENIC A-TYPE GRANITES

A common feature in several post-orogenic A-type granite suites appears to be the association with major strike-slip shear zones. The association of post-orogenic A-type granites with crustal-scale shear zones suggests that the strike-slip faults in the lower crust provide pathways for a) the mafic melts to substantially interact with the crust, and b) for the produced felsic melts to effectively migrate in upper crustal levels and produce the large variety of A-type granites. Such crustal-scale strike-slip systems have been reported from the Aegean area where the deformational fabric appears to extend to the lithospheric mantle (Endrun et al., 2011). Such crustal structures provide favourable conditions for interaction of the upper mantle with the lower crust and therefore can create a deep crustal hot zone (Annen and Sparks, 2002).

In the Cobequid Highlands, mafic rocks are most abundant in the eastern plutons, where sodic, calcic, and biotite granites are present. Elsewhere, variations in the abundance of mafic rocks also appear to be correlated with the mineralogical types of A-type granites present. The peraluminous Barabazar biotite A-type granites in India were emplaced along the faults of the reactivated South Purulia Shear Zone (Dwivedi et al., 2011). Mafic rocks are very limited (Dwivedi et al., 2011) and biotite granite is the only A-type granite present. Seltmann et al. (2000) reported metaluminous biotite and calcic granites along crustal-scale strike-slip shear zones in Kokshaal Range, Kyrgyzstan, with associated tholeiitic mafic rocks of asthenospheric origin. A large mineralogical variety of A-type granites were reported from southern Brazil, where several post-orogenic A-type granites were emplaced along the Southern Brazilian Shear Belt along with abundant

mafic rocks (Bitencourt and Nardi, 1993). Metaluminous biotite A-type granites are the dominant lithology with lesser calcic granites with edenitic amphibole (Nardi and Bitencourt, 2009). Sodic A-type granites are also present, and occur only as late phases coeval with mafic rocks of alkaline affinity, that are considered cogenetic (Nardi and Bitencourt, 2009). The variety in the mineralogy of A-type granites in post-collisional settings, therefore, may depend on whether there is abundant mafic magma to produce calcic amphibole and sodic amphibole granites.

Besides the necessary lithologies as magmatic sources for the late Paleozoic Atype granites of the Cobequid Highlands, the deep Avalonian crust also had the structure necessary to create the deep crustal hot zone. The crustal-scale faults were active in the late Paleozoic but their origin may be much older than that (Pe-Piper and Piper, 2003). Pe-Piper and Piper (1998b) demonstrated on the basis of radiogenic isotope data that the Neoproterozoic basement north and south of the Rockland Brook and Kirkhill faults has a different signature. Miller (1991) recognized structures along the Rockland Brook fault that suggest it is a Neoproterozoic reactivated fault, which is consistent with the synmagmatic deformation of dated Neoproterozoic plutons of the area (Pe-Piper et al., 1996). As reviewed in chapter 5, this Neoproterozoic lineament separates blocks with differences in stratigraphy (MacHattie and White, 2012) as well as in chemistry.

Comparison between the two Neoproterozoic blocks in the Cobequid Highlands with other parts of west Avalonia, suggests that the Bass River block is similar to the basement of SE New England, whereas the Jeffers Block resembles typical proto-Avalonian crust as exposed in the Antigonish Highlands. The available paleomagnetic data and the nature of Neoproterozoic magmatism in these terranes suggests that SE New

England and the Bass River block may have an Avalonian affinity and did not originate within proto-Avalonia. If this is correct, then the Rockland Brook fault may not have originated as a Neoproterozoic intra-plate fault, but as a suture zone. In the case of the latter, the crustal scale of this lineament not only reflects the boundaries between different Neoproterozoic crustal fragments, but also influenced the deformation in the area millions of years later.

The creation of a deep crustal hot zone soon after the main phase of orogeny could explain the variability in A-type granites that is commonly observed in postorogenic environments. The different processes, such as crustal melting and fractionation of mafic melts that result in the production of felsic melts (Annen and Sparks, 2006) may be responsible for the contrasting conditions and sources in the A-type granites in these environments (Bonin, 2004). The injected mafic melts will provide the heat necessary to create crustally-derived (Annen and Sparks, 2002) metaluminous and peraluminous Atype granites, whereas the peralkaline A-type granites (or sodic granites in this thesis) are produced through fractionation of these mafic melts. Part of the mantle-derived melts has to remain uncontaminated in order to fractionate and produce peralkaline granites. It is, therefore, not surprising that peralkaline A-type granites are the least common variety in post-orogenic settings (Bonin 2004).

The proposed petrogenetic model for the late Paleozoic A-type granites of the Cobequid Highlands includes, thus, both partial melting of feldspathic crustal rocks and extreme fractionation of a mafic source. The substantial mafic underplating and the voluminous coeval flood basalts and gabbros are associated with the genesis of the sodic granites as demonstrated in chapter 3. The reason for the observed variation in these post-

orogenic granite suites is principally because these geological processes were synchronous. These processes took place in a deep crustal hot zone that was created beneath the eastern Cobequid Highlands, during late Paleozoic mafic underplating, and associated with crustal extension in the Magdalen Basin (Fig. 3.14). While the calcic granites formed by relatively less hydrated melts compared to the biotite granites, and closer to the underplated area, the latter formed further to the west at lower magmatic temperatures. Partial melting for the biotite granites, however, was facilitated by the presence of additional water released from the volcaniclastic rocks of the Neoproterozoic Jeffers Group, which is abundant in the area (Fig. 3.14).

The A-type granites in the Cobequid Highlands comprise not only ferroan compositions, which are typical of A-type granites (Frost and Frost, 2010), but also magnesian compositions (Fig. 6.1). These magnesian compositions do not correspond to a specific mineralogical type but include biotite and calcic granites from all plutons. Frost and Frost (2010) noted that oxidized A-type granites from the literature tend to be magnesian, whereas the strongly reduced A-type granites are ferroan. As demonstrated in chapter 3, all the granites belong to the magnetite series and are therefore created under oxidizing conditions. Thus there is no convincing evidence that the ferroan and magnesian affinity of these rocks is the result of variations of the oxidation state of the parent melt. On the other hand, a significant difference of the source or magma mixing for the magnesian biotite and calcic granites is not supported by isotope data and trace element geochemistry as explained in chapter 3. Experimental work on partial melting of crustal rocks of Patiño-Douce (1997) demonstrated that with increasing pressure (>8 kbar) the produced melts will change from ferroan to magnesian compositions with a

peraluminous affinity. High pressures during partial melting may have been responsible for the magnesian A-type granites in the Cobequid Highlands. According to the estimated pressures (plag-melt) of these granites partial melting of the crust took place over a range of pressures (7-15 kbar), and based on the estimated thickness of the Avalonian crust (Marillier and Reid, 1990), it is suggested that the partial melting that produced the biotite and calcic granites took place in the mid- to lower crust.



Figure 6. 1 Ferroan vs. magnesian affinity of the A-type granites in the Cobequid Highlands (as defined from Frost and Frost, 2010).

6.3 **REE-ENRICHMENT AND MINERALIZATION IN A-TYPE GRANITES**

A-type granites are characteristically enriched in REE and other incompatible elements (Loiselle and Wones, 1979; Whalen et al., 1989). This is an important geochemical feature of these rocks, since A-type intrusions are often associated with economic deposits of these metals. There are several studies that demonstrate a magmatic REE-enrichment of A-type granites on the basis of primary mineralogical assemblages (Schmitt et al., 2001; Vlach and Gualda, 2007; Uher et al., 2009). Such an enrichment, however, is not as evident from whole-rock geochemistry alone. Among the studied plutons, the Wentworth pluton best records a primary enrichment, expressed by the presence of magmatic allanite-(Ce) and chevkinite-(Ce) (Papoutsa and Pe-Piper, 2013). In chapter 4 is demonstrated that although REE-enriched, the granites of the Pleasant Hills, North River and Cape Chignecto plutons host predominantly hydrothermal REEmineral assemblages that are found in fractures.

The structure of the crust, dominated by crustal scale strike-slip faults, not only had an impact on the genesis of A-type granites as discussed above, but it also had a profound effect on the intrusions and surrounding rocks in post-magmatic stages, as demonstrated in chapter 4. Textural relationships and mineral associations provide evidence of repeated REE hydrothermal circulation that resulted to the formation of several types of minerals, such as hydroxylbastnäsite-(Ce), parisite-(Ce), synchysite-(Ce), cerite, thorite, hingannite-(Y), chernovite-(Y) and Nb-rich minerals, in several generations. Hydroxylbastnäsite-(Ce) and thorite were among the first REE-minerals to precipitate in epidote veins in the eastern part of the Cobequid Highlands, under rather reducing conditions, probably soon after the emplacement of the granites. The majority of the REE-minerals, such as cerite, parisite-(Ce), synchysite-(Ce) and hingganite-(Y), are hosted in chlorite-magnetite and biotite veins in the central and western parts of the Cobequid Highlands. These minerals were deposited in the Visean according to published ages from hydrothermal biotite in the area. The highest abundances of hydrothermal REE-minerals are observed in late magnetite-rich veins, formed under strongly oxidizing conditions, in the central part of the Cobequid Shear Zone close to the Cobequid fault. In

such veins, chernovite-(Y) formed after the alteration of hydrothermal hingganite-(Y). These iron-rich hydrothermal assemblages may be related to the regional iron oxidecopper-gold IOCG hydrothermal circulation and mineralization event that is observed along the Cobequid-Chedabucto fault zone.

The association of REE with IOCG mineralization and A-type granites has been observed not only in Nova Scotia (Copper Lake, Kontak et al., 2009) but has also been reported from other parts of the world, such as in NW China (Pirajno et al., 2008) and south Australia (Elburg et al., 2012). Massive IOCG deposits throughout the Bushveld Complex are associated with LREE circulation that is represented by the mineral association of fluorite, copper, barite, gold and U-LREE minerals in strongly chloritized granitic rocks (Hunt, 2005). As in the Cobequid Highlands, the IOCG mineralization in Bushveld shows multiple episodes of hydrothermal alteration including potassic alteration, sericitisation and silicification. Sodic-calcic alteration in IOCG systems is commonly observed as an early stage alteration and can be transitional to a more K-rich alteration and chloritization when conditions become more oxidizing (Dilles et al., 2000; Kreiner, 2011). The association of calcic phases with hydrothermal biotite (calc-potassic alteration), which is observed in the Pleasant Hills pluton, is interpreted to form from moderate temperature fluids, whereas the presence of scapolite with biotite is considered a relatively low-temperature assemblage (Barton, 2014 and references therein). Extensive sodic alteration followed by potassic alteration and then by pervasive Fe-rich circulation is also reported from Wernecke Breccia in Yukon Territory (Hunt et al., 2005).

It appears, thus, that in several IOCG systems (Mt.Mainter-Mt. Babbage Inliers and Olympic Dam-Australia, Carajás Mineral Province in Brazil, Yukon Territories,

Cobequid Highlands) the hydrothermal fluids changed progressively from Na-Ca rich to K-rich to Fe-rich (Hunt et al., 2005; Groves et al., 2010; Elburg et al., 2012). The circulation of REE during IOCG events depends principally on the presence of suitable sources for these elements, such as alkaline magmatic rocks (Corriveau, 2007). In the granites of the Cobequid Highlands the replacement textures of hydrothermal phases on primary REE-minerals (Papoutsa and Pe-Piper, 2013) suggest that these rocks were the main sources of REE in the regional hydrothermal system. Kontak et al. (2009) reported the hydrothermal circulation of REE in the IOCG-associated Copper Lake Deposit, from the presence of monazite in sulfide-rich veins. The deposit is near an Ordovician A-type pluton in the Antigonish Highlands (Escarraga et al., 2012), which could have been the source of REE. The conditions of the REE-circulation and especially precipitation, however, are controlled by the progressive changes in the hydrothermal fluids, as documented in this thesis (Fig. 4.7).

As Midgisov and Williams-Jones (2014) argued, it is important for the REE not to be removed from the fluids in early stages through precipitation of highly insoluble phases, in order to reach the latest stages of hydrothermal circulation and consequently IOCG mineralization. Phases such as REE-oxides and REE-chlorides are highly soluble from reducing to weakly oxidizing conditions, however REE-fluorides may be insoluble in a range of conditions (Midgisov et al., 2009). This is compatible with the distribution of hydrothermal REE-minerals in the fractures of the Cobequid Highlands. The first REE-oxides precipitated probably during biotization (Fig. 4.7), when the fluids became weakly oxidizing. Post-Visean the hydrothermal fluids were probably saline as indicated by the presence of scapolite in the western Cobequid Highlands (Owen and Greenough,

1999) and the halogen-rich biotites of fractured mafic sills just north of the Cobequid fault in the central part of the area (Wisen, 2015). The presence of hydrothermal fluorite in calcite-rich fractures (Wisen, 2015) suggests that during the late hydrothermal stages the solubility of fluorite was limited, and that may also have influenced the solubility of the REE leading to significant precipitation of REE-minerals (Fig. 4.7).

Another feature of the hydrothermal circulation in the Cobequid Highlands is that the increase in REE concentrations in the fractures was accompanied by an increase in temperature as well (Fig. 4.7). The stability of REE-complexes correlates positively with temperature, and therefore high-temperature saline solutions can efficiently transport REE (Gieré, 1996). Such a transport of REE is indicated in the Cobequid Highlands after the Visean, since REE-hydrothermal mineral assemblages are found throughout the shear zone and in various lithologies beyond the borders of the A-type granite plutons (Wisen, 2015). Such late-stage REE circulation in IOCG systems is associated with Fe-oxides (Williams et al., 2005). It is mainly governed by the changes in the hydrothermal fluids (composition, temperature, oxidation state) compared with early stages that consequently affect the solubility of REE-complexes.

6.4 **CONCLUSIONS**

The original definition of the term "A-type" as proposed by Loiselle and Wones (1979) appears to concur with anorogenic A-type granites, the peralkaline geochemistry of which indicates a significant mantle input. In post-orogenic settings, however, the produced "A-type" granites are principally metaluminous and peraluminous with higher amounts of LILE and involve mainly crustal sources. Peralkaline granites may occur in post-orogenic settings as well, as in the Cobequid Highlands, where the sodic granites have predominantly a mantle-derived source.

The presence of crustal-scale strike-slip systems in post-orogenic settings, during extension and partial melting of the mantle that results in mantle upwelling, creates the conditions for the formation of deep crustal hot zones. The distribution of the various mineralogical types of A-type granites seems to be associated with the abundance of coeval mafic melts. The variability in the produced A-type granites in these environments may be, therefore, due to the more effective penetration of mafic magma into the fabric of the lower crust. Thus the dominant varieties of A-type granites in these settings have a crustal signature. Partial melting of the crust in these zones may occur in a range of depths from middle to lower crust and under variable pressure. The magnesian affinity of many A-type granites could, therefore be related to their derivation from a deep crustal hot zone.

Assessment and evaluation of REE-enrichment in the A-type granites must be done always in association with a mineralogical investigation since, as has been demonstrated in the Cobequid Highlands, such an enrichment is not necessarily a primary feature. The widespread hydrothermal REE remobilization and their concentration primarily in fractures across the Cobequid Shear Zone may lead to the local potential for ore bodies. It appears that the changes in the conditions of the hydrothermal fluids with time are similar to IOCG systems elsewhere. However, whether this similarity is due to local details of the geological history or a consequence of a general set of processes remains unclear.

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APPENDIX A

Supplementary material for Chapter 2.


Appendix A.1 Graphic expression of the calibration range for the pressure and temperature estimates during crystallization of hornblende according to the method of Ridolfi and Renzulli (2012). In this diagram P and T estimates from the calcic amphiboles from the Wentworth pluton are plotted for comparison.

Appendix	A.2 Re	presenta	tive EDS	s analyse	s of felds	spars fron	n the We	ntworth	granites.						
Sample	4636	4641	4641	4641	4641	4641	5056	5056	6419	6419	6419	6419	6419	6419	6419
Mineral	Ab (p)	Ab	Ab	Ab	Ab	Ab	Ab	01	Ab (p)	Ab (p)	Ab (p)	Ab (p)	Ab (p)	Ab (p)	Ab (p)
SiO ₂	69.48	66.57	67.26	67.19	68.33	67.00	64.63	61.82	69.48	69.10	67.23	69.35	68.63	68.52	68.78
Al_2O_3	17.93	19.90	20.20	20.18	19.35	19.71	22.75	24.30	18.95	18.67	18.74	18.37	19.42	18.57	19.14
FeO	0.62	0.77	0.33	1.33	0.00	0.00	0.55	0.00	0.35	0.00	0.00	0.00	0.50	0.69	0.00
CaO	0.00	1.57	1.51	0.85	0.57	0.78	0.92	4.98	0.00	0.00	0.00	0.00	0.00	0.00	0.00
Na ₂ O	11.47	10.65	10.22	9.94	11.76	11.11	9.56	8.52	11.22	12.24	12.67	12.28	11.06	10.97	11.28
$\mathbf{K}_{2}\mathbf{O}$	0.49	0.55	0.48	0.53	0.00	1.40	1.59	0.37	0.00	0.00	1.35	0.00	0.00	1.23	0.81
Total	99.99	100.01	100.00	100.02	100.01	100.00	100.00	99.99	100.00	100.01	<u>99.99</u>	100.00	99.61	99.98	100.01
ЧÞ	76	90	90	92	76	89	86	74	100	100	93	100	100	93	96
An	0	7	7	4	e	e	S	24	0	0	0	0	0	0	0
Or	e	e	e	e	0	7	6	7	0	0	7	0	0	L	S
Sample	6419	6419	6419	6419	6419	6490	6490	6490	6490	6490	6490	6490	6490	6490	6518
Mineral	Ab (p)	Ab (p)	Ab (p)	Ab (p)	Ab (p)	$\mathbf{A}\mathbf{b}$	Ab	Ab	Ab	Ab	Ab	Ab	Ab	Ab	Ab
SiO ₂	68.65	69.82	68.60	69.61	69.46	67.66	67.19	67.45	66.72	67.45	67.64	68.18	66.98	67.77	69.65
Al ₂ O ₃	18.44	18.78	18.67	18.88	18.52	19.35	20.16	19.24	20.43	20.56	19.48	19.93	20.22	20.27	17.93
FeO	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.45
CaO	0.00	0.00	0.00	0.00	0.00	1.09	0.92	1.32	1.44	1.47	1.01	0.80	0.87	1.23	0.00
Na ₂ O	12.93	11.39	10.91	10.61	12.01	11.92	11.72	12.00	10.79	10.53	11.88	11.10	11.93	10.73	11.73
K_2O	100.02	99.99	98.18	99.10	<u>99.99</u>	100.02	99.99	100.01	99.38	100.01	100.01	100.01	100.00	100.00	99.76
Total	0.00	0.00	1.83	0.89	0.00	0.00	0.00	0.00	0.61	0.00	0.00	0.00	0.00	0.00	0.25
$\mathbf{A}\mathbf{b}$	100	100	90	95	100	95	96	94	0 6	93	96	96	96	94	66
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Or	0	0	10	S	0	0	0	0	3	0	0	0	0	0	1

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Appendix	A.2 Re	presenta	tive EDS	analyses	s of felds	pars fron	n the We	ntworth {	granites	(continue	.(b				
Sample	6518	6518	6518	7658	7658	7658	7658	7658	7658	7658	7658	7658	7658	7658	7710
Mineral	Ab	Ab	Ab	Ab	Ab	Ab	Ab	Ab	Ab	Ab	Ab	Ab	Ab	Ю	Ab
SiO ₂	69.05	69.84	69.48	67.41	67.51	68.09	67.15	66.74	67.41	67.77	66.94	66.81	66.53	65.07	67.98
Al ₂ O ₃	18.42	18.33	18.42	20.07	19.95	19.58	19.88	20.33	19.61	20.10	20.31	20.09	21.24	20.82	17.84
FeO	0.54	0.33	0.44	0.00	0.00	0.00	0.00	0.46	0.67	0.00	0.00	0.00	0.00	0.80	0.00
CaO	0.00	0.00	0.00	1.83	1.50	1.11	1.93	1.78	1.19	1.33	1.79	1.89	1.25	2.83	0.00
Na ₂ O	11.26	11.51	11.31	10.45	10.77	10.97	10.49	10.31	10.27	10.81	10.46	10.41	9.69	9.79	11.99
$\mathbf{K}_{2}\mathbf{O}$	0.73	0.00	0.36	0.24	0.28	0.25	0.55	0.39	0.86	0.00	0.49	0.81	1.29	0.70	2.19
Total	100.00	100.01	100.01	100.00	100.01	100.00	100.00	100.01	100.01	100.01	99.99	100.01	100.00	100.01	100.00
$\mathbf{A}\mathbf{b}$	96	100	98	90	91	93	88	89	89	94	89	87	86	83	89
An	0	0	0	6	L	S	6	6	9	9	8	6	9	13	0
Or	4	0	2	1	2	1	3	2	5	0	3	4	8	4	11
Sample	7710	7710	7710	7710	7814	7814	7814	7814	7814	4641	5056	5056	5056	5056	5056
Mineral	Ab	Ab	Ab	Ab	Ab	Ab	Ab	Ab	Ab	Ab (p)	And	And	And	And	And
SiO ₂	69.63	68.28	69.33	68.60	68.50	68.15	69.25	68.05	68.58	68.58	64.80	62.06	63.19	62.51	62.94
Al ₂ O ₃	17.88	19.48	18.48	18.69	19.29	19.71	18.50	19.31	19.29	19.22	22.20	24.58	24.26	23.41	23.22
FeO	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.50	0.53	0.00
CaO	0.00	0.00	0.00	0.00	0.77	0.91	0.42	0.60	0.78	0.88	5.18	5.07	5.09	5.08	5.37
Na ₂ O	11.51	11.57	9.60	12.13	11.45	11.20	11.41	12.04	11.34	11.34	7.31	7.93	6.42	7.79	8.47
K_2O	0.99	0.67	2.61	0.58	0.00	0.00	0.42	0.00	0.00	0.00	0.53	0.36	0.54	0.69	0.00
Total	100.01	100.00	100.02	100.00	100.01	99.97	100.00	100.00	99.99	100.02	100.02	100.00	100.00	100.01	100.00
$\mathbf{A}\mathbf{b}$	95	96	85	97	96	96	96	97	96	96	70	72	67	71	74
An	0	0	0	0	4	4	7	e	4	4	27	26	29	25	26
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Sample	4636	4641	5056	6419	6490	6518	7658	7710	7814
Mineral	Kfs	Kfs	Kfs	Kfs	Kfs	Kfs	Kfs	Kfs	Kfs
SiO ₂	66.21	65.97	67.17	65.67	66.57	66.06	66.12	67.34	66.55
Al ₂ O ₃	17.63	18.16	17.86	17.61	18.31	17.38	17.88	18.14	18.25
FeO	0.00	0.00	0.00	0.62	0.00	0.00	0.00	0.00	0.00
CaO	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
Na ₂ O	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00
K_2O	16.18	15.86	14.96	16.10	15.12	16.54	16.01	14.50	15.20
Total	100.02	99.99	99.99	100.00	100.00	99.98	100.01	99.98	100.00
Ab	0	0	0	0	0	0	0	0	0
An	0	0	0	0	0	0	0	0	0
Or	100	100	100	100	100	100	100	100	100
Notes: At	: albite, <i>I</i>	And: and	esine, Ar	n: anorthit	e, Or: ort	hoclase,	Ol: oligo	clase, Kf	S: K-felc
(p): perthi	ite.								

Appendix A.2 Representative EDS analyses of feldspars from the Wentworth granites (continued).

Appendix A	v.3 Electr	on micro	probe an	alyses of	amphibo	oles from	the gran	ites of th	e Wentw	orth plut	on.				
Sample	4636	4636	4636	4636	4636	4636	4636	4636	4636	4636	4636	4636	4636	4636	4636
Analysis	169	171	172	173	174	175	176	177	178	179	184	187	188	190	193
SiO ₂	48.89	49.99	49.03	51.28	50.53	50.59	50.59	49.98	49.52	50.85	50.76	49.24	50.29	49.95	50.25
TiO ₂	1.58	1.35	1.34	0.98	1.30	1.09	1.64	1.65	1.41	1.69	1.33	1.51	1.69	1.61	1.27
Al_2O_3	2.05	1.29	1.70	1.08	1.42	1.19	1.26	1.58	1.62	1.17	0.96	2.05	1.83	1.93	1.31
FeO	31.75	32.19	31.95	32.55	32.50	31.21	32.18	31.84	31.22	33.01	32.55	31.34	31.70	32.25	32.66
MnO	0.78	0.72	0.71	0.74	0.73	0.66	0.67	0.63	0.57	0.55	0.75	0.69	0.74	0.74	0.75
MgO	1.37	1.30	1.67	1.21	1.28	2.35	1.57	1.92	2.12	1.31	1.19	1.50	1.77	1.43	1.00
CaO	2.75	2.80	2.44	1.51	2.17	2.54	2.33	2.30	3.17	1.84	2.27	2.63	2.38	2.12	1.96
Na ₂ O	7.85	7.34	7.64	7.25	7.60	6.97	7.35	6.63	6.28	6.74	7.14	7.12	6.72	7.76	7.31
K_2O	1.35	1.16	1.35	1.14	1.35	1.09	1.25	1.29	1.02	1.08	1.16	1.27	1.28	1.39	1.19
Ĩ	1.54	1.37	1.14	0.52	1.06	0.70	1.02	0.60	1.63	0.56	1.03	1.30	0.74	1.13	1.06
C	0.04	0.03	0.04	0.03	0.04	0.03	0.05	0.03	0.03	0.05	0.04	0.05	0.05	0.05	0.08
Total	99.95	99.52	98.99	98.29	96.98	98.40	16.90	98.44	98.59	98.83	99.16	98.69	99.19	100.35	98.83
Structural	formula c	alculated	on the b	asis of 23	oxygens	and Fe ³⁺	estimated	I with the	13eCNK	C normali	zation				
Si	7.776	7.927	7.788	8.037	7.925	7.927	7.912	7.819	7.828	7.916	7.997	7.825	7.830	7.809	7.949
AI	0.224	0.073	0.212	0.000	0.075	0.073	0.088	0.181	0.172	0.084	0.003	0.175	0.170	0.191	0.051
T-site	8.000	8.000	8.000	8.037	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
AI	0.160	0.168	0.105	0.200	0.187	0.146	0.144	0.110	0.130	0.130	0.175	0.208	0.166	0.164	0.192
${ m Fe}^{3+}$	0.057	0.140	0.332	0.559	0.269	0.482	0.297	0.644	0.503	0.695	0.332	0.259	0.530	0.311	0.410
Ti	0.189	0.161	0.160	0.115	0.153	0.128	0.193	0.194	0.168	0.198	0.157	0.181	0.198	0.189	0.152
M_{g}	0.324	0.306	0.396	0.283	0.299	0.549	0.366	0.448	0.498	0.303	0.279	0.354	0.411	0.332	0.236
Fe^{2+}	4.166	4.128	3.912	3.707	3.995	3.607	3.911	3.521	3.624	3.602	3.958	3.906	3.597	3.906	3.911
\mathbf{Mn}	0.105	0.096	0.095	0.099	0.097	0.088	0.088	0.083	0.076	0.072	0.100	0.092	0.097	0.098	0.100
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Ca	0.469	0.475	0.415	0.216	0.365	0.427	0.390	0.385	0.536	0.307	0.383	0.448	0.397	0.354	0.332
Na	1.531	1.525	1.585	1.784	1.635	1.573	1.610	1.615	1.464	1.693	1.617	1.552	1.603	1.646	1.668
B-site	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.888	0.733	0.767	0.418	0.677	0.543	0.620	0.397	0.460	0.342	0.564	0.640	0.425	0.707	0.575
K	0.274	0.235	0.274	0.228	0.271	0.219	0.250	0.257	0.206	0.214	0.234	0.258	0.255	0.276	0.239
A-site	1.162	0.968	1.041	0.646	0.947	0.761	0.870	0.653	0.666	0.556	0.798	0.899	0.680	0.983	0.814
Cations	16.162	15.968	16.041	15.683	15.947	15.761	15.870	15.653	15.666	15.556	15.798	15.899	15.680	15.983	15.814
Ł	0.775	0.685	0.570	0.259	0.526	0.347	0.506	0.295	0.817	0.278	0.513	0.655	0.363	0.560	0.529
CI	0.012	0.007	0.009	0.008	0.010	0.007	0.012	0.009	0.009	0.012	0.011	0.012	0.013	0.013	0.022
Oxygens	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000

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Sample	4636	4636	4636	4636	4636	4636	4636	4636	4636	4636	4636	4636	4636	4636
Analysis	220	221	224	225	226	227	228	229	230	231	238	239	240	242
SiO ₂	49.95	49.68	51.21	50.48	50.12	50.39	49.94	50.10	50.09	50.32	51.15	51.25	51.26	51.62
TiO_2	1.67	1.32	1.59	1.45	1.69	1.33	1.56	1.19	1.46	1.50	1.54	1.32	1.40	0.73
Al_2O_3	1.09	1.00	1.81	1.78	1.49	1.24	1.36	1.19	1.60	1.36	0.88	0.86	0.95	0.83
FeO	32.78	33.36	30.39	31.82	32.49	33.28	33.23	32.22	31.95	32.74	32.77	33.28	32.71	34.38
MnO	0.81	0.86	0.64	0.75	0.82	0.69	0.80	0.75	0.80	0.71	0.78	0.80	0.77	0.56
MgO	0.84	0.89	2.19	1.47	1.24	1.05	1.07	1.51	0.95	0.96	0.85	0.86	0.91	0.76
CaO	2.09	2.08	1.88	1.85	2.02	1.81	2.12	2.12	2.15	1.82	2.03	1.68	1.84	1.30
Na ₂ O	7.30	7.78	8.85	7.43	7.94	7.07	7.42	7.12	7.34	7.69	7.05	6.79	6.97	7.21
K_2O	1.30	1.33	1.16	1.24	1.37	1.21	1.35	1.17	1.44	1.37	1.26	1.12	1.21	0.82
Т	0.81	1.36	0.76	0.85	1.13	0.64	1.11	0.95	1.07	1.10	0.70	0.63	0.58	0.30
CI	0.04	0.05	0.05	0.04	0.05	0.04	0.05	0.05	0.04	0.05	0.03	0.03	0.04	0.06
Total	98.69	99.71	100.54	99.15	100.35	98.76	100.01	98.36	98.90	99.62	99.03	98.62	98.64	98.57
Structural form	ula calcul	ated on th	te basis of	23 oxygei	ns and Fe ³	⁺ estimate	3d with the	313eCNK	normaliz	ation				
Si	7.934	7.896	7.948	7.896	7.864	7.908	7.843	7.921	7.952	7.933	8.044	8.024	8.048	8.035
AI	0.066	0.104	0.052	0.104	0.136	0.092	0.157	0.079	0.048	0.067	0.000	0.000	0.000	0.000
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.044	8.024	8.048	8.035
AI	0.139	0.082	0.278	0.225	0.139	0.137	0.094	0.142	0.251	0.185	0.163	0.158	0.176	0.152
Fe^{3+}	0.304	0.330	0.000	0.418	0.233	0.639	0.449	0.518	0.165	0.286	0.300	0.635	0.412	0.836
Ti	0.200	0.158	0.186	0.171	0.199	0.157	0.185	0.141	0.175	0.177	0.182	0.155	0.166	0.085
Mg	0.199	0.211	0.508	0.343	0.290	0.246	0.250	0.357	0.225	0.226	0.198	0.200	0.212	0.177
Fe^{2+}	4.050	4.104	3.945	3.745	4.030	3.729	3.916	3.741	4.076	4.031	4.010	3.722	3.883	3.639
Mn	0.108	0.116	0.084	0.099	0.109	0.092	0.107	0.100	0.107	0.095	0.104	0.106	0.103	0.074
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Ca	0.356	0.355	0.313	0.310	0.339	0.305	0.357	0.359	0.366	0.307	0.298	0.258	0.261	0.181
Na	1.644	1.645	1.687	1.690	1.661	1.695	1.643	1.641	1.634	1.693	1.702	1.742	1.739	1.819
B-site	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.604	0.753	0.977	0.563	0.753	0.455	0.617	0.540	0.625	0.658	0.448	0.319	0.384	0.356
K	0.263	0.269	0.229	0.247	0.274	0.242	0.270	0.236	0.291	0.276	0.253	0.224	0.242	0.163
A-site	0.867	1.021	1.206	0.810	1.027	0.698	0.888	0.776	0.916	0.935	0.701	0.543	0.627	0.519
Cations	15.867	16.021	16.206	15.810	16.027	15.698	15.888	15.776	15.916	15.935	15.745	15.567	15.675	15.554
F	0.408	0.683	0.374	0.421	0.562	0.317	0.550	0.474	0.538	0.550	0.349	0.312	0.287	0.148
CI	0.012	0.014	0.014	0.011	0.013	0.009	0.013	0.012	0.011	0.012	0.009	0.009	0.010	0.016
Oxygens	23.000	23.000	23.058	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000

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Sample	4641	4641	4641	4641	4641	4641	4641	4641	4641	4641	4641	4641	4641	4641
Analysis	102	108	112	128	129	131	135	139	140	146	148	157	161	164
SiO ₂	44.54	39.75	43.68	43.53	43.66	43.50	44.51	45.35	43.52	46.35	44.94	42.94	41.82	43.53
TiO_2	0.50	0.52	1.05	1.41	1.17	1.29	1.38	0.81	1.13	0.86	0.98	1.41	0.67	1.08
Al_2O_3	6.68	11.06	6.29	7.37	7.28	7.43	6.91	5.82	7.28	5.13	6.31	7.53	8.93	7.09
FeO	26.30	26.18	26.59	27.38	27.48	26.38	25.90	26.09	26.86	27.15	26.87	27.52	27.04	27.30
MnO	0.79	0.82	0.96	0.81	0.88	0.85	0.65	0.93	0.80	1.11	1.04	0.86	0.80	0.83
MgO	5.47	3.80	5.33	4.66	4.59	4.81	5.59	5.71	4.97	5.82	5.31	4.62	4.91	5.00
CaO	10.20	11.29	10.18	10.16	10.29	10.18	10.02	10.66	10.18	9.60	10.16	10.08	11.26	10.16
Na ₂ O	1.77	1.95	1.89	2.14	2.01	1.93	2.11	1.59	2.03	1.98	1.76	2.07	1.93	1.77
K_2O	1.04	1.73	0.99	1.06	0.95	0.94	0.81	0.75	1.04	0.72	0.96	1.03	1.33	0.94
Ţ	0.60	0.48	0.52	0.38	0.48	0.39	0.52	0.39	0.48	0.40	0.47	0.39	0.58	0.46
CI	0.41	1.13	0.59	0.53	0.41	0.50	0.25	0.40	0.59	0.52	0.66	0.56	0.33	0.81
Total	98.28	98.69	98.08	99.44	99.19	98.21	98.65	98.49	98.87	99.62	99.44	00.66	99.58	98.97
Structural forn	nula calcul	ated on th	te basis of	23 oxygei	ns and Fe ³	³⁺ estimate	d with the	e 13eCNK	normaliz	ation				
Si	6.927	6.346	6.852	6.753	6.782	6.799	6.871	7.025	6.775	7.060	6.924	6.687	6.516	6.763
AI	1.073	1.654	1.148	1.247	1.218	1.201	1.129	0.975	1.225	0.920	1.076	1.313	1.484	1.237
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
AI	0.149	0.426	0.014	0.099	0.114	0.167	0.127	0.086	0.110	0.000	0.069	0.067	0.154	0.060
Fe^{3+}	0.668	0.284	0.692	0.588	0.612	0.547	0.574	0.537	0.635	0.886	0.711	0.725	0.566	0.822
Ti	0.058	0.063	0.124	0.165	0.137	0.152	0.160	0.094	0.132	0.099	0.114	0.165	0.078	0.126
Mg	1.268	0.904	1.247	1.078	1.063	1.120	1.285	1.318	1.154	1.320	1.220	1.072	1.139	1.157
Fe^{2+}	2.753	3.212	2.796	2.964	2.958	2.901	2.769	2.843	2.863	2.552	2.752	2.858	2.958	2.726
Mn	0.104	0.110	0.127	0.106	0.116	0.112	0.085	0.122	0.106	0.143	0.135	0.113	0.105	0.109
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Са	1.700	1.931	1.711	1.689	1.713	1.704	1.658	1.769	1.698	1.566	1.677	1.682	1.880	1.691
Na	0.300	0.069	0.289	0.311	0.287	0.296	0.342	0.231	0.302	0.434	0.323	0.318	0.120	0.309
B-site	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.233	0.534	0.286	0.332	0.318	0.290	0.290	0.246	0.311	0.151	0.203	0.306	0.463	0.225
K	0.207	0.352	0.197	0.210	0.188	0.188	0.160	0.149	0.206	0.139	0.189	0.204	0.264	0.187
A-site	0.440	0.886	0.484	0.542	0.506	0.478	0.451	0.395	0.517	0.290	0.392	0.510	0.728	0.412
Cations	15.440	15.886	15.484	15.542	15.506	15.478	15.451	15.395	15.517	15.290	15.392	15.510	15.728	15.412
Ŧ	0.293	0.241	0.256	0.188	0.236	0.193	0.253	0.189	0.234	0.193	0.227	0.194	0.284	0.227
CI	0.108	0.305	0.158	0.139	0.107	0.134	0.066	0.105	0.156	0.134	0.171	0.148	0.088	0.213
Oxygens	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000

Appendix A.3 El	ectron n	nicroprot	oe analys	es of amj	phiboles	from the	granites	of the W	'entworth	ı pluton (continue	cd).			
Sample	4641	4641	4641	4641	4641	4641	4641	4641	4641	4641	4641	4641	4641	4641	4641
Analysis	168	171	174	176	179	183	185	197	200	201	68	91	94	96	76
SiO ₂	44.43	45.45	45.49	43.77	43.96	43.00	43.48	39.80	40.32	40.49	43.39	44.01	43.37	44.64	42.29
TiO_2	1.42	1.28	1.46	1.27	0.92	0.79	1.11	0.42	0.62	0.57	1.48	1.37	1.22	1.07	0.59
Al_2O_3	6.98	7.18	6.81	7.51	7.33	8.59	7.54	11.17	10.04	9.66	7.66	7.15	7.44	6.29	7.89
FeO	26.32	25.02	25.78	27.26	27.11	26.03	25.90	27.79	27.27	26.37	27.14	27.47	27.13	26.21	28.00
MnO	0.65	0.62	0.61	0.76	0.87	0.91	0.94	0.83	0.84	0.86	0.93	0.90	0.84	0.78	0.86
MgO	5.73	6.19	6.01	4.68	4.81	4.87	5.02	3.07	3.75	4.42	4.40	4.36	4.31	5.14	3.66
CaO	10.21	9.68	10.15	10.02	9.81	10.89	10.57	11.07	11.25	11.18	10.18	9.81	10.41	10.29	10.69
Na ₂ O	2.04	1.94	2.11	2.13	1.75	1.88	1.92	1.78	1.84	1.82	2.13	2.01	1.97	1.90	1.94
$\mathbf{K}_{2}\mathbf{O}$	0.88	0.77	0.83	1.01	0.93	1.09	1.15	1.69	1.68	1.66	1.07	0.90	1.13	1.01	1.39
Ľ	0.50	0.48	0.52	0.44	0.37	0.48	0.45	0.28	0.36	0.41	0.53	0.54	0.55	0.50	0.43
CI	0.29	0.20	0.23	0.52	0.60	0.44	0.76	1.10	1.09	0.96	0.46	0.31	0.75	0.83	1.20
Total	99.45	98.80	96.98	99.38	98.45	98.96	98.84	99.01	99.05	98.37	99.36	98.82	99.10	98.65	98.93
Structural form	ula calcu	lated on 1	the basis	of 23 oxy	gens and	Fe ³⁺ estin	nated wit	th the 136	CNK no	rmalizati	0 U				
Si	6.808	6.907	6.906	6.775	6.815	6.693	6.798	6.326	6.415	6.446	6.750	6.836	6.804	6.984	6.735
AI	1.192	1.093	1.094	1.225	1.185	1.307	1.202	1.674	1.585	1.554	1.250	1.164	1.196	1.016	1.265
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
AI	0.068	0.192	0.124	0.145	0.153	0.268	0.187	0.416	0.296	0.257	0.153	0.143	0.178	0.143	0.216
Fe^{3+}	0.667	0.732	0.554	0.622	0.850	0.438	0.400	0.496	0.399	0.451	0.502	0.652	0.404	0.396	0.377
Ti	0.164	0.146	0.166	0.148	0.107	0.092	0.130	0.050	0.074	0.068	0.173	0.160	0.144	0.126	0.070
Mg	1.310	1.402	1.359	1.079	1.112	1.131	1.170	0.728	0.889	1.048	1.020	1.010	1.007	1.199	0.869
Fe^{2+}	2.707	2.448	2.719	2.907	2.664	2.950	2.987	3.197	3.229	3.060	3.030	2.916	3.155	3.033	3.352
Mn	0.085	0.080	0.078	0.100	0.114	0.120	0.125	0.112	0.113	0.115	0.122	0.119	0.111	0.104	0.115
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Ca	1.676	1.576	1.651	1.661	1.629	1.816	1.771	1.884	1.918	1.906	1.697	1.632	1.750	1.724	1.825
Na	0.324	0.424	0.349	0.339	0.371	0.184	0.229	0.116	0.082	0.094	0.303	0.368	0.250	0.276	0.175
B-site	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.281	0.149	0.271	0.301	0.154	0.384	0.353	0.434	0.484	0.467	0.340	0.238	0.349	0.301	0.424
K	0.171	0.150	0.161	0.200	0.185	0.216	0.229	0.343	0.340	0.336	0.212	0.179	0.227	0.201	0.283
A-site	0.453	0.299	0.432	0.501	0.339	0.600	0.582	0.777	0.824	0.803	0.552	0.417	0.576	0.502	0.706
Cations	15.453	15.299	15.432	15.501	15.339	15.600	15.582	15.777	15.824	15.803	15.552	15.417	15.576	15.502	15.706
Ŧ	0.244	0.229	0.247	0.216	0.179	0.234	0.225	0.142	0.180	0.206	0.263	0.263	0.273	0.249	0.216
CI	0.075	0.053	0.059	0.137	0.157	0.116	0.200	0.295	0.295	0.258	0.120	0.082	0.198	0.219	0.324
Oxygens	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000

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Sample	4641	4641	5056	5056	5056	5056	5056	5056	5056	5056	5056	5056	5056	5056	5056
Analysis	98	66	203	205	223	253	254	256	258	266	267	280	282	285	286
SiO ₂	43.25	44.17	45.33	45.58	45.06	45.67	45.69	51.96	53.27	45.95	45.95	46.30	52.52	46.01	45.27
TiO ₂	1.48	1.26	1.52	1.53	1.70	1.72	1.63	0.19	0.19	1.57	1.67	1.64	0.15	1.38	1.60
Al ₂ O ₃	7.63	7.60	7.10	6.95	6.97	7.08	6.83	2.13	1.97	6.86	7.09	6.66	2.10	6.35	6.77
FeO	26.91	26.07	21.50	21.06	21.38	21.77	21.77	17.98	18.55	21.47	21.99	21.85	18.60	21.73	21.70
MnO	0.90	0.89	0.53	0.55	0.56	0.55	0.54	0.38	0.46	0.56	0.59	0.53	0.35	0.38	0.53
MgO	4.50	4.87	8.44	8.95	8.65	8.64	8.77	12.22	12.33	8.80	8.59	8.94	12.21	9.00	8.96
CaO	9.97	10.17	10.86	10.78	10.81	10.79	10.81	12.32	11.71	10.71	10.38	10.73	12.13	11.13	10.68
Na ₂ O	2.20	1.96	1.67	1.71	1.82	1.67	1.57	0.51	0.47	1.59	1.69	1.61	0.49	1.33	1.65
K_2O	1.04	1.02	0.89	0.89	0.90	0.87	0.84	0.23	0.23	0.84	0.85	0.83	0.20	0.75	0.81
Н	0.68	0.43	0.18	0.22	0.14	0.20	0.19	0.21	0.17	0.19	0.23	0.22	0.17	0.17	0.20
CI	0.45	0.37	0.14	0.17	0.16	0.15	0.15	0.05	0.04	0.16	0.16	0.15	0.08	0.15	0.15
Total	98.99	98.81	98.16	98.38	98.15	99.10	98.77	98.18	99.36	98.70	99.19	99.46	98.97	98.38	98.30
Structural forn	nula calcu	lated on	the basis	of 23 oxy	gens and	Fe ³⁺ esti	mated wi	th the 13	CNK no	rmalizati	on				
Si	6.749	6.845	6.864	6.863	6.827	6.836	6.852	7.660	7.694	6.885	6.846	6.885	7.659	6.920	6.812
AI	1.251	1.155	1.136	1.137	1.173	1.164	1.148	0.340	0.306	1.115	1.154	1.115	0.341	1.080	1.188
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
AI	0.151	0.232	0.130	0.095	0.072	0.084	0.057	0.030	0.029	0.096	0.089	0.051	0.019	0.044	0.012
Fe^{3+}	0.547	0.463	0.474	0.547	0.496	0.581	0.632	0.189	0.440	0.603	0.725	0.656	0.325	0.605	0.737
Ti	0.173	0.146	0.173	0.173	0.194	0.193	0.183	0.021	0.020	0.177	0.188	0.183	0.016	0.156	0.181
Mg	1.046	1.126	1.905	2.009	1.953	1.927	1.960	2.686	2.655	1.965	1.909	1.982	2.654	2.019	2.009
Fe^{2+}	2.964	2.915	2.249	2.105	2.213	2.144	2.098	2.027	1.800	2.088	2.015	2.061	1.943	2.128	1.994
Mn	0.119	0.117	0.068	0.071	0.072	0.070	0.068	0.047	0.056	0.071	0.075	0.067	0.043	0.048	0.068
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Ca	1.667	1.688	1.762	1.739	1.755	1.731	1.738	1.945	1.812	1.719	1.657	1.710	1.895	1.794	1.721
Na	0.333	0.312	0.238	0.261	0.245	0.269	0.262	0.055	0.131	0.281	0.343	0.290	0.105	0.206	0.279
B-site	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	1.942	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.332	0.277	0.251	0.238	0.289	0.216	0.193	0.091	0.000	0.182	0.145	0.173	0.033	0.181	0.202
K	0.208	0.202	0.173	0.171	0.173	0.165	0.160	0.043	0.042	0.161	0.162	0.158	0.036	0.143	0.155
A-site	0.539	0.479	0.424	0.408	0.462	0.381	0.354	0.134	0.042	0.343	0.307	0.331	0.070	0.324	0.356
Cations	15.539	15.479	15.424	15.408	15.462	15.381	15.354	15.134	14.985	15.343	15.307	15.331	15.070	15.324	15.356
F	0.335	0.212	0.086	0.104	0.066	0.097	0.091	0.098	0.078	0.091	0.106	0.103	0.076	0.082	0.094
CI	0.118	0.098	0.037	0.043	0.041	0.039	0.039	0.012	0.009	0.040	0.041	0.038	0.020	0.038	0.038
Oxygens	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000

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Sample	5056	5056	5056	5056	5056	5056	5056	5056	5056	5056	5056	5056	5056	5056	5056
Analysis	288	292	294	297	300	301	302	311	312	313	314	315	324	325	332
SiO ₂	44.17	44.26	44.70	44.72	44.92	45.45	52.00	51.46	50.15	45.53	45.56	44.94	44.35	45.64	45.17
TiO ₂	1.74	3.52	1.81	1.74	1.65	1.47	0.14	0.24	0.41	1.50	1.32	1.55	1.67	1.97	1.80
Al_2O_3	7.41	6.85	7.13	6.99	7.13	6.74	2.16	2.53	3.64	6.64	6.64	6.77	6.98	7.07	7.04
FeO	22.03	20.40	21.62	22.00	22.01	21.82	18.59	18.86	20.19	22.15	21.95	22.38	22.03	21.22	21.95
MnO	0.53	0.49	0.55	0.57	0.49	0.55	0.41	0.39	0.43	0.45	0.42	0.52	0.55	0.45	0.55
MgO	8.46	8.08	8.67	8.68	8.29	8.59	12.23	11.98	10.90	8.52	8.57	8.44	8.77	8.67	8.51
CaO	10.63	11.92	10.96	10.77	11.19	10.89	11.92	11.96	11.51	11.05	11.63	10.68	10.81	10.65	10.90
Na ₂ O	1.83	1.47	1.74	1.77	1.55	1.58	0.58	0.60	0.82	1.45	1.36	1.64	1.73	1.55	1.68
K_2O	0.88	0.78	0.85	0.88	0.86	0.83	0.20	0.23	0.38	0.81	0.76	0.80	0.87	0.85	0.85
Ţ	0.21	0.22	0.23	0.20	0.14	0.21	0.23	0.12	0.15	0.16	0.17	0.16	0.19	0.12	0.13
CI	0.16	0.15	0.17	0.16	0.17	0.14	0.09	0.10	0.13	0.15	0.18	0.17	0.16	0.16	0.17
Total	98.03	98.14	98.42	98.46	98.39	98.27	98.54	98.46	98.69	98.41	98.56	98.04	98.10	98.32	98.74
Structural	l formula	calculate	ed on the	basis of 2	23 oxygen	is and Fe ³	^{t+} estimat	ed with t	he 13eCN	NK norm	alization				
Si	6.702	6.809	6.762	6.754	6.811	6.868	7.615	7.553	7.389	6.875	6.898	6.807	6.721	6.859	6.803
AI	1.298	1.191	1.238	1.243	1.189	1.132	0.373	0.437	0.611	1.125	1.102	1.193	1.245	1.141	1.197
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
AI	0.025	0.051	0.032	0.000	0.084	0.068	0.000	0.000	0.020	0.056	0.082	0.015	0.000	0.109	0.052
Fe^{3+}	0.712	0.000	0.568	0.681	0.471	0.581	0.411	0.420	0.561	0.573	0.401	0.721	0.712	0.545	0.569
Ti	0.199	0.407	0.206	0.197	0.188	0.168	0.015	0.026	0.045	0.170	0.150	0.177	0.190	0.222	0.203
Mg	1.913	1.852	1.956	1.955	1.874	1.936	2.669	2.621	2.395	1.918	1.934	1.906	1.982	1.943	1.910
Fe^{2+}	2.083	2.625	2.167	2.095	2.319	2.177	1.853	1.885	1.926	2.224	2.379	2.114	2.046	2.122	2.196
Mn	0.068	0.064	0.071	0.072	0.063	0.070	0.051	0.048	0.053	0.058	0.054	0.067	0.070	0.058	0.070
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Ca	1.728	1.964	1.776	1.742	1.817	1.763	1.871	1.881	1.817	1.787	1.887	1.733	1.756	1.714	1.759
Na	0.272	0.036	0.224	0.258	0.183	0.237	0.129	0.119	0.183	0.213	0.113	0.267	0.244	0.286	0.241
B-site	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.265	0.404	0.285	0.259	0.274	0.225	0.035	0.052	0.052	0.212	0.286	0.216	0.264	0.166	0.248
K	0.170	0.153	0.164	0.169	0.167	0.160	0.037	0.042	0.072	0.156	0.146	0.154	0.167	0.162	0.163
A-site	0.435	0.557	0.449	0.428	0.441	0.385	0.072	0.094	0.123	0.368	0.432	0.370	0.431	0.328	0.411
Cations	15.435	15.557	15.449	15.428	15.441	15.385	15.072	15.094	15.123	15.368	15.432	15.370	15.431	15.328	15.411
Ţ	0.101	0.106	0.109	0.096	0.065	0.099	0.106	0.058	0.070	0.074	0.082	0.076	0.092	0.055	0.063
C	0.041	0.039	0.044	0.041	0.043	0.035	0.023	0.024	0.032	0.037	0.046	0.042	0.042	0.039	0.042
Oxygens	23.000	23.098	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000

Appendix 1	A.3 Elect	ron micr	oprobe a	nalyses	of amphil	boles fro	m the gr	anites of	the Weni	tworth pl	uton (coi	ntinued).			
Sample	6419	6419	6419	6419	6490	6490	6490	6490	6490	6490	6490	6490	6490	6490	6490
Analysis	484	485	492	493	308	312	326	328	332	338	339	340	341	342	354
SiO ₂	49.03	50.41	48.98	49.20	43.00	43.77	44.03	43.88	44.03	44.72	45.23	45.29	44.87	44.94	43.99
TiO ₂	1.19	1.35	1.66	2.32	1.15	0.98	0.85	1.22	1.04	0.96	0.95	0.93	1.13	1.02	0.98
Al_2O_3	1.92	1.07	1.33	1.35	8.22	7.70	7.70	7.48	7.66	7.06	6.48	6.58	6.53	6.79	7.41
FeO	31.60	32.41	33.21	31.51	26.34	26.58	26.24	26.49	26.00	26.54	26.88	26.94	26.42	26.84	26.81
MnO	0.75	0.70	0.77	0.80	1.12	1.16	1.17	1.15	1.20	0.90	0.88	0.92	0.97	0.95	0.95
MgO	1.48	1.09	1.07	0.58	5.17	5.20	5.13	5.18	5.37	5.33	5.61	5.62	5.57	5.45	5.21
CaO	2.66	2.25	3.42	2.38	10.97	10.84	10.83	10.61	10.85	11.07	10.88	10.93	10.64	10.98	10.82
Na ₂ O	7.63	7.52	7.07	7.80	2.25	2.14	1.88	1.88	2.02	2.07	2.05	2.13	2.24	2.24	2.21
K_2O	1.33	1.06	1.18	1.35	1.32	1.19	1.19	1.07	1.19	1.11	0.96	0.96	0.94	0.97	1.14
Ч	1.05	0.86	0.99	0.94	0.33	0.26	0.26	0.22	0.28	0.34	0.42	0.75	0.78	0.36	0.49
CI	0.06	0.05	0.07	0.05	0.22	0.14	0.17	0.22	0.17	0.13	0.13	0.16	0.17	0.18	0.24
Total	98.70	98.77	99.74	98.27	100.08	96.96	99.44	99.41	99.80	100.24	100.46	101.20	100.26	100.72	100.24
Structural	formula	calculate	ed on the	basis of 2	23 oxygen	is and Fe	³⁺ estimat	ted with t	he 13eCN	VK norm	lization				
Si	7.825	7.991	7.782	7.953	6.637	6.731	6.789	6.754	6.767	6.867	6.905	6.892	6.892	6.866	6.771
AI	0.175	0.009	0.218	0.047	1.363	1.269	1.211	1.246	1.233	1.133	1.095	1.108	1.108	1.134	1.229
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
AI	0.186	0.190	0.030	0.209	0.130	0.126	0.187	0.109	0.152	0.144	0.070	0.071	0.074	0.088	0.115
Fe^{3+}	0.160	0.205	0.208	0.000	0.405	0.473	0.453	0.583	0.430	0.288	0.453	0.443	0.421	0.367	0.435
Ti	0.143	0.161	0.199	0.281	0.133	0.113	0.098	0.142	0.120	0.111	0.110	0.106	0.130	0.117	0.114
Mg	0.352	0.258	0.254	0.140	1.190	1.193	1.179	1.189	1.229	1.220	1.276	1.275	1.276	1.242	1.196
Fe^{2+}	4.058	4.091	4.206	4.260	2.995	2.945	2.931	2.827	2.912	3.120	2.978	2.985	2.973	3.063	3.017
Mn	0.102	0.094	0.103	0.109	0.146	0.151	0.152	0.150	0.156	0.117	0.114	0.119	0.126	0.123	0.124
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Ca	0.455	0.382	0.582	0.413	1.814	1.786	1.789	1.749	1.787	1.822	1.780	1.783	1.752	1.797	1.784
Na	1.545	1.618	1.418	1.587	0.186	0.214	0.211	0.251	0.213	0.178	0.220	0.217	0.248	0.203	0.216
B-site	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.818	0.694	0.761	0.859	0.487	0.424	0.351	0.311	0.389	0.439	0.386	0.412	0.418	0.459	0.444
K	0.270	0.215	0.240	0.278	0.260	0.234	0.235	0.210	0.234	0.218	0.187	0.186	0.183	0.189	0.223
A-site	1.088	0.908	1.000	1.137	0.747	0.658	0.586	0.521	0.623	0.657	0.573	0.598	0.601	0.648	0.667
Cations	16.088	15.908	16.000	16.137	15.747	15.658	15.586	15.521	15.623	15.657	15.573	15.598	15.601	15.648	15.667
Ч	0.531	0.430	0.495	0.483	0.163	0.127	0.125	0.106	0.136	0.167	0.202	0.361	0.377	0.172	0.240
CI	0.016	0.014	0.018	0.012	0.058	0.038	0.043	0.058	0.045	0.034	0.032	0.040	0.045	0.047	0.061
Oxygens	23.000	23.000	23.000	23.137	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000

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Samla	2026	2026	ZDEK	2026	9202	6410	6410	6410	6410	6410	6410	6410	6410	6410	6410
Analvsis	335	337	338	339	342	461	462	463	468	478	479	480	481	482	483
SiO ₂	51.80	46.07	52.98	45.83	46.01	50.25	48.78	48.31	49.59	49.83	49.16	49.70	49.03	48.43	48.83
TiO_2	0.18	1.62	0.15	1.66	1.59	0.60	1.32	1.24	1.61	1.73	1.31	1.51	1.79	1.34	1.32
Al ₂ O ₃	2.64	6.93	2.06	7.16	6.90	1.30	2.36	2.37	1.18	1.15	1.72	1.26	1.51	2.37	1.98
FeO	18.92	21.25	18.50	21.50	21.66	30.66	30.45	31.13	32.73	32.83	31.11	31.71	31.70	30.37	31.13
MnO	0.41	0.53	0.45	0.49	0.53	0.72	0.73	0.72	0.78	0.81	0.74	0.70	0.75	0.79	0.71
MgO	11.64	9.02	12.23	8.88	8.64	2.66	2.19	1.67	0.42	0.38	1.63	1.63	1.14	2.34	2.00
CaO	11.83	10.44	11.80	10.69	10.61	2.67	3.20	2.91	2.36	2.67	3.09	2.93	2.55	3.34	3.23
Na ₂ O	0.62	1.71	0.53	1.71	1.63	7.56	7.30	7.46	7.76	7.57	7.34	6.93	7.55	7.31	7.31
K_2O	0.28	0.84	0.19	0.87	0.81	1.28	1.31	1.33	1.26	1.29	1.39	1.10	1.36	1.33	1.26
Ţ	0.09	0.19	0.08	0.22	0.17	1.38	1.15	1.12	0.96	1.00	1.10	0.76	0.97	1.14	1.24
CI	0.08	0.14	0.05	0.17	0.17	0.04	0.03	0.04	0.04	0.06	0.06	0.05	0.05	0.06	0.06
Total	98.50	98.73	99.02	99.17	98.71	99.10	98.82	98.30	98.68	99.31	98.63	98.27	98.39	98.82	90.06
Structural	l formula	calculate	ed on the	basis of 2	23 oxygen	is and Fe ³	⁺⁺ estimat	ted with t	he 13eCN	VK norm	alization				
Si	7.603	6.876	7.686	6.838	6.893	7.924	7.744	7.738	7.974	7.978	7.867	7.892	7.870	7.698	7.764
AI	0.397	1.124	0.314	1.162	1.107	0.076	0.256	0.262	0.026	0.022	0.133	0.108	0.130	0.302	0.236
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
AI	0.059	0.093	0.038	0.096	0.110	0.165	0.185	0.185	0.197	0.195	0.191	0.129	0.155	0.142	0.135
Fe^{3+}	0.348	0.674	0.393	0.614	0.605	0.298	0.157	0.193	0.000	0.000	0.007	0.267	0.039	0.179	0.178
Ti	0.020	0.182	0.017	0.186	0.179	0.071	0.158	0.150	0.195	0.208	0.157	0.180	0.216	0.160	0.158
Mg	2.548	2.006	2.645	1.974	1.929	0.625	0.518	0.398	0.101	0.092	0.388	0.386	0.273	0.555	0.473
Fe^{2+}	1.974	1.978	1.852	2.068	2.109	3.744	3.885	3.976	4.402	4.396	4.156	3.944	4.215	3.858	3.960
Mn	0.051	0.067	0.055	0.062	0.068	0.096	0.097	0.098	0.106	0.109	0.100	0.094	0.102	0.107	0.096
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Ca	1.861	1.670	1.833	1.709	1.703	0.451	0.544	0.499	0.406	0.459	0.530	0.499	0.438	0.568	0.549
Na	0.139	0.330	0.148	0.291	0.297	1.549	1.456	1.501	1.594	1.541	1.470	1.501	1.562	1.432	1.451
B-site	2.000	2.000	1.981	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.037	0.164	0.000	0.204	0.177	0.764	0.790	0.816	0.825	0.807	0.807	0.631	0.787	0.822	0.802
K	0.052	0.160	0.035	0.166	0.154	0.257	0.265	0.271	0.259	0.264	0.283	0.222	0.279	0.270	0.255
A-site	0.089	0.324	0.035	0.370	0.331	1.021	1.055	1.087	1.084	1.071	1.090	0.853	1.065	1.093	1.057
Cations	15.089	15.324	15.016	15.370	15.331	16.021	16.055	16.087	16.084	16.071	16.090	15.853	16.065	16.093	16.057
ľ	0.042	0.090	0.038	0.102	0.081	0.690	0.578	0.568	0.489	0.504	0.555	0.383	0.493	0.573	0.625
CI	0.020	0.036	0.013	0.043	0.042	0.010	0.009	0.011	0.010	0.017	0.015	0.012	0.012	0.015	0.015
Oxygens	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.025	23.060	23.000	23.000	23.000	23.000	23.000

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Sample	6490	6490	6490	6490	6490	6490	6490	6490	6490	6490	6490	6490	6490	6490	6490
Analysis	401	402	403	404	405	406	407	422	423	424	425	426	427	430	432
SiO ₂	44.47	44.11	43.63	44.87	44.18	44.34	43.53	43.42	45.01	45.69	44.28	44.49	44.70	44.30	44.97
TiO ₂	0.93	0.98	1.02	0.87	1.00	1.09	1.34	1.12	0.89	1.04	1.27	0.91	1.50	0.88	1.00
Al_2O_3	7.52	7.55	7.61	7.14	7.43	7.47	7.74	7.12	6.75	6.23	7.41	7.58	7.19	7.74	6.84
FeO	27.08	27.08	27.55	27.08	27.06	27.45	27.24	28.42	26.53	26.60	27.19	26.43	26.72	26.67	26.40
MnO	1.10	1.08	1.11	1.10	1.08	1.13	1.11	1.00	1.03	1.05	1.02	0.97	1.00	0.99	1.01
MgO	5.03	4.92	4.90	5.06	5.03	5.05	4.77	5.01	5.68	5.78	5.02	5.30	5.05	5.27	5.45
CaO	10.92	11.08	10.96	10.90	11.03	10.82	10.76	10.66	10.96	10.77	10.89	10.99	10.65	10.98	10.75
Na ₂ O	1.93	1.83	2.05	1.97	1.88	2.12	2.27	1.54	1.81	2.15	2.22	1.74	2.25	1.88	2.16
$\mathbf{K}_{2}\mathbf{O}$	1.21	1.23	1.26	1.23	1.24	1.12	1.11	0.97	1.16	0.89	1.04	1.20	1.01	1.27	1.03
Ţ	0.29	0.25	0.52	0.36	0.27	0.62	0.44	0.19	0.32	0.57	0.47	0.24	0.55	0.25	0.52
CI	0.14	0.19	0.16	0.13	0.14	0.15	0.25	0.11	0.11	0.11	0.17	0.13	0.19	0.16	0.13
Total	100.63	100.29	100.79	100.69	100.36	101.36	100.56	99.54	100.24	100.88	100.98	76.99	100.81	100.39	100.25
Structural	l formula	calculate	ed on the	basis of 2	23 oxygen	is and Fe	³⁺ estimat	ted with t	he 13eCN	VK norm	alization				
Si	6.793	6.776	6.700	6.860	6.778	6.750	6.699	6.662	6.871	6.942	6.768	6.810	6.836	6.769	6.890
AI	1.207	1.224	1.300	1.140	1.222	1.250	1.301	1.286	1.129	1.058	1.232	1.190	1.164	1.231	1.110
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
AI	0.145	0.141	0.076	0.144	0.121	0.089	0.101	0.000	0.084	0.057	0.102	0.178	0.130	0.162	0.124
Fe^{3+}	0.467	0.425	0.523	0.405	0.438	0.537	0.447	0.929	0.493	0.451	0.411	0.448	0.335	0.469	0.381
Ti	0.106	0.114	0.118	0.099	0.116	0.125	0.155	0.129	0.102	0.118	0.146	0.104	0.173	0.101	0.116
Mg	1.146	1.126	1.123	1.152	1.150	1.145	1.094	1.147	1.293	1.309	1.145	1.208	1.152	1.201	1.246
Fe^{2+}	2.993	3.054	3.015	3.057	3.034	2.958	3.058	2.665	2.895	2.930	3.065	2.935	3.082	2.939	3.002
Mn	0.143	0.140	0.145	0.142	0.141	0.146	0.145	0.130	0.133	0.135	0.131	0.126	0.129	0.128	0.131
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Са	1.787	1.823	1.803	1.785	1.814	1.765	1.774	1.752	1.793	1.753	1.783	1.802	1.745	1.797	1.765
Na	0.213	0.177	0.197	0.215	0.186	0.235	0.226	0.248	0.207	0.247	0.217	0.198	0.255	0.203	0.235
B-site	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.358	0.368	0.415	0.367	0.374	0.391	0.451	0.209	0.329	0.387	0.441	0.319	0.412	0.354	0.406
K	0.236	0.240	0.246	0.240	0.243	0.218	0.218	0.190	0.226	0.173	0.202	0.234	0.198	0.247	0.201
A-site	0.595	0.608	0.661	0.608	0.617	0.609	0.669	0.399	0.555	0.560	0.643	0.553	0.610	0.601	0.607
Cations	15.595	15.608	15.661	15.608	15.617	15.609	15.669	15.399	15.555	15.560	15.643	15.553	15.610	15.601	15.607
Ţ	0.141	0.122	0.253	0.172	0.131	0.297	0.213	0.091	0.154	0.275	0.229	0.114	0.266	0.121	0.252
CI	0.037	0.049	0.041	0.033	0.036	0.037	0.064	0.029	0.028	0.029	0.044	0.034	0.050	0.041	0.033
Oxygens	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000

Sample	6490	6490	6490	6490	6490	6490	6490	6490	6490	6490	6490	6490	6490	6490	6518
Analysis	436	437	438	439	440	441	442	447	451	452	453	454	455	456	107
SiO ₂	41.49	44.17	44.29	44.58	45.05	44.72	44.46	44.33	46.26	45.14	47.53	49.07	48.81	45.28	49.03
TiO ₂	0.31	1.08	0.88	0.93	0.93	0.93	1.14	0.83	0.90	1.02	0.47	0.35	0.33	0.33	1.85
Al ₂ O ₃	10.07	7.06	6.86	6.96	6.82	6.43	6.84	7.52	5.86	5.89	5.17	3.97	4.43	6.83	1.63
FeO	28.28	28.27	28.37	27.89	28.22	28.45	27.95	27.08	25.68	28.21	25.13	24.41	24.72	25.90	33.80
MnO	1.04	1.08	1.06	1.11	1.05	1.03	1.01	1.05	0.89	0.96	0.95	0.92	0.86	0.83	1.30
MgO	4.29	4.50	4.36	4.67	4.42	4.42	4.48	4.88	6.52	6.45	7.28	7.65	7.47	6.40	0.46
CaO	11.16	10.86	10.86	10.77	10.65	10.87	10.96	10.89	11.20	10.62	11.22	11.12	11.13	11.24	5.19
Na ₂ O	2.46	2.22	2.28	2.18	1.60	2.15	2.08	2.17	1.74	1.76	1.61	1.04	0.96	1.67	4.30
K_2O	1.21	1.14	1.01	1.14	1.15	1.02	1.12	1.22	0.98	0.95	0.93	0.75	0.74	1.15	0.98
ľ	0.74	0.64	0.61	0.60	0.10	0.55	0.22	0.42	0.54	0.46	0.45	0.22	0.20	0.33	0.05
CI	0.03	0.39	0.36	0.26	0.33	0.33	0.28	0.02	0.11	0.11	0.07	0.05	0.05	0.11	0.04
Total	101.09	101.39	100.94	101.08	100.32	100.91	100.52	100.41	100.67	101.55	100.81	99.54	69. 66	100.07	98.62
Structural	l formula	calculate	ed on the	basis of 2	3 oxygen	is and Fe	³⁺ estimat	ted with t	he 13eCN	VK norm	alization				
Si	6.382	6.791	6.841	6.843	6.901	6.900	6.857	6.811	7.011	6.756	7.131	7.367	7.309	6.887	7.737
AI	1.618	1.209	1.159	1.157	1.099	1.100	1.143	1.189	0.989	1.038	0.869	0.633	0.691	1.113	0.263
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
AI	0.207	0.068	0.089	0.100	0.132	0.069	0.100	0.172	0.056	0.000	0.044	0.068	0.091	0.110	0.039
Fe^{3+}	0.688	0.430	0.388	0.429	0.556	0.373	0.315	0.354	0.390	0.918	0.463	0.462	0.535	0.548	0.517
Ti	0.036	0.124	0.102	0.107	0.108	0.108	0.133	0.095	0.103	0.115	0.053	0.039	0.038	0.038	0.219
Mg	0.984	1.032	1.004	1.068	1.009	1.018	1.031	1.117	1.472	1.439	1.629	1.711	1.667	1.450	0.107
Fe^{2+}	2.950	3.205	3.278	3.152	3.060	3.297	3.290	3.125	2.864	2.406	2.690	2.603	2.561	2.746	3.943
\mathbf{Mn}	0.135	0.140	0.139	0.144	0.136	0.134	0.132	0.136	0.115	0.122	0.120	0.116	0.108	0.107	0.173
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Ca	1.840	1.790	1.797	1.770	1.749	1.797	1.811	1.793	1.818	1.702	1.804	1.789	1.786	1.832	0.877
Na	0.160	0.210	0.203	0.230	0.251	0.203	0.189	0.207	0.182	0.298	0.196	0.211	0.214	0.168	1.123
B-site	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.572	0.450	0.480	0.420	0.223	0.442	0.432	0.440	0.329	0.212	0.273	0.092	0.063	0.324	0.194
K	0.238	0.223	0.199	0.223	0.224	0.202	0.220	0.239	0.189	0.181	0.177	0.143	0.140	0.223	0.197
A-site	0.810	0.673	0.679	0.643	0.447	0.643	0.652	0.679	0.518	0.394	0.450	0.236	0.204	0.547	0.391
Cations	15.810	15.673	15.679	15.643	15.447	15.643	15.652	15.679	15.518	15.394	15.450	15.236	15.204	15.547	15.391
Ŀ	0.358	0.313	0.297	0.292	0.049	0.266	0.105	0.203	0.261	0.218	0.211	0.105	0.094	0.160	0.025
CI	0.009	0.101	0.095	0.066	0.087	0.085	0.073	0.006	0.029	0.027	0.019	0.014	0.013	0.029	0.010
Oxygens	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000

Appendix 1	A.3 Elect	tron micr	oprobe a	malyses o	of amphi	boles fro	m the gra	anites of	the Wen	tworth pl	uton (co	ntinued).			
Sample	6518	6518	6518	6518	6518	6518	6518	7658	7658	7658	7658	7658	7658	7658	7658
Analysis	142	145	162	165	30	83	86	261	286	292	294	298	301	302	303
SiO ₂	48.29	49.67	48.16	48.59	49.50	48.72	50.75	41.33	41.96	43.37	43.25	43.62	41.51	42.78	42.57
TiO ₂	1.85	1.04	1.86	1.89	2.04	1.82	0.88	1.28	1.19	1.75	1.20	1.51	1.24	1.43	1.61
Al_2O_3	1.82	1.65	1.88	1.90	1.23	1.84	1.16	7.52	7.75	7.10	7.10	6.82	8.04	7.34	7.55
FeO	33.43	33.33	33.28	33.52	32.88	33.97	34.12	28.45	28.26	27.10	27.27	27.58	28.46	27.81	28.13
MnO	0.73	0.69	0.80	0.84	1.31	0.85	0.79	0.85	0.84	0.95	0.90	0.93	0.91	1.01	0.98
MgO	0.81	1.53	0.64	0.56	0.77	0.66	1.03	3.69	3.55	4.46	4.63	4.13	3.49	3.89	3.67
CaO	6.11	4.84	6.17	5.94	4.97	5.84	3.35	10.41	10.40	10.22	10.36	9.88	10.56	10.28	10.31
Na ₂ O	3.81	4.52	4.38	4.19	4.23	3.85	5.38	2.24	2.09	2.32	2.30	2.15	2.23	2.33	2.30
K_2O	1.05	0.70	1.08	1.08	1.18	0.97	0.79	1.23	1.29	0.95	0.92	0.98	1.29	1.01	0.99
F	0.07	0.01	0.06	0.07	0.04	0.00	0.03	0.75	0.34	0.48	0.60	0.21	0.56	0.36	0.38
CI	0.05	0.02	0.05	0.05	0.02	0.04	0.01	0.30	0.43	0.23	0.15	0.22	0.36	0.27	0.25
Total	98.02	98.00	98.34	98.61	98.16	98.56	98.28	98.05	98.09	98.91	99.66	98.02	98.63	98.51	98.72
Structural	formula	calculate	ed on the	basis of 2	23 oxygen	is and Fe ³	⁺ estimat	ted with t	he 13eCN	NK norm	alization				
Si	7.691	7.768	7.708	7.720	7.825	7.689	7.900	6.620	6.683	6.780	6.776	6.849	6.607	6.749	6.708
AI	0.309	0.232	0.292	0.280	0.175	0.311	0.100	1.380	1.317	1.220	1.224	1.151	1.393	1.251	1.292
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
AI	0.033	0.072	0.062	0.076	0.054	0.031	0.113	0.037	0.137	0.087	0.085	0.110	0.113	0.112	0.109
Fe^{3+}	0.358	0.784	0.088	0.223	0.421	0.497	0.884	0.514	0.440	0.405	0.498	0.510	0.431	0.408	0.422
Ti	0.222	0.122	0.224	0.226	0.242	0.217	0.103	0.154	0.143	0.205	0.142	0.178	0.148	0.170	0.191
Mg	0.193	0.356	0.152	0.133	0.182	0.155	0.238	0.882	0.843	1.039	1.081	0.967	0.828	0.914	0.862
Fe^{2+}	4.096	3.574	4.366	4.229	3.926	3.987	3.558	3.297	3.324	3.138	3.075	3.112	3.357	3.262	3.286
Mn	0.099	0.091	0.108	0.113	0.175	0.114	0.104	0.115	0.113	0.125	0.120	0.123	0.122	0.134	0.131
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Ca	1.042	0.811	1.058	1.010	0.842	0.988	0.558	1.786	1.774	1.712	1.738	1.662	1.801	1.738	1.740
Na	0.958	1.189	0.942	0.990	1.158	1.012	1.442	0.214	0.226	0.288	0.262	0.338	0.199	0.262	0.260
B-site	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.219	0.181	0.416	0.302	0.137	0.166	0.181	0.482	0.419	0.416	0.436	0.316	0.490	0.451	0.442
K	0.213	0.139	0.220	0.218	0.237	0.196	0.158	0.252	0.261	0.189	0.184	0.197	0.261	0.203	0.198
A-site	0.432	0.320	0.636	0.520	0.374	0.362	0.339	0.734	0.680	0.605	0.620	0.513	0.751	0.654	0.640
Cations	15.432	15.320	15.636	15.520	15.374	15.362	15.339	15.734	15.680	15.605	15.620	15.513	15.751	15.654	15.640
H	0.034	0.006	0.030	0.033	0.018	0.000	0.014	0.377	0.169	0.238	0.296	0.105	0.282	0.181	0.188
CI	0.013	0.006	0.013	0.012	0.005	0.010	0.003	0.080	0.116	0.060	0.039	0.058	0.096	0.072	0.066
Oxygens	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000

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Sample	7658	7658	7658	7658	7658	7658	7658	7658	7658	7710	7710	7710	7710	7710	7710
Analysis	305	306	311	335	337	338	340	342	344	102	103	104	111	112	113
SiO ₂	44.22	42.36	41.68	44.44	42.47	42.33	44.53	45.12	43.93	48.20	47.73	48.45	47.06	47.23	48.44
TiO_2	1.23	1.51	1.40	1.57	1.63	1.58	1.60	1.21	1.62	1.37	1.40	1.30	1.19	1.24	1.33
Al_2O_3	6.33	7.59	7.93	6.68	7.10	7.68	6.89	6.18	7.10	2.53	2.51	2.31	3.26	3.42	2.56
FeO	28.09	28.23	28.51	26.34	28.45	28.76	26.17	27.04	25.73	30.16	30.23	30.08	29.35	29.06	29.27
MnO	0.97	0.89	0.91	0.71	0.81	0.86	0.67	0.87	0.85	0.80	0.83	0.90	0.80	0.80	0.80
MgO	4.10	3.61	3.67	5.90	4.76	3.84	6.01	5.07	5.51	2.53	2.65	2.68	3.03	3.44	3.13
CaO	10.11	10.38	10.59	10.22	10.07	10.07	10.10	10.53	10.04	6.85	6.66	6.68	7.90	8.04	6.91
Na ₂ O	2.24	2.22	2.22	2.44	2.25	2.42	2.43	1.69	2.26	4.30	4.26	4.43	4.02	3.87	4.18
K_2O	0.81	1.07	1.30	0.87	0.90	0.99	0.86	0.99	0.92	1.09	1.12	1.11	1.06	1.04	1.07
ľ	0.56	0.52	0.61	0.99	0.60	0.78	0.81	0.32	0.48	0.82	0.73	0.79	0.94	0.81	0.73
CI	0.20	0.23	0.29	0.22	0.26	0.31	0.23	0.24	0.26	0.05	0.05	0.05	0.06	0.07	0.06
Total	98.86	98.61	99.10	100.38	99.30	99.62	100.29	99.26	98.68	98.70	98.17	98.78	98.65	99.01	98.48
Structural	l formula	calculate	ed on the	basis of 2	23 oxyger	is and Fe	3+ estimat	ted with t	he 13eC ^N	NK norm	alization				
Si	6.924	6.702	6.596	6.805	6.602	6.625	6.790	6.966	6.804	7.653	7.597	7.676	7.503	7.461	7.652
AI	1.076	1.298	1.404	1.195	1.300	1.375	1.210	1.034	1.196	0.347	0.403	0.324	0.497	0.539	0.348
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
AI	0.091	0.116	0.074	0.009	0.000	0.039	0.027	0.088	0.098	0.125	0.066	0.106	0.115	0.096	0.129
Fe^{3+}	0.463	0.407	0.464	0.576	0.806	0.655	0.632	0.481	0.530	0.018	0.188	0.053	0.000	0.031	0.067
Ti	0.145	0.179	0.166	0.181	0.190	0.186	0.184	0.140	0.188	0.163	0.168	0.155	0.143	0.147	0.159
Mg	0.957	0.850	0.865	1.346	1.104	0.897	1.366	1.168	1.271	0.598	0.630	0.632	0.720	0.811	0.737
Fe^{2+}	3.216	3.328	3.309	2.797	2.794	3.109	2.705	3.010	2.802	3.988	3.836	3.932	3.914	3.807	3.800
Mn	0.128	0.120	0.123	0.092	0.107	0.113	0.086	0.113	0.111	0.108	0.112	0.121	0.107	0.107	0.108
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Ca	1.695	1.760	1.795	1.677	1.677	1.688	1.650	1.742	1.665	1.166	1.136	1.133	1.349	1.361	1.169
Na	0.305	0.240	0.205	0.323	0.323	0.312	0.350	0.258	0.335	0.834	0.864	0.867	0.651	0.639	0.831
B-site	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.376	0.441	0.477	0.402	0.356	0.423	0.367	0.248	0.345	0.490	0.450	0.495	0.592	0.547	0.450
K	0.162	0.216	0.262	0.170	0.179	0.198	0.167	0.195	0.182	0.221	0.227	0.225	0.215	0.210	0.215
A-site	0.538	0.657	0.739	0.572	0.535	0.621	0.534	0.443	0.527	0.712	0.677	0.721	0.807	0.756	0.665
Cations	15.538	15.657	15.739	15.572	15.535	15.621	15.534	15.443	15.527	15.712	15.677	15.721	15.807	15.756	15.665
Ţ	0.276	0.260	0.307	0.478	0.296	0.385	0.392	0.155	0.234	0.413	0.367	0.393	0.472	0.403	0.365
CI	0.054	0.063	0.077	0.058	0.068	0.083	0.060	0.063	0.069	0.014	0.015	0.014	0.016	0.019	0.015
Oxygens	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.030	23.000	23.000

A.3 El	ectron mic	roprobe a	malyses o	of amphil	ooles froi	m the gra	anites of	the Went	tworth pl	uton (co	ntinued).			
	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710
	15	154	155	164	17	175	176	177	187	188	189	190	21	221
-	47.54	52.30	49.45	50.52	46.96	47.17	47.47	47.32	47.11	47.00	47.48	48.83	48.20	46.79
	1.11	0.31	0.78	0.44	1.23	1.05	1.04	1.07	1.02	1.09	1.07	1.55	1.29	1.18
	3.39	0.50	2.03	1.85	3.66	3.65	3.42	3.59	3.71	3.54	3.41	2.01	2.37	3.64
	28.36	28.49	28.69	24.75	28.37	27.74	27.70	28.03	28.37	28.17	28.81	30.76	30.14	29.30
	0.76	0.82	0.85	0.70	0.73	0.78	0.79	0.74	0.76	0.78	0.78	0.88	0.79	0.76
	3.82	4.51	4.02	6.92	3.61	4.46	4.39	3.99	3.83	3.97	3.73	1.90	2.56	3.26
	7.95	8.08	6.54	8.87	8.33	8.30	8.31	8.38	8.35	8.32	8.30	5.89	6.75	8.28
	3.70	2.76	4.37	3.04	3.71	3.95	3.97	3.90	3.91	4.00	3.89	4.61	4.40	3.93
	1.01	0.99	1.35	0.59	0.97	0.87	0.92	0.89	0.92	0.90	0.88	1.14	1.15	1.03
	0.54	0.29	0.86	1.03	0.73	1.04	1.13	0.95	1.01	0.99	0.86	0.64	0.76	0.94
	0.05	0.04	0.04	0.03	0.07	0.05	0.05	0.04	0.06	0.04	0.06	0.07	0.05	0.07
2	98.24	90.06	98.98	98.73	98.38	99.05	99.20	98.89	99.04	98.80	99.27	98.27	98.47	99.18
E	la calculat	ed on the	basis of 2	3 oxygen	s and Fe ³	^{tt} estimat	ted with t	he 13eCN	VK norm	alization				
4	7.498	8.059	7.732	7.760	7.452	7.409	7.464	7.469	7.440	7.440	7.471	7.774	7.674	7.420
9	0.502	0.000	0.268	0.240	0.548	0.591	0.536	0.531	0.560	0.560	0.529	0.226	0.326	0.580
2	8.000	8.059	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
5	0.127	0.091	0.107	0.094	0.136	0.084	0.098	0.136	0.129	0.101	0.103	0.150	0.119	0.100
0	0.088	0.036	0.191	0.105	0.000	0.089	0.000	0.000	0.000	0.000	0.013	0.043	0.005	0.000
ŝ	0.132	0.035	0.092	0.051	0.147	0.124	0.123	0.126	0.121	0.129	0.126	0.185	0.155	0.141
$\tilde{\mathbf{\omega}}$	0.899	1.036	0.937	1.585	0.855	1.044	1.030	0.939	0.902	0.937	0.875	0.450	0.608	0.770
0	3.652	3.636	3.561	3.074	3.764	3.555	3.643	3.700	3.747	3.729	3.779	4.052	4.007	3.887
\sim	0.101	0.107	0.112	0.090	0.098	0.104	0.106	0.099	0.101	0.104	0.104	0.119	0.107	0.102
9	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
4	1.344	1.274	1.096	1.460	1.417	1.396	1.400	1.418	1.413	1.412	1.399	1.005	1.151	1.407
9	0.656	0.726	0.904	0.540	0.583	0.604	0.600	0.582	0.587	0.588	0.601	0.995	0.849	0.593
0	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
5	0.476	0.098	0.420	0.364	0.559	0.599	0.609	0.611	0.611	0.639	0.586	0.427	0.508	0.615
0	0.204	0.194	0.269	0.115	0.197	0.175	0.185	0.179	0.185	0.182	0.177	0.231	0.233	0.209
9	0.679	0.291	0.689	0.479	0.756	0.774	0.794	0.790	0.796	0.821	0.763	0.658	0.742	0.824
Æ,) 15.679	15.351	15.689	15.479	15.756	15.774	15.794	15.790	15.796	15.821	15.763	15.658	15.742	15.824
7	0.269	0.139	0.427	0.502	0.368	0.517	0.562	0.472	0.504	0.496	0.429	0.320	0.384	0.471
∞	0.014	0.009	0.011	0.007	0.018	0.013	0.012	0.011	0.015	0.012	0.017	0.019	0.014	0.019
Ξ) 23.000	23.000	23.000	23.000	23.027	23.000	23.002	23.033	23.010	23.016	23.000	23.000	23.000	23.017

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Sample	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710
Analysis	222	223	224	225	226	227	228	229	230	231	235	239	242	244	245
SiO ₂	47.48	47.09	46.76	46.80	48.35	47.63	47.73	46.87	46.63	47.09	47.80	48.60	47.06	47.15	47.01
TiO_2	1.09	1.05	1.19	1.17	1.26	1.11	1.17	1.20	1.24	1.09	1.07	1.36	1.09	1.03	1.02
Al_2O_3	3.27	3.60	3.66	3.25	3.49	3.63	3.35	3.64	3.66	3.47	3.29	2.34	3.33	3.54	3.63
FeO	27.98	28.37	28.63	28.78	28.17	27.89	27.03	27.83	28.57	28.00	28.82	29.97	27.89	27.22	27.60
MnO	0.76	0.79	0.75	0.78	0.73	0.75	0.73	0.72	0.77	0.75	0.69	0.79	0.73	0.81	0.79
MgO	4.20	3.82	3.49	3.23	4.05	4.36	5.00	4.30	4.00	4.16	3.76	2.60	3.99	4.59	4.17
CaO	8.33	8.50	8.42	8.17	8.18	8.49	8.39	8.53	8.56	8.41	8.32	6.65	8.29	8.50	8.34
Na ₂ O	4.04	4.07	3.86	3.98	3.93	4.08	4.01	3.82	3.85	3.88	3.66	4.26	3.84	3.90	4.00
K_2O	0.95	0.90	1.00	1.08	1.11	0.99	0.95	1.03	1.10	0.99	0.96	1.08	1.00	0.90	0.89
Ţ	1.16	1.06	0.84	0.95	0.71	1.11	1.06	1.03	1.09	1.01	0.67	0.48	0.80	0.92	0.85
CI	0.06	0.06	0.06	0.07	0.06	0.07	0.07	0.07	0.08	0.06	0.07	0.06	0.05	0.05	0.05
Total	99.32	99.31	98.67	98.27	100.05	100.10	99.48	99.04	99.54	98.92	99.12	98.17	98.06	98.62	98.35
Structural	l formula	calculate	ed on the	basis of 2	23 oxygen	is and Fe ²	³⁺ estimat	ted with th	he 13eCN	IK norma	lization				
Si	7.486	7.447	7.436	7.506	7.516	7.443	7.454	7.399	7.359	7.443	7.507	7.706	7.489	7.440	7.451
AI	0.514	0.553	0.564	0.494	0.484	0.557	0.546	0.601	0.641	0.557	0.493	0.294	0.511	0.560	0.549
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
AI	0.093	0.117	0.121	0.121	0.155	0.111	0.071	0.075	0.038	0.089	0.116	0.144	0.113	0.098	0.129
Fe^{3+}	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.012	0.000	0.018	0.037	0.000	0.000	0.000
Ti	0.129	0.125	0.142	0.141	0.147	0.130	0.137	0.143	0.148	0.130	0.127	0.163	0.130	0.122	0.121
Mg	0.988	0.900	0.828	0.772	0.939	1.015	1.164	1.011	0.941	0.979	0.881	0.613	0.947	1.080	0.984
Fe^{2+}	3.689	3.752	3.807	3.861	3.663	3.645	3.531	3.674	3.758	3.701	3.767	3.937	3.712	3.592	3.659
Mn	0.101	0.106	0.102	0.106	0.095	0.099	0.097	0.097	0.103	0.101	0.092	0.106	0.098	0.108	0.107
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Ca	1.408	1.441	1.435	1.404	1.362	1.422	1.404	1.443	1.447	1.424	1.399	1.130	1.413	1.438	1.417
Na	0.592	0.559	0.565	0.596	0.638	0.578	0.596	0.557	0.553	0.576	0.601	0.870	0.587	0.562	0.583
B-site	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.642	0.690	0.626	0.642	0.547	0.657	0.618	0.613	0.626	0.614	0.514	0.439	0.597	0.632	0.646
K	0.192	0.181	0.203	0.222	0.219	0.198	0.189	0.207	0.222	0.200	0.192	0.218	0.202	0.182	0.179
A-site	0.834	0.870	0.830	0.864	0.767	0.854	0.807	0.820	0.848	0.815	0.706	0.657	0.799	0.814	0.825
Cations	15.834	15.870	15.830	15.864	15.767	15.854	15.807	15.820	15.848	15.815	15.706	15.657	15.799	15.814	15.825
Ţ	0.580	0.528	0.422	0.480	0.351	0.551	0.525	0.515	0.544	0.506	0.333	0.239	0.402	0.461	0.428
CI	0.016	0.015	0.015	0.019	0.017	0.017	0.017	0.018	0.020	0.016	0.020	0.016	0.014	0.014	0.015
Oxygens	23.040	23.063	23.053	23.088	23.047	23.045	23.005	23.012	23.000	23.015	23.000	23.000	23.037	23.017	23.032

			-	r	-		9			-	/				
Sample	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710
Analysis	246	247	256	259	260	261	266	267	268	269	27	270	271	272	273
SiO ₂	46.60	46.69	49.05	49.60	48.78	49.18	47.69	46.94	46.91	47.52	47.88	46.90	47.76	47.41	47.17
TiO_2	1.12	1.21	0.51	0.44	0.73	0.96	1.22	1.10	1.18	1.05	0.99	1.10	1.10	1.26	1.00
Al_2O_3	3.69	3.51	2.70	2.34	5.28	9.18	2.77	3.65	3.40	3.60	3.23	3.64	2.99	3.39	3.38
FeO	28.25	28.88	26.19	25.65	26.52	23.15	29.37	27.75	27.93	27.57	28.63	27.86	29.18	28.24	26.62
MnO	0.81	0.79	0.86	0.78	0.71	0.64	0.80	0.71	0.74	0.71	0.72	0.73	0.80	0.75	0.73
MgO	3.85	3.25	5.62	5.14	3.58	3.23	2.82	4.33	4.26	4.61	3.64	3.94	2.93	3.90	5.02
CaO	8.57	8.11	7.95	9.40	5.95	7.21	7.45	8.53	8.49	8.40	7.61	8.39	7.76	8.63	8.40
Na ₂ O	3.90	3.86	4.10	3.48	5.25	3.49	4.23	3.82	3.51	3.76	4.01	3.68	3.91	3.83	4.00
K_2O	0.97	1.08	0.95	0.77	0.84	2.34	1.04	0.96	0.99	0.88	0.88	1.01	0.98	1.05	0.93
Ţ	1.01	0.69	1.09	0.62	0.80	0.83	0.76	1.10	0.69	1.01	0.67	0.71	0.64	0.99	1.07
CI	0.06	0.08	0.05	0.04	0.06	0.06	0.06	0.06	0.07	0.05	0.07	0.06	0.06	0.07	0.07
Total	98.82	98.14	90.66	98.26	98.48	100.26	98.20	98.94	98.16	99.16	98.31	98.02	98.10	99.53	98.38
Structural	l formula	calculate	ed on the	basis of 2	3 oxygen	is and Fe ³	³⁺ estimat	ed with the	he 13eCN	VK normé	lization				
Si	7.403	7.458	7.625	7.820	7.581	7.484	7.625	7.412	7.425	7.436	7.555	7.453	7.612	7.479	7.455
AI	0.597	0.542	0.375	0.180	0.419	0.516	0.375	0.588	0.575	0.564	0.445	0.547	0.388	0.521	0.545
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
AI	0.093	0.118	0.119	0.253	0.546	1.129	0.146	0.092	0.059	0.099	0.154	0.134	0.174	0.109	0.084
Fe^{3+}	0.000	0.000	0.065	0.000	0.000	0.000	0.000	0.000	0.078	0.082	0.080	0.000	0.000	0.000	0.000
Ti	0.133	0.145	0.060	0.052	0.085	0.109	0.147	0.130	0.140	0.124	0.117	0.131	0.131	0.150	0.119
Mg	0.911	0.774	1.303	1.208	0.828	0.733	0.672	1.019	1.005	1.075	0.855	0.934	0.696	0.916	1.183
Fe^{2+}	3.754	3.857	3.340	3.382	3.447	2.946	3.927	3.664	3.619	3.527	3.697	3.703	3.890	3.726	3.517
Mn	0.109	0.107	0.113	0.104	0.093	0.082	0.108	0.095	0.099	0.094	0.096	0.098	0.108	0.100	0.098
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Ca	1.459	1.388	1.324	1.587	0.990	1.175	1.276	1.443	1.440	1.409	1.286	1.429	1.326	1.458	1.423
Na	0.541	0.612	0.676	0.413	1.010	0.825	0.724	0.557	0.560	0.591	0.714	0.571	0.674	0.542	0.577
B-site	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.660	0.583	0.560	0.653	0.571	0.205	0.587	0.613	0.518	0.551	0.512	0.563	0.533	0.629	0.648
K	0.196	0.220	0.188	0.155	0.167	0.455	0.211	0.193	0.200	0.176	0.177	0.206	0.198	0.212	0.187
A-site	0.856	0.803	0.747	0.808	0.738	0.659	0.799	0.806	0.717	0.727	0.690	0.769	0.731	0.841	0.835
Cations	15.856	15.803	15.747	15.808	15.738	15.659	15.799	15.806	15.717	15.727	15.690	15.769	15.731	15.841	15.835
Ţ	0.508	0.347	0.533	0.307	0.393	0.400	0.386	0.548	0.345	0.500	0.335	0.358	0.324	0.496	0.533
CI	0.017	0.021	0.013	0.011	0.015	0.015	0.015	0.016	0.019	0.014	0.018	0.016	0.017	0.018	0.018
Oxygens	23.039	23.029	23.000	23.286	23.013	23.333	23.070	23.007	23.000	23.000	23.000	23.024	23.053	23.093	23.016

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Sample	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710	7710
Analysis	274	275	28	296	297	298	299	300	301	31	40	41	49	53	60
SiO ₂	47.00	48.32	48.14	47.09	46.84	47.01	47.16	46.62	47.97	50.01	47.12	47.04	47.04	51.32	49.92
TiO ₂	1.18	1.14	0.94	1.24	1.18	1.11	1.19	1.12	1.38	0.82	1.24	1.07	1.13	0.16	0.49
Al ₂ O ₃	3.66	2.27	3.05	3.43	3.56	3.57	3.52	3.67	2.47	1.65	3.62	3.71	3.65	0.40	2.42
FeO	27.73	29.84	24.18	29.85	27.84	27.88	27.41	27.48	29.84	27.94	27.73	27.62	27.25	25.28	21.27
MnO	0.72	0.83	0.64	0.71	0.69	0.78	0.73	0.74	0.83	0.82	0.78	0.76	0.73	0.93	0.58
MgO	4.20	2.96	6.95	2.69	4.10	4.14	4.59	4.44	2.52	4.07	4.24	4.48	4.49	3.72	9.31
CaO	8.54	6.88	8.63	7.80	8.73	8.47	8.44	8.42	6.99	6.46	8.37	8.36	8.50	13.62	8.93
Na_2O	3.93	4.43	3.54	3.86	3.75	3.81	3.68	3.84	4.50	4.53	3.75	3.66	3.73	3.25	3.84
K_2O	0.97	1.06	0.96	1.07	1.09	0.88	1.03	0.99	1.11	1.02	0.98	0.97	1.04	0.28	0.62
Ŧ	1.04	0.85	1.27	0.58	1.01	1.07	0.98	1.04	0.85	0.94	1.06	1.00	0.87	0.27	1.37
CI	0.07	0.05	0.05	0.07	0.07	0.05	0.04	0.06	0.05	0.03	0.05	0.05	0.07	0.01	0.02
Total	99.04	98.62	98.34	98.39	98.84	98.78	98.77	98.42	98.51	98.29	98.93	98.72	98.51	99.24	98.76
Structural	l formula	calculate	ed on the	basis of 2	23 oxygen	is and Fe	³⁺ estimat	ted with t	he 13eCN	IK norm	alization				
Si	7.428	7.669	7.490	7.499	7.436	7.435	7.423	7.389	7.667	7.861	7.426	7.400	7.431	8.404	7.608
AI	0.572	0.331	0.510	0.501	0.564	0.565	0.577	0.611	0.333	0.139	0.574	0.600	0.569	0.000	0.392
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.404	8.000
AI	0.109	0.092	0.048	0.142	0.101	0.100	0.076	0.075	0.133	0.167	0.099	0.086	0.109	0.076	0.043
Fe^{3+}	0.000	0.051	0.106	0.000	0.000	0.000	0.044	0.028	0.000	0.015	0.014	0.132	0.000	0.000	0.067
Ti	0.141	0.136	0.110	0.149	0.141	0.132	0.140	0.133	0.166	0.097	0.147	0.127	0.135	0.020	0.056
Mg	0.990	0.700	1.611	0.638	0.970	0.976	1.078	1.050	0.601	0.954	0.995	1.052	1.058	0.909	2.116
Fe^{2+}	3.665	3.908	3.041	3.975	3.695	3.688	3.565	3.615	3.988	3.658	3.641	3.502	3.601	3.461	2.645
Mn	0.096	0.112	0.084	0.096	0.093	0.104	0.097	0.099	0.112	0.109	0.104	0.101	0.098	0.130	0.074
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Ca	1.446	1.170	1.439	1.331	1.485	1.436	1.423	1.430	1.197	1.089	1.413	1.408	1.439	1.986	1.459
Na	0.554	0.830	0.561	0.669	0.515	0.564	0.577	0.570	0.803	0.911	0.587	0.592	0.561	0.014	0.541
B-site	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.652	0.532	0.507	0.523	0.638	0.603	0.547	0.609	0.591	0.469	0.557	0.524	0.583	1.018	0.592
K	0.195	0.215	0.191	0.217	0.220	0.178	0.206	0.201	0.227	0.205	0.196	0.195	0.209	0.058	0.120
A-site	0.847	0.747	0.698	0.741	0.858	0.782	0.753	0.810	0.818	0.674	0.753	0.720	0.792	1.075	0.712
Cations	15.847	15.747	15.698	15.741	15.858	15.782	15.753	15.810	15.818	15.674	15.753	15.720	15.792	16.479	15.712
Ţ	0.518	0.428	0.622	0.290	0.506	0.537	0.489	0.520	0.432	0.466	0.527	0.498	0.435	0.138	0.660
CI	0.018	0.012	0.013	0.020	0.018	0.013	0.011	0.017	0.013	0.007	0.014	0.014	0.019	0.002	0.004
Oxygens	23.055	23.000	23.000	23.005	23.080	23.008	23.000	23.000	23.073	23.000	23.000	23.000	23.020	24.397	23.000

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Sample	7710	7710	7710	7710	7710	7710	7710	7710	7710	7814	7814	7814	7814	7814	7814
Analysis	65	99	71	72	73	74	82	83	84	11	13	14	19	20	21
SiO ₂	49.08	47.68	46.64	47.12	47.30	47.08	46.70	46.79	46.85	43.90	45.70	44.62	44.68	43.95	44.71
TiO ₂	0.59	1.18	1.17	1.07	1.08	1.20	1.19	1.15	1.24	0.97	0.94	0.85	0.77	0.87	0.83
Al_2O_3	2.93	2.60	3.70	3.61	3.50	3.50	3.67	3.68	3.60	6.72	5.97	6.38	6.01	5.69	5.78
FeO	25.30	29.90	28.37	27.34	27.53	27.86	27.70	27.56	28.20	28.39	28.09	27.19	28.33	28.97	28.01
MnO	0.74	0.85	0.75	0.74	0.75	0.73	0.79	0.74	0.74	1.32	1.38	1.31	1.38	1.40	1.42
MgO	6.27	2.89	3.60	4.57	4.57	3.96	3.87	4.18	3.83	3.67	4.41	4.42	4.27	3.80	4.04
CaO	8.47	7.11	8.41	8.21	8.25	8.09	8.52	8.32	8.46	10.32	10.29	10.23	10.50	10.12	10.40
Na ₂ O	3.71	4.22	3.86	3.91	3.96	3.90	3.85	3.91	3.77	2.17	2.28	2.18	2.18	2.14	2.02
K_2O	0.82	1.06	0.98	1.05	0.86	1.04	0.95	0.96	1.06	1.16	1.11	1.06	1.18	1.16	1.12
Ч	0.64	0.84	0.87	0.86	1.06	1.00	0.96	0.97	0.92	0.49	0.67	0.54	0.71	0.69	0.62
CI	0.04	0.05	0.07	0.06	0.05	0.06	0.06	0.05	0.06	0.25	0.26	0.27	0.30	0.32	0.23
Total	98.60	98.38	98.42	98.54	98.91	98.43	98.25	98.31	98.73	99.35	101.09	99.04	100.29	99.10	99.18
Structural	l formula	calculate	ed on the	basis of 2	3 oxygen	is and Fe	³⁺ estimat	ted with t	he 13eC ^N	VK norm	alization				
Si	7.586	7.590	7.431	7.427	7.436	7.470	7.448	7.427	7.432	6.886	7.017	6.971	6.951	6.935	7.020
AI	0.414	0.410	0.569	0.573	0.564	0.530	0.552	0.573	0.568	1.114	0.983	1.029	1.049	1.058	0.980
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
AI	0.119	0.078	0.125	0.097	0.083	0.125	0.136	0.115	0.105	0.127	0.096	0.145	0.053	0.000	0.088
Fe^{3+}	0.078	0.107	0.000	0.044	0.066	0.000	0.000	0.000	0.000	0.399	0.386	0.389	0.425	0.548	0.358
Ti	0.068	0.141	0.140	0.127	0.127	0.143	0.143	0.137	0.148	0.114	0.109	0.100	0.090	0.103	0.098
Mg	1.444	0.687	0.855	1.074	1.071	0.936	0.920	0.989	0.905	0.859	1.009	1.029	0.990	0.893	0.947
Fe^{2+}	3.193	3.873	3.780	3.560	3.553	3.697	3.695	3.659	3.741	3.326	3.221	3.164	3.261	3.269	3.320
Mn	0.097	0.114	0.101	0.099	0.100	0.098	0.107	0.100	0.100	0.175	0.179	0.173	0.181	0.188	0.189
C-site	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
Ca	1.403	1.212	1.436	1.386	1.390	1.375	1.456	1.415	1.437	1.735	1.694	1.713	1.750	1.711	1.749
Na	0.597	0.788	0.564	0.614	0.610	0.625	0.544	0.585	0.563	0.265	0.306	0.287	0.250	0.289	0.251
B-site	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
Na	0.516	0.514	0.628	0.581	0.598	0.576	0.647	0.618	0.598	0.393	0.371	0.372	0.408	0.367	0.364
K	0.162	0.216	0.198	0.211	0.173	0.210	0.194	0.195	0.215	0.232	0.218	0.210	0.233	0.233	0.225
A-site	0.678	0.730	0.826	0.792	0.771	0.786	0.841	0.813	0.812	0.625	0.589	0.582	0.642	0.599	0.589
Cations	15.678	15.730	15.826	15.792	15.771	15.786	15.841	15.813	15.812	15.625	15.589	15.582	15.642	15.599	15.589
Г	0.311	0.420	0.440	0.428	0.529	0.503	0.483	0.486	0.459	0.243	0.326	0.266	0.351	0.344	0.309
C	0.010	0.013	0.020	0.015	0.013	0.016	0.016	0.014	0.017	0.066	0.067	0.072	0.078	0.084	0.061
Oxygens	23.000	23.000	23.049	23.000	23.000	23.021	23.083	23.022	23.042	23.000	23.000	23.000	23.000	23.000	23.000

A.3 El	ectron mic	roprobe a	analyses	of amphil	boles fro	m the gra	anites of	the Wen	tworth pl	uton (co	ntinued).			
	7814	7814	7814	7814	7814	7814	7814	7814	7814	7814	7814	7814	7814	7814
	28	29	30	31	32	34	36	44	48	62	63	73	74	78
	44.40	44.99	44.43	45.67	45.14	44.97	45.15	44.29	44.97	44.48	44.23	45.04	44.20	44.99
	1.01	1.20	0.78	0.75	1.07	0.73	1.00	1.11	1.02	1.00	0.86	0.95	0.93	1.02
	5.98	6.26	6.38	6.49	6.60	5.86	6.20	6.46	5.83	6.53	6.16	6.51	6.24	5.86
-	27.54	27.54	27.32	27.67	27.63	28.77	27.67	27.85	27.90	28.13	28.50	27.87	27.75	27.22
	1.39	1.35	1.38	1.33	1.37	1.39	1.23	1.32	1.42	1.44	1.52	1.41	1.45	1.27
	4.36	4.34	4.21	4.46	4.22	4.62	4.48	4.27	4.34	3.96	3.95	3.63	3.98	4.75
_	10.32	10.58	10.32	10.45	10.24	9.82	10.45	10.41	10.26	9.86	10.30	10.18	10.36	9.84
	1.97	2.01	2.12	2.16	2.13	2.16	1.98	2.10	2.22	2.11	1.87	2.11	2.10	1.95
	1.14	1.13	1.15	1.12	1.09	1.05	1.05	1.18	1.07	1.11	1.07	1.17	1.14	1.12
	0.44	0.59	0.45	0.58	0.49	0.65	0.43	0.49	0.70	0.58	0.36	0.38	0.58	0.56
	0.28	0.28	0.27	0.24	0.22	0.24	0.22	0.25	0.31	0.25	0.25	0.27	0.24	0.28
0	98.83	100.28	98.81	100.91	100.19	100.24	99.85	99.71	100.03	99.45	99.05	99.52	98.96	98.87
2	la calculat	ed on the	basis of 2	23 oxygen	is and Fe ³	^{tt} estimat	ted with t	he 13eCN	VK norm	alization				
5	6.962	6.970	6.973	7.002	6.963	6.916	6.983	6.895	6.993	6.919	6.916	7.025	6.956	6.997
\mathbf{c}	1.038	1.030	1.027	0.998	1.037	1.061	1.017	1.105	1.007	1.081	1.084	0.975	1.044	1.003
0	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
4	0.066	0.112	0.153	0.174	0.162	0.000	0.112	0.080	0.060	0.115	0.050	0.222	0.111	0.071
4	0.439	0.303	0.346	0.356	0.390	0.829	0.407	0.425	0.409	0.590	0.602	0.256	0.350	0.600
2	0.119	0.139	0.092	0.086	0.124	0.084	0.116	0.130	0.120	0.117	0.101	0.112	0.109	0.120
4	1.019	1.003	0.986	1.019	0.970	1.059	1.032	0.992	1.005	0.919	0.921	0.843	0.933	1.102
9	3.173	3.265	3.240	3.191	3.174	2.848	3.172	3.200	3.219	3.070	3.125	3.380	3.303	2.941
∞	0.185	0.178	0.183	0.173	0.179	0.181	0.161	0.174	0.187	0.189	0.201	0.187	0.194	0.167
0	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000	5.000
4	1.734	1.755	1.735	1.717	1.693	1.619	1.732	1.737	1.709	1.643	1.726	1.701	1.747	1.640
9	0.266	0.245	0.265	0.283	0.307	0.381	0.268	0.263	0.291	0.357	0.274	0.299	0.253	0.360
0	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000	2.000
\mathbf{c}	0.334	0.358	0.379	0.359	0.329	0.263	0.327	0.370	0.377	0.280	0.291	0.340	0.388	0.230
S	0.228	0.224	0.230	0.218	0.214	0.205	0.207	0.234	0.213	0.220	0.214	0.232	0.229	0.222
s	0.562	0.582	0.608	0.577	0.543	0.468	0.534	0.604	0.590	0.500	0.505	0.572	0.617	0.452
ž	3 15.562	15.582	15.608	15.577	15.543	15.468	15.534	15.604	15.590	15.500	15.505	15.572	15.617	15.452
\mathbf{c}	0.216	0.289	0.223	0.279	0.240	0.317	0.210	0.239	0.343	0.285	0.176	0.186	0.288	0.275
	0.074	0.073	0.073	0.063	0.057	0.062	0.057	0.065	0.081	0.067	0.067	0.072	0.064	0.073
Ξ	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000	23.000

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Appendix A.4 Zoning profiles for selected traverses on grains of (A) sodic and (B) calcic amphiboles in the granites of the Wentworth pluton.

Comple	2026	2026	9909	9909	9909	2026	2026	2026	2026	2026	9202	9202		2026
Analysis	238 238	239 239	240 240	246 246	247	250	251	252	0000 271	272	273 273	5 7	5 4 L	174 275
SiO ₂	37.34	37.04	37.19	35.73	36.85	37.01	37.35	37.09	36.88	36.25	36.30	36.3	80	80 36.31
TiO ₂	4.12	3.78	4.09	9.09	5.32	3.55	3.28	3.24	4.35	3.77	3.58	4	50	26 4.02
Al ₂ O ₃	13.27	13.37	13.13	11.80	13.12	14.02	14.53	14.57	13.39	13.14	13.63	13.5	∞	8 13.36
FeO	23.81	23.95	24.21	17.79	20.82	23.65	23.10	23.38	24.46	23.75	24.01	24.2	∞	8 25.08
MnO	0.19	0.18	0.20	0.15	0.18	0.22	0.20	0.19	0.18	0.18	0.17	0.1	6	9 0.19
MgO	8.93	8.79	8.48	7.81	8.88	8.03	8.45	7.69	8.45	8.57	8.28	8.4	4	4 8.03
CaO	0.01	0.03	0.03	7.04	3.00	0.12	0.04	0.10	0.08	0.07	0.09	0.0	4	4 0.05
Na ₂ O	0.14	0.14	0.15	0.10	0.09	0.12	0.07	0.10	0.13	0.20	0.13	0.0	S	5 0.10
K_2O	9.58	9.54	9.51	5.89	8.34	9.36	9.63	9.43	9.74	8.93	9.03	9.7	\sim	7 9.09
Ľ	0.43	0.41	0.40	0.42	0.52	0.32	0.33	0.29	0.38	0.42	0.42	0.3	∞	8 0.40
C	0.35	0.40	0.42	0.20	0.24	0.29	0.28	0.27	0.34	0.35	0.31	0.3	4	4 0.34
0≡F,Cl	0.26	0.26	0.26	0.22	0.27	0.2	0.2	0.18	0.24	0.26	0.25	0.2	4	4 0.24
H_2O	1.59	1.57	1.57	1.63	1.57	1.63	1.65	1.65	1.60	1.54	1.56	1.6	0	0 1.57
Total	100.01	99.45	99.63	97.87	99.19	98.48	99.12	98.17	100.23	97.42	97.74	6 .00	7	7 98.79
Structura	al formula	ie on the	basis of 2	2 oxygens	7									
Si	5.964	5.959	5.98	5.703	5.873	5.981	5.977	5.997	5.905	5.948	5.937	5.90	\sim	2 5.908
Al ^{IV}	2.036	2.041	2.02	2.218	2.127	2.019	2.023	2.003	2.095	2.052	2.063	2.09	∞	8 2.092
Al ^{VI}	0.461	0.493	0.466	0	0.335	0.649	0.716	0.771	0.43	0.487	0.563	0.46°		7 0.468
Ti	0.495	0.458	0.495	1.092	0.638	0.431	0.395	0.394	0.524	0.466	0.44	0.51_{4}	. +	4 0.492
Fe^{2+}	3.180	3.222	3.255	2.375	2.775	3.196	3.092	3.162	3.275	3.259	3.283	3.257		7 3.412
Mn	0.026	0.025	0.027	0.02	0.024	0.03	0.027	0.025	0.025	0.024	0.023	0.026	5	0.026
Mg	2.126	2.109	2.032	1.859	2.109	1.934	2.017	1.853	2.017	2.096	2.018	2.018	\sim	3 1.948
Ca	0.002	0.005	0.004	1.203	0.512	0.021	0.007	0.018	0.013	0.012	0.015	0.00	5	5 0.009
Na	0.042	0.044	0.046	0.032	0.028	0.037	0.023	0.032	0.04	0.063	0.04	0.01:	5	5 0.033
K	1.953	1.958	1.952	1.2	1.696	1.929	1.966	1.945	1.99	1.87	1.885	1.999	6) 1.888
Cations	16.285	16.314	16.277	15.702	16.117	16.227	16.243	16.2	16.314	16.277	16.267	16.302	•	2 16.276
Ŀ	0.429	0.416	0.404	0.421	0.52	0.322	0.33	0.301	0.389	0.438	0.43	0.38_{4}	+	4 0.407
CI	0.187	0.216	0.23	0.109	0.131	0.156	0.154	0.146	0.186	0.195	0.173	0.187		7 0.189
HO	1.692	1.684	1.683	1.735	1.674	1.761	1.758	1.776	1.712	1.684	1.698	1.714		1.702

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Sample	5056	5056	5056	5056	6490	6490	6490	6490	6490	6490	6490	7658	7658	7658
Analysis	318	328	330	340	314	318	361	362	386	394	431	276	278	346
SiO ₂	35.71	36.71	36.25	36.84	37.22	38.45	38.27	37.95	38.53	37.95	37.31	36.15	34.72	35.38
TiO_2	4.02	4.12	4.08	3.66	1.86	2.51	2.66	2.77	2.41	2.23	2.32	3.72	3.61	4.06
Al_2O_3	13.12	13.11	13.40	13.86	14.19	14.68	13.60	13.49	14.26	15.08	13.78	12.79	13.10	12.63
FeO	24.91	24.92	24.25	25.09	24.98	24.76	24.83	24.86	24.73	24.96	25.87	30.09	31.13	29.84
MnO	0.15	0.20	0.18	0.15	0.51	0.52	0.40	0.45	0.51	0.50	0.59	0.26	0.34	0.35
MgO	8.03	8.14	7.93	7.91	7.42	6.28	7.41	7.24	69.9	6.82	7.48	4.19	4.33	4.22
CaO	0.01	0.03	0.06	0.01	0.04	0.03	0.02	0.04	0.05	0.06	0.08	0.02	0.04	0.02
Na ₂ O	0.09	0.13	0.08	0.09	0.06	0.09	0.06	0.08	0.04	0.05	0.08	0.04	0.09	0.02
$\mathbf{K}_{2}\mathbf{O}$	9.28	9.57	9.35	9.44	8.32	9.38	9.35	8.91	9.53	8.29	7.92	9.01	8.06	9.19
Ŧ	0.36	0.35	0.31	0.35	0.46	0.42	0.63	0.63	0.41	0.28	0.41	0.19	0.14	0.19
CI	0.35	0.36	0.33	0.35	0.14	0.14	0.10	0.12	0.18	0.12	0.12	0.27	0.33	0.34
O≡F,CI	0.23	0.23	0.2	0.23	0.23	0.21	0.29	0.29	0.21	0.14	0.2	0.14	0.13	0.15
H_2O	1.56	1.60	1.61	1.60	1.57	1.64	1.54	1.52	1.63	1.70	1.61	1.64	1.63	1.61
Total	97.82	99.47	98.03	99.58	97.01	99.10	99.16	98.34	99.17	98.16	97.77	98.49	97.66	97.98
Structural	formulae	on the b	asis of 22	oxygens										
Si	5.882	5.934	5.925	5.935	6.11	6.173	6.165	6.16	6.189	6.113	6.087	6.027	5.868	5.955
Al ^{IV}	2.118	2.066	2.075	2.065	1.89	1.827	1.835	1.84	1.811	1.887	1.913	1.973	2.132	2.045
AIVI	0.428	0.431	0.504	0.565	0.853	0.949	0.744	0.74	0.887	0.973	0.734	0.537	0.475	0.459
Ti	0.498	0.501	0.501	0.443	0.229	0.304	0.322	0.338	0.291	0.27	0.285	0.466	0.459	0.514
Fe^{2+}	3.431	3.369	3.314	3.380	3.429	3.325	3.345	3.374	3.322	3.362	3.530	4.194	4.400	4.201
Mn	0.021	0.028	0.025	0.021	0.071	0.071	0.055	0.062	0.069	0.068	0.082	0.036	0.048	0.049
Mg	1.971	1.961	1.933	1.901	1.816	1.503	1.779	1.752	1.602	1.637	1.819	1.042	1.092	1.058
Ca	0.002	0.004	0.011	0.001	0.008	0.005	0.004	0.007	0.008	0.01	0.013	0.003	0.007	0.004
Na	0.029	0.041	0.024	0.027	0.018	0.026	0.017	0.025	0.013	0.015	0.026	0.011	0.03	0.007
K	1.949	1.975	1.95	1.941	1.743	1.921	1.92	1.846	1.953	1.703	1.648	1.916	1.738	1.973
Cations	16.329	16.31	16.262	16.279	16.167	16.104	16.186	16.144	16.145	16.038	16.137	16.205	16.249	16.265
Ы	0.374	0.358	0.317	0.36	0.48	0.421	0.642	0.642	0.414	0.281	0.426	0.195	0.145	0.198
CI	0.197	0.197	0.181	0.191	0.08	0.074	0.055	0.066	0.098	0.064	0.069	0.153	0.191	0.194
НО	1.714	1.723	1.751	1.725	1.72	1.752	1.652	1.646	1.744	1.827	1.753	1.826	1.832	1.804

Sample	7658	7658	7658	7658
Analysis	347	348	351	354
SiO ₂	35.22	34.99	35.96	35.72
TiO ₂	4.02	3.93	3.82	3.79
Al ₂ O ₃	12.49	12.90	12.55	11.89
FeO	30.74	30.39	29.03	30.32
MnO	0.35	0.35	0.32	0.36
MgO	4.08	4.13	4.72	4.35
CaO	0.02	0.04	0.03	0.02
Na2O	0.06	0.08	0.00	0.05
$\mathbf{K_2O}$	9.15	9.14	9.00	9.02
Ч	0.22	0.10	0.15	0.18
CI	0.33	0.29	0.31	0.31
0≡F,Cl	0.16	0.11	0.13	0.15
H_2O	1.59	1.66	1.64	1.61
Total	98.42	98.10	97.66	97.78
Structural fo	ormulae on	the basis of	22 oxygens	
Si	5.930	5.895	6.027	6.035
AIIV	2.070	2.105	1.973	1.965
AIVI	0.407	0.455	0.505	0.401
Ti	0.509	0.498	0.482	0.481
${\rm F}e^{2+}$	4.329	4.282	4.069	4.284
Mn	0.050	0.050	0.045	0.051
Mg	1.025	1.037	1.179	1.097
Ca	0.003	0.007	0.005	0.004
Na	0.019	0.025	0.001	0.017
K	1.965	1.965	1.923	1.945
Cations	16.307	16.319	16.209	16.280
F	0.230	0.102	0.154	0.196
CI	0.187	0.165	0.178	0.179
HO	1.791	1.866	1.834	1.813

APPENDIX B

Whole-rock analytical data, estimated magmatic parameters, partial melt and fractionation modelling methodologies for the late Paleozoic A-type granites of the Cobequid Highlands are presented here as supplementary data for Chapter 3. Petrographic description and whole-rock analyses of the Neoproterozoic rocks used as sources in the geochemical models are included as well.

	biotite	WP	7372		76.2	0.22	11.99	1.82	0.04	0.31	0.3	3.63	4.81	0.03	0.46		129	178	26	79.3	492	68	14	24	34	30	b.d.	6	7	51.4	121
	biotite	WP	7630		72.76	0.3	13.87	2.39	0.05	0.25	0.97	3.8	5.17	0.06	0.53		395	182	65	60.2	394	51.4	94	20	50	b.d.	b.d.	٢	b.d.	72.9	137
- -	biotite	WP	6970		74.67	0.27	12.93	2.18	0.04	0.09	0.34	3.69	5.39	0.03	0.55		257	240	36	70	372	61	S	27	52	8	b.d.	b.d.	b.d.	n.d.	n.d.
	biotite	WP	9827b		78.23	0.126	11.92	1.51	0.015	0.13	0.44	3.4	4.8	0.02	0.13		64	226	36	55.9	273	56.4	9	31	17	10	5	7	b.d.	33.2	97.2
	biotite	WP	<u>4648</u>		73.62	0.21	13.58	2.22	0.05	0.84	0.52	3.95	5.21	0.04	0.4		223	254	37	6.69	328	68.9	18	23	30	b.d.	b.d.	b.d.	b.d.	67.4	157
-	biotite	PHP	7378		72.05	0.28	13.39	3.26	0.03	0	0.44	4.57	4.84	0.02	0.2		567	142	45	92	582	41	19	24	74	15	b.d.	b.d.	S	n.d.	n.d.
	biotite	PHP	7391		70.96	0.54	14.16	3.1	0.03	0.33	1.47	4.61	4.6	0.09	0.3		702	110	124	54	521	28	b.d.	17	٢	S	7	20	8	n.d.	n.d.
	biotite	PHP	4672		75.16	0.25	12.64	2.42	0.02	0.83	0.26	3.74	4.79	0.04	0.4		186	211	26	111	337	50	20	25	22	b.d.	10	9	44	109	123.7
	biotite	PHP	7385		70.11	0.39	14.09	2.98	0.04	0.4	1.45	3.78	4.77	0.07	0.3		723	175	103	46	318	24	17	19	14	b.d.	9	22	14	n.d.	n.d.
	biotite	PHP	4694		73.59	0.24	13.16	2.45	0.02	0.88	0.54	3.85	5.04	0.05	0.7		188	298	34	84	301	42	13	23	22	b.d.	6	9	44	n.d.	n.d.
•	biotite	PHP	6262		71.2	0.41	13.2	2.79	0.07	0.38	1.4	4.91	4.6	0.08	0.85		582	185	48	50	372	25	b.d.	16	36	b.d.	b.d.	22	13	n.d.	n.d.
: -	biotite	PHP	7397		74.45	0.25	13.15	2.07	0.01	0.05	0.53	3.85	5.03	0.04	0.8		265	147	47	44	269	35	16	19	16	b.d.	9	11	9	n.d.	n.d.
	biotite	PHP	4686		75.29	0.19	12.86	1.95	0.02	0.75	0.29	3.58	4.87	0.03	0.3		180	213	45	60	228	43	14	22	20	b.d.	9	7	33	63.41	115
	biotite	CCP	1897		73.97	0.15	13.57	1.4	0.01	0.25	0.86	3.73	5.2	0.08	0.69		128	137	30	51	334	34	16	21	12	b.d.	15	7	26	n.d.	n.d.
•	biotite	CCP	2039	<u>()</u>	72.64	0.33	14.02	2.73	0.02	0.39	0.34	3.68	5.38	0.11	0.46	<u>(m</u>	471	179	52	42	298	27	17	21	11	1	16	9	26	n.d.	n.d.
	biotite	CCP	2616	des (wt. ⁰	75.2	0.11	13.5	1.53	0.03	0.52	0.21	5.19	3.35	0.02	0.2	nents (pp	117	115	27	79	241	61	17	26	51	b.d.	37	b.d.	7	n.d.	n.d.
Granite	type	Pluton	Sample	Major oxi	SiO_2	TiO_2	Al_2O_3	$\mathrm{Fe_2O_3}$	MnO	MgO	CaO	Na_2O	K_2O	P_2O_5	IOI	Trace eler	Ba	Rb	Sr	Y	Zr	Nb	Pb	Ga	Zn	Cu	Ni	^	Cr	La	Ce

biotite	WP	7372	14	52.1	12.6	0.57	13.1	2.41	15	3.21	9.03	1.55	10.1	1.38	107	1.3	11.7	b.d.	n.d.	7.18	19.5	4.27		884.0	15.26	4.24		30.6	31.4	15.5	25.8
biotite	WP	7630	15.7	53.2	10.9	1.03	10.4	1.84	11.1	2.3	6.42	1.07	6.91	0.94	67	1.8	9.1	b.d.	n.d.	4.32	30.2	5.6		870.5	13.98	5.77		21.2	29.7	24.0	24.9
biotite	WP	6970	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	18	n.d.		866.5	12.21	4.62		30.6	24.8	17.3	24.2
biotite	WP	9827b	9.44	33.3	8.41	0.333	7.99	1.74	10.9	2.29	6.93	1.16	8.33	1.45	b.d.	2.4	8.5	n.d.	1	5.8	19.7	4.43		840.3	17.53	5.31		27.9	26.4	18.7	24.3
biotite	WP	<u>4648</u>	15.7	53.8	11.7	0.78	11.2	2.08	12.8	2.66	7.67	1.28	8.14	1.15	b.d.	9	8.6	b.d.	n.d.	4.58	30.5	5.93		832.0	13.60	6.44		16.7	21.9	21.6	20.0
biotite	PHP	7378	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	18	n.d.		903.2	14.23	4.87		30.6	42.0	25.8	32.8
biotite	PHP	7391	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	20	n.d.		879.2	11.50	5.70		27.5	54.2	37.7	39.8
biotite	PHP	4672	n.d.	81.94	17.18	1.49	n.d.	3.51	n.d.	n.d.	n.d.	n.d.	12.02	1.62	0.63	2.42	12.7	n.d.	2.13	5.53	23	4.32		860.8	13.64	5.18		23.9	28.2	21.1	24.4
biotite	PHP	7385	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	17	n.d.		841.9	7.35	5.66		32.4	34.1	44.0	36.8
biotite	PHP	4694	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	22	n.d.		843.0	12.17	5.78		25.0	20.0	25.1	23.4
biotite	PHP	6262	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	21	n.d.		833.8	12.96	6.65		26.2	32.2	42.2	33.5
biotite	PHP	7397	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	16	n.d.		834.5	13.97	5.93		34.4	40.5	30.2	35.0
biotite	dHd	4686	n.d.	45.71	8.9	0.59	n.d.	1.69	n.d.	n.d.	n.d.	n.d.	7.09	0.99	0.84	1.65	8.91	n.d.	2.16	4.4	25	4.78		826.5	11.07	5.43		22.0	28.0	24.6	24.8
biotite	CCP	1897	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	17	n.d.		853.0	12.83	5.70	(0)	32.4	43.5	31.1	35.6
biotite	CCP	2039	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	19	n.d.	S	850.6	9.27	5.23	d melt (%	28.9	33.3	39.1	33.8
biotite	CCP	2616	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	22	n.d.	<u>arameter</u>	828.3	3.32	4.68	f extracte	25.0	51.8	17.3	31.4
Granite type	Pluton	Sample	\mathbf{Pr}	Nd	Sm	Eu	Gd	Tb	Dy	Но	Er	Tm	Yb	Lu	Co	C_S	Hf	\mathbf{Sb}	Sc	Та	Th	N	Intensive t	$T_{Zr}(^{\circ}C)$	P _{plg-melt} (kbar)	${\rm H_2O_{melt}}$	Fraction o	Th	Rb	Nb	Average

Highlands (co	ntinued).															
Granite	biotite	biotite	calcic													
type Pluton	WP	WP	дНр	дНр	dHd	dHd	дНд	WP	WP	WP	WP	WР	WP	WР	WP	WP
Sample	8010	9827a	4689	7393	6260	4670	4681	4645	4652	4642	4640	35-5-1	4663	4651	4650	4626
Major oxides	(wt. %)															
SiO_2	75.76	77.23	72.45	72.72	70.3	72.42	70.77	71.21	73.52	71.78	70.88	73.35	72.49	73.48	73.03	73.56
TiO_2	0.21	0.211	0.38	0.32	0.25	0.45	0.48	0.34	0.2	0.43	0.49	0.23	0.25	0.22	0.2	0.32
Al_2O_3	11.5	11.84	13.17	13.88	13.3	13.14	13.18	14.5	13.61	13.55	13.87	13.1	13.43	13.45	13.75	12.95
Fe_2O_3	2.47	2.37	3.23	2.63	3.06	3.2	4.09	2.18	2.1	3.27	3.49	2.14	2.4	1.94	2.33	2.51
MnO	0.03	0.016	0.07	0.03	0.09	0.05	0.07	0.06	0.06	0.06	0.07	0.04	0.05	0.03	0.06	0.04
MgO	0.14	0.25	0.99	0.2	0.15	1.15	1.13	1	0.77	1.1	1.01	0.82	0.9	1.04	0.83	0.98
CaO	0.41	0.57	0.65	0.7	0.81	0.84	0.91	1.04	0.64	0.74	1.29	0.46	0.74	0.34	0.57	0.61
Na_2O	3.7	3.86	3.89	4.62	4.84	3.39	3.45	4.06	4.05	3.59	4.65	4.29	4.09	4.28	3.84	3.68
$\rm K_2O$	4.64	3.7	5.11	5.02	4.82	5.02	5.49	4.95	5.21	4.94	4.39	5.11	5.04	5.03	5.14	5.31
P_2O_5	0.02	0.02	0.07	0.05	0.04	0.09	0.09	0.08	0.04	0.1	0.11	0.03	0.06	0.04	0.05	0.05
LOI	0.5	0.26	0.3	0.1	0.35	0.3	0.2	0.1	0.2	0.8	0.4	0.1	0.6	0.5	0.2	0.3
Trace elemen	its (ppm)															
Ba	83	113	223	496	655	368	360	570	316	555	642	206	254	158	204	287
Rb	228	152	262	174	177	176	206	143	232	210	166	254	230	211	240	223
Sr	29	63	27	47	50	57	51	152	38	88	109	35	54	30	39	59
Y	98	80.3	88	59	101	89	78	38	LL	50	48	85	75.1	124	84	86
Zr	566	503	361	392	585	472	831	198	274	250	295	292	403	298	286	325
Nb	75	57.7	48	30	42	46	37	20	64.6	19	23	54	63.7	56	51	45
Pb	31	4	45	13	14	b.d.	б	21	18	23	8	25	22	b.d.	8	24
Ga	29	32	24	21	25	25	25	18	23	19	19	25	21	23	25	24
Zn	74	20	80	16	116	30	55	36	34	50	51	58	50	27	32	51
Cu	14	16	2	5	9	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	9	b.d.	b.d.	b.d.	b.d.
Ni	b.d.	9	6	b.d.	b.d.	7	12	8	8	6	9	7	8	6	6	8
^	b.d.	11	6	9	b.d.	17	10	16	1	9	15	4	8	4	1	10
Cr	7	b.d.	29	13	10	24	37	39	27	40	32	32	24	34	40	24
La	38	73.2	77.76	n.d.	n.d.	80.51	91.18	n.d.	81.7	37.8	41.4	n.d.	92.7	n.d.	87.8	61.8
Ce	200	175	168.98	n.d.	n.d.	160.04	213.09	n.d.	144	67	76	n.d.	176	n.d.	136	121

Appendix B. 1 Whole-rock geochemistry, estimated intensive parameters and fractions of extracted melt for the biotite and calcic granites from the Cobequid

calcic	WP 4626	n.d.	48	10.8	0.99	n.d.	2.1	n.d.	n.d.	n.d.	n.d.	8.81	1.34	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	24	4.4		847.1		10.09	5.20		22.9	26.7	23.5	24.4
calcic	WP 4650	n.d.	62	14.2	0.96	n.d.	7	n.d.	n.d.	n.d.	n.d.	8.51	1.31	2.1	5.6	10.5	n.d.	3.06	4.5	21	5.4		840.8		8.98	5.42		26.2	24.8	20.7	23.9
calcic	WP 4651	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	19	n.d.		840.6		11.46	5.77		28.9	28.2	18.9	25.4
calcic	WP 4663	19.2	62.1	12.3	0.82	12.3	2.19	13.5	2.84	8.04	1.32	8.4	1.12	1	4	10.2	b.d.	n.d.	4.31	31	6.23		838.4		8.77	5.47		21.2	23.7	22.5	22.4
calcic	WP 35-5-1	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	20	n.d.		833.0		12.40	6.01		27.5	23.5	19.6	23.5
calcic	WP 4640	n.d.	33	7.92	1.47	n.d.	1.1	n.d.	n.d.	n.d.	n.d.	4.6	0.73	4.4	2.4	9.7	n.d.	6.95	1.9	14	4.6		824.3		6.99	6.14		39.3	35.9	45.9	40.4
calcic	WP 4642	n.d.	31	7.16	1.24	n.d.	1	n.d.	n.d.	n.d.	n.d.	4.3	0.67	4	4.7	8.3	n.d.	6.05	1.8	29	4		827.9		7.23	5.66		19.0	28.4	55.6	34.3
calcic	WP 4652	18.5	64.7	13.8	0.78	13.3	2.35	14.2	2.86	8.17	1.33	8.44	1.18	b.d.	5.4	7	b.d.	n.d.	4.85	26.8	6.25		828.8		10.22	5.85		20.4	23.1	20.7	21.4
calcic	WP 4645	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	21	n.d.		801.3		8.19	6.81		26.2	41.7	52.8	40.2
calcic	PHP 4681	n.d.	86.73	17.18	1.72	n.d.	2.95	n.d.	n.d.	n.d.	n.d.	8.86	1.27	3.21	3.15	22.42	n.d.	4.15	3.87	17	3.6		940.6		13.05	4.13		32.4	28.9	28.5	29.9
calcic	PHP 4670	n.d.	73.93	15.4	1.49	n.d.	2.94	n.d.	n.d.	n.d.	n.d.	9.95	1.37	2.38	2.42	14.47	n.d.	4.29	4.07	18	2.95		889.0		11.12	4.78		30.6	33.9	23.0	29.1
calcic	9260	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	20	n.d.		889.4		14.70	5.52		27.5	33.7	25.1	28.8
calcic	7393	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	20	n.d.		859.1		12.15	5.66		27.5	34.3	35.2	32.3
calcic	PHP 4689	n.d.	68.45	13.67	1.12	n.d.	2.62	n.d.	n.d.	n.d.	n.d.	9.3	1.32	3.3	3.08	12.6	n.d.	3.68	5.05	21	4.22		855.7		8.32	4.97		26.2	22.7	22.0	23.6
biotite	WP 9827a	18.7	64.8	13.9	0.426	12.2	2.4	14.8	3.12	9.39	1.54	10.9	1.77	b.d.	2.3	13	n.d.	7	6.58	25.5	5.24		902.6		20.61	4.90	lt (%)	21.6	39.2	18.3	26.4
biotite	WP 8010	n.d.	89	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	101	n.d.	n.d.	n.d.	11	n.d.	23	7	neters	905.7		16.79	3.93	racted me	23.9	26.1	14.1	21.4
Granite type	Pluton Samule	Pr	Nd	Sm	Eu	Gd	Tb	Dy	Но	Er	Tm	Yb	Lu	Co	\mathbf{Cs}	Hf	\mathbf{Sb}	Sc	Та	Th	U	Intensive parai	$T_{Zr}(^{\circ}C)$	P plg-melt	(kbar)	$\mathrm{H_2O}_{\mathrm{melt}}$	Fraction of ext	Th	Rb	Nb	Average

Granite	aalaia	aalaia	aalaia	aalaia	aalaia	aalaia	aalaia	aalaia	aalaia	aalaia	aalaia
type	calcic	calcic	calcic	calcic	calcic	calcic	calcic	calcic	calcic	calcic	calcic
Pluton	WP	WP	WP	WP	WP	WP	WP	WP	WP	WP	WP
Sample	4657	36-7-1	7658	7373	<u>4641</u>	6808	9819	7374	8003	4614	4615
Major oxides (wt. %)										
SiO ₂	75.81	73.83	72.62	75.72	70.04	73.58	73.09	75.44	70.07	74.87	74.49
TiO ₂	0.21	0.23	0.32	0.25	0.45	0.22	0.589	0.27	0.49	0.24	0.31
Al ₂ O ₃	12.32	13.13	13.6	12.09	14.25	13.6	13.29	11.86	14.55	12.55	12.55
Fe ₂ O ₃	2.3	2.04	2.72	2.35	3.48	2.57	2.41	3.08	3.18	2.25	1.46
MnO	0.04	0.03	0.08	0.05	0.07	0.02	0.085	0.03	0.05	0.04	0.04
MgO	0.7	1.06	0.24	0.28	1.05	0.15	0.71	0.25	0.38	0.82	1.01
CaO	0.31	0.43	0.9	0.71	0.92	0.45	1.98	0.73	0.83	0.77	1.37
Na ₂ O	3.57	4.1	3.72	3.78	2.07	4.07	4.2	2.89	4.5	3.66	3.21
K ₂ O	5.92	4.79	5.24	4.69	5.36	4.87	4.17	5.49	5.4	5.32	5.77
P_2O_5	0.02	0.05	0.07	0.03	0.1	0.02	0.13	0.03	0.06	0.03	0.05
LOI	0	0.3	0.57	0.29	0.9	0.74	0.34	0.32	0.3	0.2	0.3
Trace elements	s (ppm)	010	010 /	0>	015	0., .	0.00	0.02	010	•.=	0.0
Ba	125	179	365	126	684	209	561	310	419	147	187
Rb	169	220	253	171	185	221	108	189	219	242	214
Sr	13	33	65	36	90	29	124	53	64	47	84
Y	85	116	82	78.5	46.3	66	52.2	100	69	93	98
Zr	370	308	383	482	396	414	596	688	707	710	1009
Nb	45	55	47	64	24.3	44	28.8	68.7	43	57	47
Pb	8	15	29	10	27	43	6	19	27	b.d.	18
Ga	25	22	24	24	18	22	18	22	24	25	21
Zn	70	27	81	60	60	63	27	40	59	49	39
Cu	b.d.	2	6	b.d.	b.d.	31	17	15	27	b.d.	b.d.
Ni	6	7	b.d.	b.d.	b.d.	b.d.	10	b.d.	3	10	7
V	b.d.	4	11	11	11	b.d.	24	12	17	1	14
Cr	34	31	b.d.	10	34	b.d.	b.d.	8	4	33	32
La	n.d.	n.d.	n.d.	60.9	40.7	n.d.	67.5	43	43	69.2	n.d.
Ce	n.d.	n.d.	n.d.	136	87	n.d.	142	92.1	138	146	n.d.
Pr	n.d.	n.d.	n.d.	15.8	9.94	n.d.	15.8	14.7	n.d.	n.d.	n.d.
Nd	n.d.	n.d.	n.d.	58.2	36.8	n.d.	57.4	57.9	62	63	n.d.
Sm	n.d.	n.d.	n.d.	13	7.82	n.d.	11.7	15.7	n.d.	13	n.d.
Eu	n.d.	n.d.	n.d.	0.65	1.23	n.d.	1.39	0.89	n.d.	0.65	n.d.
Gd	n.d.	n.d.	n.d.	13.3	7.36	n.d.	10.4	17.5	n.d.	n.d.	n.d.
Tb	n.d.	n.d.	n.d.	2.37	1.31	n.d.	1.81	3.21	n.d.	2.4	n.d.
Dy	n.d.	n.d.	n.d.	14.4	7.75	n.d.	10.3	19.6	n.d.	n.d.	n.d.
Но	n.d.	n.d.	n.d.	3.02	1.62	n.d.	2.03	4.1	n.d.	n.d.	n.d.
Er	n.d.	n.d.	n.d.	8.61	4.7	n.d.	5.54	11.2	n.d.	n.d.	n.d.
Tm	n.d.	n.d.	n.d.	1.44	0.75	n.d.	0.811	1.81	n.d.	n.d.	n.d.
Yb	n.d.	n.d.	n.d.	9.21	5.05	n.d.	5.1	11.3	n.d.	9.93	n.d.
Lu	n.d.	n.d.	n.d.	1.29	0.71	n.d.	0.824	1.53	n.d.	1.54	n.d.
Co	n.d.	n.d.	n.d.	83	2	n.d.	n.d.	81	72	n.d.	n.d.
Cs	n.d.	n.d.	n.d.	1.8	5.2	n.d.	0.7	3.4	n.d.	n.d.	n.d.
Hf	n.d.	n.d.	n.d.	11	8.8	n.d.	13.7	15	n.d.	n.d.	n.d.
Sb	n.d.	n.d.	n.d.	b.d.	b.d.	n.d.	n.d.	b.d.	n.d.	n.d.	n.d.
Sc	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	7	n.d.	12	n.d.	n.d.
Та	n.d.	n.d.	n.d.	5.45	2.08	n.d.	1.8	6.18	n.d.	n.d.	n.d.

Appendix B. 1 Whole-rock geochemistry, estimated intensive parameters and fractions of extracted melt for the biotite and calcic granites from the Cobequid Highlands (continued).

Granite	calcic	calcic	calcic	calcic	calcic	calcic	calcic	calcic	calcic	calcic	calcic
type	culcie	carcie	culcie	calcie	culcic	culcic	calcie	calcie	culcie	culcic	calcie
Pluton	WP	WP	WP	WP	WP	WP	WP	WP	WP	WP	WP
Sample	4657	36-7-1	7658	<u>7373</u>	<u>4641</u>	6808	9819	<u>7374</u>	8003	4614	4615
Th	15	24	29	17.3	19.2	25	12.1	20.1	22	25	25
U	n.d.	n.d.	n.d.	4.43	4.64	n.d.	1.71	5.09	7	6.1	n.d.
Intensive parar	neters										
$T_{Zr}(^{\circ}C)$	856.8	845.6	863.8	871.5	874.2	878.9	889.7	918.7	918.7	923.2	953.5
P _{plg-melt}											
(kbar)	12.93	11.70	12.83	16.58	5.40	14.50	11.19	15.05	12.14	13.36	16.52
H ₂ O _{melt}	4.79	5.62	5.61	4.06	3.97	5.36	5.19	3.43	4.75	3.72	3.68
Fraction of ext	racted m	<u>elt (%)</u>									
Th	36.7	22.9	19.0	36.7	32.4	22.0	45.5	25.0	25.0	22.0	22.0
Rb	35.3	27.1	23.6	33.1	28.1	27.0	55.2	28.4	27.2	24.6	27.9
Nb	23.5	19.2	22.5	18.5	52.8	24.0	36.7	18.2	24.6	18.5	22.5
Average	31.8	23.1	21.7	29.4	37.8	24.3	45.8	23.9	25.6	21.7	24.1

Note: The fractions of extracted melt were estimated through the equation of batch partial melting and using as a source composition the average trace element concentration of Neoproterozoic intermediate rocks from both Jeffers and Bass River blocks. Samples in bold indicate new major and trace element analyses, underlined bold sample numbers indicate new trace element analyses. Abbreviations: b.d.= below detection limit, n.d.= not determined. When not specified, analytical data are from Pe-Piper (1991, 1995, 1998), and Pe-Piper and Piper (1998a).

APPENDIX B.2 METHODOLOGY FOR GEOCHEMICAL MODELS

Fractionation of mafic sources

The fractionation models used in this work include a two-step process. First the changes in major elements through fractionation were modeled by the use of PELE software for Windows (Boudreau, 1999). Whole-rock analyses of selected mafic sources were used as input. PELE uses a list of mineral phases that can crystallize during the fractionation sequence. From the given list certain minerals were excluded such as leucite, nepheline, mellilite, garnet, kalsilite, whitlockite, corundum, perovskite, muscovite, troilite, pyrrhotite, pyrite, sulphide liquid, graphite, calcite, dolomite, anhydrite and immiscible liquid. Although pyrite is reported as an accessory phase in the Wentworth gabbros, the exclusion of this mineral is not considered to significantly compromise the model, since there no S was determined in whole-rock for these rocks, and therefore there was no input for that element. Other parameters involve temperature, pressure, oxygen fugacity, entropy, enthalpy and volume. In the absence of constraints for entropy, enthalpy and volume, an isobaric fractionation was assumed and was tested for a range of pressures from 2 to 0.6 GPa. An initial 0.5 wt.% H₂O was assumed although not measured. An anhydrous composition for the Fountain Lake basalts is assumed from their mineralogy which includes olivine, plagioclase, pyroxene and Feoxides. Fractionation models were performed under different oxygen fugacities ranging from NNO to +3 QFM. For different oxygen fugacities the program will adjust the amounts of FeO and Fe₂O₃ in the whole-rock input. The range of temperature for all runs was from the liquidus, as determined by the program for a given pressure and oxygen

fugacity, to 800 °C, which is an average zircon-saturation temperature for the studied granites.

After the parameters were set, PELE performed a fractionation simulation until the lowest temperature was achieved. During the simulation, the program predicted mineralogical assemblages that would be in equilibrium with the remaining melt for a given temperature. For each equilibrium stage (or "step" as reported by the program) the following are predicted: mass of liquid, mass of solids, fractionating mineral phases (chemistry and mass for each phase), whole-rock chemistry of the remaining melt, viscosity, volume, enthalpy, entropy and oxygen fugacity. Any invalid parameters or unrealistic set of parameters would crash the program.

The most realistic output for the fractionation models was considered to be from the run at 0.6 GPa, under an oxygen fugacity of -2 QFM. The fractionated minerals predicted were olivine, plagioclase, pyroxene, apatite, and Fe-Ti oxides. The minerals predicted to remain in the melt at the latest stages were alkali feldspar, quartz, titanite, ilmenite, and aenigmatite. Since all the predicted fractionating minerals reflect the mineralogy of the mafic rocks, the output was considered reliable. The predicted viscosity was comparable with the viscosity estimates for the sodic granites, using the method of Giordano et al. (2008). The felsic melt under these conditions was produced in ten steps.

As a second procedure the trace elements were modeled using Rayleigh fractionation (Rayleigh, 1896) as applied by Neuman et al. (1954) which is described by the equation

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$$C_{L} = C_{O} F^{(D-1)} \tag{1}$$

Where C_L : concentration of element in remaining liquid, C_O : concentration of element in the parent melt, F: fraction of remaining melt defined as M_L/M_O (M_L : mass of original melt, M_O : mass of remaining melt), D: the bulk partition coefficient for element calculated as $\Sigma X_A D_i$ (X_A : weight fraction of mineral A in the rock, D_i : partition coefficient of element i in mineral A).

Trace elements were therefore modeled for each step produced by PELE, where D was estimated by the modal amounts of the predicted fractionating minerals and the chemistry of the remaining melt, and F was calculated by the estimated masses of the original and remaining melts. Since the produced melts varied from mafic to felsic, the partition coefficients used, varied accordingly. For produced melts with SiO₂ up to 52 wt.% the partition coefficients used were for basaltic melts. For melts from 52 to 54 wt.% the partition coefficients used were those for basaltic andesite. For melts with 54 to 62 wt.% SiO₂ partition coefficients for andesite were used. For melts with SiO₂ ranging from 62 to 69 wt.% partition coefficients for dacites were used. Melts more silicic than 69 wt.% were considered rhyolitic and the appropriate partition coefficients were used (Table DR2).

Partial melting models

Batch partial melting was modeled by the equation

$$C_L/C_0=1/(D(1-F)+F)$$
 (2)
Where C_L: the concentration of element in extracted melt, C₀: concentration of element at the source, D: bulk partition coefficient as defined above, F: fraction of extracted melt. For highly incompatible elements such as Rb, Nb, and Th, the above equation may be simplified as:

$$C_L/C_0 = 1/F \tag{3}$$

which if rearranged will be:

$$F = C_0 / C_L \tag{4}$$

Using the above simplified equation for selected highly incompatible elements, and their average concentrations in Neoproterozoic intermediate rocks, the degrees of partial melting were estimated. The rest of the trace elements were modeled with the original equation for the range of F values determined by the highly incompatible elements.

		н	3asalt			Basaltic	andesite	
Mineral	pyroxene	plagioclase	olivine	magnetite	pyroxene	plagioclase	olivine	magnetite
Ba	0.03^{-1}	0.151 ¹⁵	0.005 ²⁰	0.028^{28}	0.03^{-1}	0.16^{32}	0.005^{20}	0.028 ²⁸
Rb	0.011^{2}	0.024 ¹⁶	0.0133 ¹⁵	0.11^{29}	0.00941^{33}	0.056^{38}	0.017 38	0.11^{29}
\mathbf{Sr}	0.096^{3}	2.845 ¹⁷	0.00025^{21}		0.12 35	3.06^{32}	0.003^{20}	
Y	$0.421^{\ 4}$	0.031 ¹⁷	0.0125^{21}	0.0039^{26}	1 35	0.032^{32}		0.0039 ²⁶
Zr	$0.128^{\ 4}$	0.0127 ¹⁷	0.011^{22}	0.02^{30}	0.4^{35}	0.0009 ³⁸	0.12^{-40}	0.02^{30}
ЧN	0.004^{4}	0.0925 ¹⁷	$0.01^{\ 2}$	0.01^{30}	0.004^{4}	0.045^{38}	0.0065 ³⁸	0.01^{30}
Pb	0.0102 5	0.901 ¹⁷	$0.0001^{\ 2}$		0.0102 5	0.901 ¹⁷		
Ga	0.74^{6}	0.71 ¹⁸	0.04^{23}	2 ²⁹	0.74^{6}	0.71 ¹⁸		2 ²⁹
Zn	0.5 7	0.11^{7}	0.86^{7}	2.6^{29}	0.5 ⁷	0.11 7	1.2 ³⁴	2.6^{29}
Cu	0.36^{6}	0.38 18	0.5^{24}	0.42 ²⁹	0.12^{-18}	0.38 18	0.19^{41}	0.42^{29}
Ni	2.6 8	1.16^{17}	12.8 25	29 ⁶¹	4 ³⁴	0.06^{31}	12.8 25	9.6 ³¹
\mathbf{Cr}	3.8 6	0.191 ¹⁷	1.85^{21}	40^{27}	30 ³⁵	0.075 ³⁸	1.08 ²⁰	40^{27}
La	0.05^{4}	0.258 ¹⁷	0.000012^{21}	0.015 ²⁹	0.11 ³⁵	0.227^{32}		0.015^{29}
Ce	$0.1^{\ 3}$	0.233 ¹⁷	0.0001 ²⁶	0.016^{29}	0.13^{35}	0.085^{32}	0.012 ²⁰	0.016^{29}
\mathbf{Pr}	0.112^{-1}	0.282 ¹⁷	0.000084 ²¹		0.112^{-1}	0.035^{38}	0.009 ³⁸	
рŊ	0.382^{9}	0.276 ¹⁷	0.0003 ²⁶	0.026 ²⁹	0.4 ³⁵	0.114^{32}		0.026 ²⁹
Sm	$0.281^{\ 4}$	0.106^{17}	0.00063 ²¹	0.024 ²⁹	0.6 35	0.083^{32}	0.028 ³⁸	0.024 ²⁹
Eu	0.32 10	0.061 ¹⁷	0.001^{-26}	0.025 ²⁹	0.72^{31}	0.078^{32}	0.023 ³⁸	0.025 ²⁹
Tb	0.28^{-1}	0.102 ¹⁹	0.0015^{26}	0.019^{29}	0.57^{34}	$0.061^{\ 38}$	0.035^{38}	0.019^{29}
Dy	0.42^{4}	0.0318 16	0.002 ²⁶		0.8 35	0.052^{32}	0.034^{38}	
Er	0.36^{4}	0.0194 ¹⁷	0.0025 ²⁶		0.8 35	0.056^{32}	0.045^{38}	
Чb	0.4^{12}	0.141 ¹⁹	0.01^{-26}	0.018 ²⁹	1.1 ³⁵	0.041 ²⁸	0.071 ³⁸	0.018 ²⁹
Lu	$0.439^{\ 4}$	0.0093 17	0.051 ²⁶		0.55^{34}	0.043^{32}	0.11 ³⁸	
Co	1.7^{-13}	1.03 ¹⁷	3.56^{21}	3.4^{29}	1.2 ³⁴	0.07^{39}	6.6^{34}	3.4 ²⁹
Th	0.014^{5}	0.382 ¹⁹	0.00001 ²⁷	0.05^{31}	0.03^{34}	0.19^{38}	0.039^{38}	0.05^{31}
U	0.0146^{14}	0.073 ¹⁶	0.00001 ²⁷	0.11^{29}	0.04^{34}	0.34^{38}		0.11^{29}

Appendix B. 3 Partition coefficients for selected trace elements used in geochemical models.

		And	esite			Dac	ite	
Mineral	pyroxene	plagioclase	olivine	magnetite	pyroxene	plagioclase	olivine	magnetite
Ba	0.1^{39}	0.27^{39}	$0.02^{\ 40}$	0.028 ²⁸	0.131 ¹⁵	1.05^{42}	0.023 ⁵¹	0.07 54
Rb	0.03^{39}	0.039^{15}	$0.01^{\ 40}$	0.11^{29}	0.0366^{15}	$0.041^{\ 48}$	$0.01^{\ 40}$	0.01^{29}
Sr	0.5^{39}	3.4 ³⁹	0.018^{38}		$0.39^{\ 45}$	2.9^{18}	0.31 ⁴²	0.077^{42}
Υ	2.4 ⁴²	0.066^{42}		0.0039 ²⁶	7.6 42	0.51 ⁴²	0.12 ⁴²	0.95^{42}
Zr	0.33 ⁴²	0.15^{42}	$0.12^{\ 40}$	0.02^{-30}	0.23 ⁴⁵	0.55^{42}	$0.01^{\ 40}$	3.98^{29}
Nb	2.1 42	1.3^{-42}	0.11^{42}	0.01^{-30}	$0.03^{\ 46}$	2.5 ⁴²	0.049^{42}	
Pb	0.87 ⁴²	0.18^{38}	0.43^{42}		$0.1^{\ 46}$	0.29 ⁴⁹	0.43 ⁴²	0.71 42
Ga	0.47 ⁴²	1.5 42	0.25^{42}	2 ²⁹	0.32 ⁴²	1.1^{-18}	0.15^{42}	2.8^{42}
Zn	2 ³⁹	0.25 ³¹	2.8^{42}	2.6^{29}	7.8^{42}	0.26^{50}		26.6^{42}
Cu	0.66^{42}	$0.37^{\ 42}$	2.2 ⁴²	0.42^{-29}	0.44^{-18}	0.32 ¹⁸	0.05^{34}	3.8 42
Ni	4.6 ³¹	0.61^{31}	12.8 25	9.6 ³¹	6^{40}	1.5 42	2.2 52	$5.2 \ 4^2$
Cr	30 ³⁵	0.25^{31}	5.5^{+0}	40^{27}	2.3 ¹⁸	0.67^{50}		5.23 62
La	0.28 35	0.18^{39}	0.01^{40}	0.015^{29}	0.67 ⁴⁵	0.49^{50}	0.4^{29}	0.29^{54}
Ce	0.47 35	0.12 39	0.01^{31}	0.016^{29}	0.96^{46}	0.241^{43}	0.1 ⁵³	0.35^{54}
Pr	0.112^{-1}	0.13^{38}				0.13^{38}		
Nd	0.7 ³¹	0.09 39	0.011 ³⁸	0.026^{29}	2.06^{45}	0.172^{43}		0.93^{39}
Sm	1.3^{31}	0.06^{39}	0.01^{31}	0.024 ²⁹	3.06^{45}	0.125^{43}	0.1 53	0.55^{54}
Eu	0.72^{31}	0.75 39	0.03^{31}	0.025 ²⁹	2.65 ⁴⁵	2.11 ⁴³	0.21 ²⁹	0.53 ⁵⁴
Tb	2.7 ³⁹	0.15^{39}	0.11^{31}	0.019^{29}	1.86^{45}	0.21^{50}	0.1 ⁵³	0.5^{54}
Dy	1.2^{31}	0.09^{31}	0.1^{31}		3.77 ⁴⁷	0.086^{43}	0.014^{38}	1.6^{56}
Er	1.33^{43}	0.041 ⁴³	0.045^{38}		3.21 45	$0.084^{\ 43}$		
Чb	0.9^{12}	0.033^{43}	0.03^{31}	0.018 ²⁹	2.69 45	0.022^{43}	0.35 53	0.26^{54}
Lu	1.4 ³¹	0.1^{39}	0.08^{31}			$0.062^{\ 43}$	0.44 ⁵³	0.6^{54}
Co	5.5 39	0.07 39	1.96^{41}	3.4 ²⁹	3.2^{-18}	0.51^{50}		72 54
Th	0.1^{39}	0.01 39	0.02^{31}	0.05^{31}	$0.01^{\ 46}$	0.088^{50}	0.12^{29}	0.06^{54}
U	0.04^{34}	0.01^{-34}	0.04^{38}	0.11^{29}	$0.09^{\ 46}$		0.14^{29}	0.12^{54}

Appendix B. 3 Partition coefficients for selected trace elements used in geochemical models (continued).

			Rhyolite		
Mineral	pyroxene	plagioclase	K-feldspar	magnetite	hbl
Ba	0.3 57	4 60	4 ⁶³	0.1 39	0.28 42
Rb	0.013 42	0.011 42	0.74 57		0.37 42
Sr	6.4 ⁵⁸	4.04 42	2.1 42	0.01 39	4.7 ⁴²
Y	0.7 42	0.55 42	0.086 42	0.95 42	5.2 ⁴²
Zr	0.33 58	0.36 56	0.21 42	0.24 42	0.76 42
Nb	0.2 42	0.26 42	0.16 42		0.98 42
Pb	0.85 42	0.35 42	0.1 42	0.71 42	0.19 42
Ga	0.32 42	0.59 42	0.7 42	2.8 42	0.77 42
Zn	2 39	0.48 39	0.23 42	2 65	1.4 ⁴²
Cu	0.2 42	0.41 42	0.6 42	3.8 42	0.77 42
Ni	6 40	1.5 ⁴²	0.9 42	5.2 4 ²	1.5 42
Cr	2.6 58	0.01 39	0.15 64	5.23 62	3 64
La	0.3 59	0.3 39	0.15 63	0.8 56	0.36 39
Ce	0.646 43	0.22 39	0.06 56	1.02 56	0.68 39
Pr		0.13 38			
Nd	1.28 43	0.19 39	0.025 43	0.98 39	1.6 ³⁹
Sm	1.81 43	0.12 39	0.04 56	0.77 65	2.3 ³⁹
Eu	2.01 43	2 ³⁹	1.13 43	0.91 39	3.2 ³⁹
Tb	3.8 ³⁹	0.14 39	0.01 56	1.3 39	2.4 ³⁹
Dy	3.8 ⁵⁸	0.18 56	0.04 56	1.6 56	
Er	1.14 43		0.162 43		0.86 64
Yb	1.14 43	0.1 39	0.024 57	1 56	1.8 ³⁹
Lu	1.28 43	0.1 39	0.003 63	0.3 39	1.8 ³⁹
Co	17 ³⁹	0.15 39	0.39 57	80 ³⁹	7 ³⁹
Th	0.1 39	0.08 56	0.03 56	0.09 56	0.16 39
U	0.12 56	0.05 56	0.04 56	0.21 56	

Appendix B. 3 Partition coefficients for selected trace elements used in geochemical models (continued).

Note: 1: Skulski et al. (1994), 2: McKenzie and O'Nions (1991), 3: Jenner et al. (1994), 4: Johnson (1994), 5: Hauri et al. (1994), 6: Hart and Dunn (1993), 7: Bougault and Hekinian (1974), 8: Mysen (1978), 9: Nagasawa et al. (1969), 10: Johnson (1998), 11: Frey (1969), 12: Nicholls and Harris (1980), 13: Duke (1976), 14: Latourette and Burnett (1982), 15: Philpotts and Schnetzler (1970), 16: Bindeman and Davis (2000), 17: Bindeman et al. (1998), 18: Ewart et al. (1973), 19: Aignertorres et al. (2007), 20: McKay and Weill (1977), 21: Beattie (1994), 22: Fujimaki et al. (1984), 23: Kloeck and Palme (1988), 24: Pedersen (1979), 25: Nabelek (1980), 26: Nielsen et al. (1992), 27: Beattie (1993), 28: Okamoto (1979), 29: Lemarchand et al. (1987), 30: Nielsen (1992), 31: Luhr and Carmichael (1980), 32: Drake and Weill (1975), 33: Hart and Brooks (1974), 34: Dostal et al. (1983), 35: Ronov and Yaroshevskiy (1976), 36: Reid (1983), 37: Gallahan and Nielsen (1992), 38: Dunn and Sen (1994), 39: Bacon and Druitt (1988), 40: Villemant (1988), 41: Gaetani and Grove (1997), 42: Ewart and Griffin (1994), 43: Schnetzler and Philpotts (1970), 44: Nielsen (1988), 45: Huang et al. (2006), 46: Marks et al. (2004), 47: Nasagawa (1973), 48: Nasagawa and Schnetzler (1971), 49: Leeman (1979), 50: Dudas et al. (1971), 51: Larsen (1979), 52: Mysen (1976), 53: Mahood and Stimac (1990), 54: Luhr et al. (1984), 55: Latourrette et al. (1991), 56: Nash and Crecraft (1985), 57: Mahood and Hildreth (1983), 58: Sisson (1991), 59: Michael (1988), 60: Long (1978), 61: Esperança et al. (1997), 62: Streck and Grunder (1997), 63: Stix and Gorton (1990), 64: Bea et al. (1994), 64: Sisson (1994), 65: Mysen and Virgo (1980).

Block	Unit	Sample	Major minerals	SiO_2 (wt.%)	Rock-type*	Reference
	Mount Thom	6980	Kfs, plg, qz, bt, hbl		Granodiorite	Pe-Piper et al. (1998)
	Gunshot Brook	7403	Kfs, plg, qz, ep, bt	67.82	Granodiorite	Pe-Piper et al. (1998)
	Gunshot Brook	7405	Kfs, plg, qz, bt, hbl	68.65	Granodiorite	Pe-Piper et al. (1998)
¥	Gunshot Brook	7408	Kfs, plg, qz, bt, hbl	68.7	Granodiorite	Pe-Piper et al. (1998)
Bloc	Gunshot Brook	7435	Kfs, plg, qz, bt	68.72	Granodiorite	Pe-Piper et al. (1998)
effers	Gunshot Brook	7440	Plg, qz, bt, ep	67.57	Tonalite	Pe-Piper et al. (1998)
	River John	7412	Kfs, plg, qz, ep, bt	63.91	Granodiorite	Pe-Piper et al. (1998)
	River John	7436	Plg, qz, bt, ep	64.69	Tonalite	Pe-Piper et al. (1998)
	River John	7437	Plg, qz, bt, ep, hbl	65.36	Tonalite	Pe-Piper et al. (1998)
	River John	7442	Plg, Kfs, qz, bt, ep	63.6	Granodiorite	Pe-Piper et al. (1998)
						Pe-Piper et al.
ck ck	Frog Lake	3134	Plg, qz, bt,	64.1	Tonalite	(2010) Pe-Piper et al.
r Blo	Debert River	7347	Kfs, plg, qz Kfs, plg, qz	69.88	Granodiorite	(1998) Pe-Piper et al
Rive	Debert River	6621	hbl, bt	61.54	Granodiorite	(1998) De Diner et al
Bass	Debert River	6633	plg, bt, qz, ep	61.44	Tonalite	(1998)
	Debert River	6636	qz, bt, plg	69.01	Tonalite	Pe-Piper et al. (1998)

Appendix B. 4 Mineralogical description of Neoproterozoic intermediate rocks from the Cobequid Highlands.

* Names for rock types were assigned in the present work, based on the relative abudances of feldspars.

		6636 Ton		69.01	0.82	14.04	5.13	0.06	1.5	1.18	1.97	4.19	0.1	1.5	99.5		718	158	162	42	275	16	13	118	17	411	24	32	105	85	n.d.	n.d.	n.d.	n.d.
	lock	6633 Ton		61.44	1.02	18.24	7.8	0.1	2.72	1.03	2.11	3.68	0.08	2.6	100.8		904	138	143	46	220	18	14	19	24	131	22	39	172	111	n.d.	n.d.	n.d.	n.d.
	River B	3134 Ton		64.09	0.97	15.6	5.38	0.09	2.77	5.95	3.04	0.58	0.25	0.8	99.52		183	15	375	21	383	11	6	13	18	63	24	10	120	10	n.d.	45.6	LL	n.d.
	Bass	7347 Gd		69.88	0.36	16.27	2.83	0.1	0.95	1.87	4.21	3.42	0.14	0.89	100.9		1052	103	300	24	137	19	14	16	21	50	12	19	49	4	5	36	52	n.d.
		6621 Gd		61.54	0.72	18.8	4.31	0.14	1.32	3.66	5.02	3.39	0.25	0.6	99.75		2712	162	756	36	437	20	13	27	19	84	10	8	47	9	n.d.	n.d.	n.d.	n.d.
		6061 Bas		46.1	2.98	14.9	15	0.21	5.24	9.8	2.65	0.5	0.43	2.3	100		210	13	289	45	214	17	2.67	0	23	136	30	69	365	93	46.15	18.73	40.66	0.00
		4601 Bas		47.96	1.81	17.04	11.85	0.26	5.81	5.42	4.75	0.41	0.23	3.8	99.34		277	14	316	19	122	0	0.00	0	17	120	45	127	273	232	0.00	0.00	0.00	0.00
		1950 Gb		48.26	1.18	19.20	9.16	0.20	6.19	8.25	2.57	2.17	0.19	2.23	9.66		491	85	363	20	78	6	2.70	40	18	243	38	23	266	32	41.00	11.50	27.00	0.00
		1347 Gb		43.32	2.35	15.59	12.03	0.20	8.63	9.78	2.49	1.26	0.43	2.85	98.9		440	32	656	22	173	33	2.10	5	22	115	69	228	236	321	59.00	27.20	47.00	0.00
		7437 Ton		65.36	0.41	17.27	4.01	0.12	1.44	5	4.18	1.52	0.17	0.72	100.2		516	47	590	11	133	8	n.d.	18	20	75	17	n.d.	62	14	n.d.	n.d.	n.d.	n.d.
		7436 Ton		64.69	0.43	17.54	4.16	0.11	1.44	5.16	4.41	1.65	0.17	0.56	100.3		576	48	588	13	130	7	n.d.	14	20	65	16	9	69	10	n.d.	n.d.	n.d.	n.d.
		7440 Ton		67.57	0.32	17.59	2.27	0.07	0.8	3.62	4.9	1.93	0.11	1.03	100.2		972	46	865	16	137	10	n.d.	18	21	69	12	9	28	6	n.d.	n.d.	n.d.	n.d.
	N	7442 Gd		63.6	0.43	19.15	2.93	0.08	1.07	4.53	5.27	1.63	0.15	1.22	100.1		1023	40	1041	8	168	9	n.d.	17	23	95	13	n.d.	46	10	n.d.	n.d.	n.d.	n.d.
	ers Bloc	7435 Gd		68.72	0.26	16.72	2.39	0.08	0.75	3.36	4.26	2.56	0.12	0.75	79.97		1598	55	969	5	103	5	n.d.	18	19	59	12	n.d.	35	n.d.	n.d.	n.d.	n.d.	n.d.
	Jeff	7412 Gd		63.91	0.44	18.02	3.33	0.08	1.32	3.97	4.77	1.54	0.17	1.8	99.35		611	37	821	8	114	8	n.d.	16	20	81	n.d.	n.d.	61	6	n.d.	n.d.	n.d.	n.d.
I		7408 Gd		68.7	0.25	16.98	1.88	0.06	0.63	3.12	5.05	1.96	0.09	0.8	99.52		1189	45	864	7	112	7	n.d.	15	18	48	n.d.	n.d.	28	10	n.d.	n.d.	n.d.	n.d.
		7405 Gd		68.65	0.24	16.92	1.83	0.04	0.57	2.53	4.68	2.42	0.09	1.3	99.27		1398	48	812	9	66	9	n.d.	21	16	47	n.d.	5	24	13	n.d.	n.d.	n.d.	n.d.
		7403 Gd	wt.%)	57.82	0.26	17.46	1.81	0.07	0.65	2.12	5.15	2.63	0.09	1.2	99.26	s (ppm)	1403	55	755	7	104	7	n.d.	25	18	59	n.d.	9	27	5	n.d.	n.d.	n.d.	n.d.
		e /pe	oxides (. –										<u>elements</u>																		
		Sample Rockty	Major	SiO_2	TiO_2	Al_2O_3	$\mathrm{Fe_2O_3}$	MnO	MgO	CaO	Na_2O	${ m K}_2{ m O}$	P_2O_5	LOI	Total	Trace	Ba	Rb	Sr	Y	Zr	ЧN	Th	Pb	Ga	Zn	Cu	Ni	>	\mathbf{Cr}	Co	La	Ce	\mathbf{Pr}

Appendix B. 5 Whole-rock compositions of intermediate and mafic rocks used for geochemical modelling.

			-													
	6636	Ton	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	1. (2002)
lock	6633	Ton	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	iper et a
River B	3134	Ton	20	5.5	0.62	n.d.	0.9	n.d.	n.d.	n.d.	n.d.	2.37	0.41	n.d.	n.d.	ind Pe-P
Bass	7347	Gd	22	n.d.	12	4	(1987) 8									
	6621	Gd	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	I Turner
	6061	Bas	24.91	6.76	2.56	0.00	1.23	0.00	0.00	0.00	0.00	3.88	0.49	n.d.	0.01	Piper and
	4601	Bas	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	0.00	n.d.	0.00	rom Pe-I
	1950	Gb	15.00	2.93	0.97	0.00	0.50	0.00	0.00	0.00	0.00	1.56	0.27	n.d.	0.60	l. Data fi
	1347	Gb	21.00	5.31	1.85	0.00	0.50	0.00	0.00	0.00	0.00	1.38	0.24	n.d.	0.70	terminec
	7437	Ton	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	= not de
	7436	Ton	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	asalt n.d.
	7440	Ton	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	Bas = bs
<u>ck</u>	7442	Gd	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	gabbro,
fers Bloc	7435	Gd	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	ite, Gb=
Jef	7412	Gd	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n= tonal
	7408	Gd	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	orite, To
	7405	Gd	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	granodi
	7403	Gd	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	n.d.	ns: Gd=
	Sample	Rocktype	Nd	Sm	Eu	Gd	Tb	Dy	Но	Er	Tm	Yb	Lu	Sc	U	Abbreviatic

APPENDIX C

Electron Microprobe analyses and recalculated mineral formulae of important hydrothermal major and REE-bearing minerals hosted in the fractures of the late Paleozoic A-type granites of the Cobequid Highlands.

Sample	6497	4106	4106	4106		6497	6497	6497
Mineral	ер	ер	ер	ер		And	And	And
SiO ₂	36.01	38.86	39.01	39.01		34.39	35.48	35.31
TiO ₂	b.d.	0.10	b.d.	0.01		1.20	1.04	1.12
Al ₂ O ₃	21.40	24.61	24.86	24.55		3.26	4.33	3.92
FeO _T	12.88	11.55	11.17	11.50		25.11	23.04	23.79
MnO	0.11	0.15	0.11	0.22		0.69	0.70	0.61
MgO	b.d.	b.d.	b.d.	0.01		0.06	0.07	0.07
CaO	22.07	23.77	23.63	23.23		32.90	33.15	33.20
Total	92.48	99.04	98.78	98.52		97.61	97.81	98.02
Si	3.008	3.012	3.027	3.041	Si	2.929	2.991	2.978
Al ^{IV}	0.000	0.000	0.000	0.000	Al ^{IV}	0.071	0.009	0.022
T-site sum	3.008	3.012	3.027	3.041	Sum_Z	3.000	3.000	3.000
<u>A1</u>					Al ^{VI}	0.256	0.421	0.367
Ca^{2+}	1.000	1.000	1.000	1.000	Fe ³⁺	1.608	1.460	1.508
Al ^{VIII}	0.000	0.000	0.000	0.000	Ti	0.077	0.066	0.071
Fe ³⁺	0.000	0.000	0.000	0.000	Sum_Y	1.940	1.947	1.948
Mn ²⁺	0.015	0.026	0.013	0.008	Mg	0.008	0.009	0.009
Mg	0.000	0.000	0.000	0.000	Mn	0.050	0.050	0.044
sum	1.015	1.026	1.013	1.008	Ca	3.002	2.994	3.000
<u>A2</u>					Sum_X	3.060	3.053	3.052
Ca^{2+}	0.975	0.974	0.965	0.939	Andradite	82.85	74.98	77.43
Al ^{VIII}	0.000	0.000	0.000	0.000	Grossular	15.28	23.06	20.79
Fe ²⁺	0.000	0.000	0.000	0.000	Pyrope	0.25	0.28	0.29
Fe ³⁺	0.000	0.000	0.000	0.000	Spessartine	1.62	1.63	1.43
Mn ²⁺	0.000	0.000	0.000	0.000	Uvarovite	0.00	0.03	0.07
Mg	0.000	0.000	0.000	0.000				
sum	0.975	0.974	0.965	0.939				
<u>M1</u>								
Al VIII	1.000	1.000	1.000	1.000				
Ti	0.000	0.000	0.000	0.000				
Fe ³⁺	0.000	0.000	0.000	0.000				
Mn ³⁺	0.000	0.000	0.000	0.000				
Mg	0.000	0.000	0.000	0.000				
Fe ²⁺	0.000	0.000	0.000	0.000				
Mn^{2+}	0.007	0.005	0.000	0.000				
sum	1.007	1.005	1.000	1.000				
$\underline{M2}$	1 000	1 000	1 000	1 000				
AI^{VIII}	1.000	1.000	1.000	1.000				
Fe	0.000	0.000	0.000	0.000				
sum	1.000	1.000	1.000	1.000				
<u>M3</u>	0.107	0.240	0 274	0.005				
Al ·····	0.107	0.248	0.274	0.225				
11 E 2^{3+}	0.000	0.005	0.000	0.000				
re Mn ³⁺	0.900	0.748	0.723	0.749				
Ma	0.000	0.000	0.000	0.000				
Fe^{2+}	0.000	0.000	0.000	0.000				
Mn ²⁺	0.000	0.000	0.000	0.000				
sum	1.007	1.001	1.000	1.000				

Appendix C. 1 Representative microprobe analyses of epidote and garnet.

Notes: Chemical formula of epidote calculated on the basis of 8 cations (Armbruster et al. 2006), stoichiometric formula of garnet calculated on the basis of 12 oxygens, b.d.= below detection limit.

Pluton PHP PHP PHP PHP CCP CCP Sample 4106 4106 4106 4106 6284 6284 SiO2 37.98 36.88 36.48 38.52 34.17 36.59 TiO2 2.05 2.14 1.75 1.95 1.18 1.6 Al2O3 19.28 17.67 17.06 18.38 17.37 17.96 FeOT 19.66 22.14 22.33 17.81 24.16 22.81 MnO 0.06 0.05 0.04 0.20 0.11 MgO 7.36 6.74 7.58 8.76 7.47 7.22 CaO 0.03 0.02 0.05 b.d. b.d. b.d. Na2O 0.10 0.05 0.09 0.07 0.04 0.03 K2O 9.83 9.72 9.49 10.06 8.67 8.84 H2O* 1.92 1.87 1.85 1.92 1.80 <							
$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	Pluton	PHP	PHP	PHP	PHP	ССР	ССР
SiO2 37.98 36.88 36.48 38.52 34.17 36.59 TiO2 2.05 2.14 1.75 1.95 1.18 1.6 Al2O3 19.28 17.67 17.06 18.38 17.37 17.96 FeOT 19.66 22.14 22.33 17.81 24.16 22.81 MnO 0.06 0.06 0.05 0.04 0.20 0.11 MgO 7.36 6.74 7.58 8.76 7.47 7.22 CaO 0.03 0.02 0.05 b.d. b.d. b.d. Na2O 0.10 0.05 0.09 0.07 0.04 0.03 K2O 9.83 9.72 9.49 10.06 8.67 8.84 H ₂ O* 1.92 1.87 1.85 1.92 1.80 1.86 Total 98.27 97.29 96.73 97.51 95.06 97.02 Structural formula 5 1.155 1.173 1.117 1.275 1.184 Ti 0.115 0.124 0.102 </th <th>Sample</th> <th>4106</th> <th>4106</th> <th>4106</th> <th>4106</th> <th>6284</th> <th>6284</th>	Sample	4106	4106	4106	4106	6284	6284
$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	SiO ₂	37.98	36.88	36.48	38.52	34.17	36.59
Al2O3 19.28 17.67 17.06 18.38 17.37 17.96 FeOT 19.66 22.14 22.33 17.81 24.16 22.81 MnO 0.06 0.06 0.05 0.04 0.20 0.11 MgO 7.36 6.74 7.58 8.76 7.47 7.22 CaO 0.03 0.02 0.05 b.d. b.d. b.d. Na2O 0.10 0.05 0.09 0.07 0.04 0.03 K2O 9.83 9.72 9.49 10.06 8.67 8.84 H ₂ O* 1.92 1.87 1.85 1.92 1.80 1.86 Total 98.27 97.29 96.73 97.51 95.06 97.02 Structural formula Si 2.842 2.835 2.827 2.883 2.725 2.816 Al 1.158 1.165 1.173 1.117 1.275 1.184 Ti 0.115 0.124 0.102 0.110 0.071 0.093 Al 0.541 0.4	TiO ₂	2.05	2.14	1.75	1.95	1.18	1.6
$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	Al ₂ O ₃	19.28	17.67	17.06	18.38	17.37	17.96
$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	FeO _T	19.66	22.14	22.33	17.81	24.16	22.81
$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	MnO	0.06	0.06	0.05	0.04	0.20	0.11
$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	MgO	7.36	6.74	7.58	8.76	7.47	7.22
$\begin{array}{c ccccccccccccccccccccccccccccccccccc$	CaO	0.03	0.02	0.05	b.d.	b.d.	b.d.
K2O9.839.729.4910.068.678.84H2O*1.921.871.851.921.801.86Total98.2797.2996.7397.5195.0697.02Structural formulaSi2.8422.8352.8272.8832.7252.816Al1.1581.1651.1731.1171.2751.184Ti0.1150.1240.1020.1100.0710.093Al0.5410.4340.3840.5030.3570.444Fe ⁺² 1.2301.4231.4471.1151.6121.468Mn0.0040.0040.0030.0030.0140.007Mg0.8210.7720.8760.9780.8880.828	Na ₂ O	0.10	0.05	0.09	0.07	0.04	0.03
H_2O^* 1.921.871.851.921.801.86Total98.2797.2996.7397.5195.0697.02Structural formulaSi2.8422.8352.8272.8832.7252.816Al1.1581.1651.1731.1171.2751.184Ti0.1150.1240.1020.1100.0710.093Al0.5410.4340.3840.5030.3570.444Fe ⁺² 1.2301.4231.4471.1151.6121.468Mn0.0040.0040.0030.0030.0140.007Mg0.8210.7720.8760.9780.8880.828Co0.0020.0020.0020.0000.0000.000	K ₂ O	9.83	9.72	9.49	10.06	8.67	8.84
Total 98.27 97.29 96.73 97.51 95.06 97.02 Structural formula Si 2.842 2.835 2.827 2.883 2.725 2.816 Al 1.158 1.165 1.173 1.117 1.275 1.184 Ti 0.115 0.124 0.102 0.110 0.071 0.093 Al 0.541 0.434 0.384 0.503 0.357 0.444 Fe ⁺² 1.230 1.423 1.447 1.115 1.612 1.468 Mn 0.004 0.004 0.003 0.003 0.014 0.007 Mg 0.821 0.772 0.876 0.978 0.888 0.828	H ₂ O*	1.92	1.87	1.85	1.92	1.80	1.86
Structural formula Si 2.842 2.835 2.827 2.883 2.725 2.816 Al 1.158 1.165 1.173 1.117 1.275 1.184 Ti 0.115 0.124 0.102 0.110 0.071 0.093 Al 0.541 0.434 0.384 0.503 0.357 0.444 Fe ⁺² 1.230 1.423 1.447 1.115 1.612 1.468 Mn 0.004 0.003 0.003 0.014 0.007 Mg 0.821 0.772 0.876 0.978 0.888 0.828 Co 0.002 0.004 0.004 0.000 0.000 0.000	Total	98.27	97.29	96.73	97.51	95.06	97.02
Si2.8422.8352.8272.8832.7252.816Al1.1581.1651.1731.1171.2751.184Ti0.1150.1240.1020.1100.0710.093Al0.5410.4340.3840.5030.3570.444Fe ⁺² 1.2301.4231.4471.1151.6121.468Mn0.0040.0040.0030.0030.0140.007Mg0.8210.7720.8760.9780.8880.828Ca0.0020.0020.0040.0000.0000.000	Structural	formula					
Al1.1581.1651.1731.1171.2751.184Ti0.1150.1240.1020.1100.0710.093Al0.5410.4340.3840.5030.3570.444 Fe^{+2} 1.2301.4231.4471.1151.6121.468Mn0.0040.0040.0030.0030.0140.007Mg0.8210.7720.8760.9780.8880.828Ca0.0020.0020.0040.0040.0000.000	Si	2.842	2.835	2.827	2.883	2.725	2.816
Ti 0.115 0.124 0.102 0.110 0.071 0.093 Al 0.541 0.434 0.384 0.503 0.357 0.444 Fe ⁺² 1.230 1.423 1.447 1.115 1.612 1.468 Mn 0.004 0.004 0.003 0.003 0.014 0.007 Mg 0.821 0.772 0.876 0.978 0.888 0.828 Ca 0.002 0.002 0.004 0.000 0.000 0.000	Al	1.158	1.165	1.173	1.117	1.275	1.184
Al 0.541 0.434 0.384 0.503 0.357 0.444 Fe^{+2} 1.230 1.423 1.447 1.115 1.612 1.468 Mn 0.004 0.004 0.003 0.003 0.014 0.007 Mg 0.821 0.772 0.876 0.978 0.888 0.828 Ca 0.002 0.002 0.004 0.000 0.000 0.000	Ti	0.115	0.124	0.102	0.110	0.071	0.093
Fe ⁺² 1.230 1.423 1.447 1.115 1.612 1.468 Mn 0.004 0.004 0.003 0.003 0.014 0.007 Mg 0.821 0.772 0.876 0.978 0.888 0.828 Ca 0.002 0.002 0.004 0.000 0.000 0.000	Al	0.541	0.434	0.384	0.503	0.357	0.444
Mn 0.004 0.004 0.003 0.003 0.014 0.007 Mg 0.821 0.772 0.876 0.978 0.888 0.828 Ca 0.002 0.002 0.004 0.000 0.000 0.000	Fe ⁺²	1.230	1.423	1.447	1.115	1.612	1.468
Mg 0.821 0.772 0.876 0.978 0.888 0.828 Ga 0.002 0.002 0.004 0.000 0.000 0.000	Mn	0.004	0.004	0.003	0.003	0.014	0.007
	Mg	0.821	0.772	0.876	0.978	0.888	0.828
Ca 0.002 0.002 0.004 0.000 0.000 0.000	Ca	0.002	0.002	0.004	0.000	0.000	0.000
Na 0.015 0.007 0.014 0.010 0.006 0.004	Na	0.015	0.007	0.014	0.010	0.006	0.004
K 0.938 0.953 0.938 0.961 0.882 0.868	K	0.938	0.953	0.938	0.961	0.882	0.868

Appendix C. 2 Representative microprobe analyses of hydrothermal biotite.

Notes: Biotite formula calculated on the basis of 10 oxygens, and 2 OH atoms in the A-site as per Rieder et al. (1998).

Pluton	NRP	NRP	NRP	PHP										
Sample	1779	1779	1779	4672	4672	4672	4672	4672	4672	4672	4672	4672	4672	4672
Analysis	81	83	68	53	55	56	57	58	59	61	63	64	65	6 6
SiO ₂	26.63	26.78	30.30	29.83	23.86	23.66	27.68	26.06	28.44	23.68	23.44	24.99	28.47	25.75
TiO ₂	b.d.	0.02	b.d.	0.02	0.06	b.d.	0.01	b.d.	b.d.	0.14	0.01	b.d.	b.d.	b.d.
Al ₂ O ₃	16.65	17.32	14.93	19.28	20.16	20.94	19.50	19.39	19.21	20.06	20.81	20.86	19.29	19.67
FeOr	31.72	29.64	25.76	30.05	37.65	36.46	31.86	31.69	31.48	37.57	38.88	35.78	30.47	34.45
MnO	0.54	0.42	0.29	0.17	0.16	0.16	0.15	0.12	0.15	0.20	0.18	0.16	0.14	0.18
MgO	10.43	10.37	11.17	3.45	5.00	5.57	4.13	4.76	3.35	4.30	4.20	3.98	3.25	3.75
CaO	0.05	0.16	0.43	0.27	b.d.	0.03	0.20	0.14	0.28	b.d.	b.d.	0.11	0.22	0.12
Na ₂ O	b.d.	0.02	0.08	0.45	b.d.	0.02	0.14	0.20	0.26	0.04	0.02	0.05	0.14	0.06
K_2O	0.02	0.02	0.26	0.59	0.03	0.04	0.41	0.94	0.47	0.05	0.02	0.23	0.53	0.22
Total	86.03	84.75	83.22	84.10	86.92	86.87	84.07	83.30	83.63	86.03	87.57	86.15	82.51	84.19
Structural	formula													
Si	5.951	5.997	6.726	6.663	5.451	5.370	6.279	6.027	6.464	5.474	5.348	5.680	6.520	5.951
AIIV	2.049	2.003	1.274	1.337	2.549	2.630	1.721	1.973	1.536	2.526	2.652	2.320	1.480	2.049
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
	2.333	2.565	2.629	3.736	2.875	2.969	3.488	3.309	3.606	2.935	2.941	3.265	3.722	3.304
Ti	0.000	0.003	0.000	0.003	0.010	0.000	0.002	0.000	0.000	0.024	0.002	0.000	0.000	0.000
Fe^{3+}	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000
Fe^{2+}	5.928	5.552	4.782	5.614	7.194	6.921	6.045	6.129	5.985	7.265	7.420	6.802	5.835	6.658
Cr	0.000	0.000	0.000	0.007	0.000	0.000	0.006	0.002	0.005	0.000	0.006	0.002	0.011	0.001
\mathbf{Mn}	0.102	0.080	0.054	0.032	0.031	0.030	0.028	0.024	0.028	0.039	0.034	0.031	0.026	0.035
Mg	3.475	3.463	3.695	1.148	1.703	1.886	1.395	1.640	1.136	1.484	1.430	1.347	1.111	1.292
Ca	0.011	0.039	0.102	0.065	0.000	0.006	0.049	0.033	0.068	0.000	0.000	0.026	0.054	0.029
Na	0.000	0.007	0.034	0.193	0.002	0.008	0.060	0.090	0.114	0.018	0.009	0.024	0.061	0.027
K	0.007	0.006	0.074	0.169	0.009	0.011	0.119	0.278	0.136	0.014	0.007	0.067	0.155	0.063
Cations	19.856	19.715	19.370	18.967	19.824	19.831	19.192	19.505	19.078	19.779	19.849	19.564	18.975	19.409

Appendix C. 3 Representative microprobe analyses of chlorite.

Pluton	PHP	PHP	PHP	PHP	PHP	NRP	NRP	NRP	NRP	CCP						
Sample	4672	4672	4672	4672	4672	4824	4824	4824	4824	6289	6289	6289	6289	6289	6289	6289
Analysis	67	68	69	70	71	74	75	<i>LT</i>	80	102	107	111	115	90	93	96
SiO ₂	28.06	27.62	27.29	27.76	25.64	25.65	26.17	24.86	25.63	26.17	25.48	34.17	36.59	24.73	24.64	26.57
TiO ₂	0.01	0.02	b.d.	b.d.	0.03	b.d.	0.02	0.01	0.09	0.08	0.16	1.18	1.60	0.17	b.d.	0.11
Al_2O_3	19.84	19.62	20.62	19.35	20.41	19.12	18.78	18.81	17.54	19.18	20.42	17.37	17.96	21.13	21.43	18.44
FeO_{T}	30.90	32.25	33.07	31.09	35.50	33.19	31.91	33.77	30.41	30.32	29.35	24.16	22.81	30.65	31.20	30.21
MnO	0.13	0.16	0.15	0.13	0.09	0.55	0.39	0.42	0.46	0.21	0.25	0.20	0.11	0.28	0.30	0.32
MgO	4.18	4.04	3.81	4.88	5.00	9.14	9.59	8.33	8.60	10.70	11.18	7.47	7.22	10.09	10.20	10.26
CaO	0.20	0.10	0.09	0.10	0.05	0.06	0.09	0.05	0.77	0.09	0.05	b.d.	b.d.	0.11	0.01	0.06
Na ₂ O	0.46	0.22	0.17	0.18	0.07	0.02	0.08	0.05	0.23	b.d.	b.d.	0.04	0.03	b.d.	b.d.	0.01
$\mathbf{K}_{2}\mathbf{O}$	0.35	0.29	0.55	0.88	0.18	0.07	0.10	0.04	0.12	0.15	b.d.	8.67	8.84	0.02	0.02	0.30
Total	84.11	84.31	85.75	84.37	86.97	87.79	87.12	86.34	83.85	86.88	86.90	93.27	95.15	87.17	87.78	86.28
Structur:	ul formul	a														
Si	6.319	6.255	6.106	6.269	5.750	5.651	5.765	5.606	5.873	5.722	5.539	6.937	7.168	5.404	5.360	5.857
Al ^{IV}	1.681	1.745	1.894	1.731	2.250	2.349	2.235	2.394	2.127	2.278	2.461	1.063	0.832	2.596	2.640	2.143
T-site	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000	8.000
Al ^{VI}	3.581	3.487	3.539	3.415	3.141	2.612	2.637	2.603	2.607	2.661	2.768	3.089	3.312	2.843	2.848	2.644
Ti	0.002	0.003	0.000	0.000	0.005	0.000	0.003	0.001	0.015	0.012	0.026	0.181	0.235	0.027	0.000	0.018
Fe^{3+}	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000
Fe ²⁺	5.820	6.110	6.187	5.871	6.657	6.116	5.878	6.370	5.828	5.545	5.337	4.103	3.737	5.602	5.675	5.570
Cr	0.010	0.006	0.005	0.005	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000
Mn	0.024	0.031	0.029	0.024	0.017	0.103	0.072	0.080	0.089	0.039	0.046	0.034	0.018	0.053	0.055	0.061
Mg	1.402	1.365	1.269	1.644	1.670	3.002	3.147	2.801	2.939	3.489	3.624	2.262	2.108	3.287	3.306	3.372
Ca	0.048	0.023	0.022	0.024	0.011	0.014	0.022	0.013	0.190	0.020	0.011	0.000	0.000	0.025	0.003	0.015
Na	0.199	0.097	0.074	0.077	0.030	0.007	0.033	0.023	0.104	0.000	0.000	0.017	0.010	0.000	0.000	0.003
K	0.099	0.082	0.158	0.252	0.052	0.021	0.029	0.012	0.035	0.041	0.000	2.247	2.208	0.006	0.004	0.084
Cations	19.185	19.204	19.283	19.312	19.583	19.875	19.821	19.903	19.807	19.807	19.812	19.933	19.628	19.843	19.891	19.767
Notes: St	ructural f	òrmula c	alculate c	on the base	sis of 10	O and 8 (OH atom	s.								

Sample	6497	4106	1779	4829	4829	4829	4829	4829	4829	1779	1779
Mineral	bastnäsite	bastnäsite	bastnäsite	parisite	parisite	parisite	parisite	parisite	parisite	synchysite	synchysite
P_2O_5	b.d.	0.37	0.02	0.01	b.d.	0.07	0.01	0.10	b.d.	b.d.	0.07
Nb2O5	b.d.	b.d.	0.02	b.d.	0.05	0.03	0.08	b.d.	b.d.	b.d.	0.03
$A_{S2}O_5$	b.d.	0.06	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.
SiO ₂	6.99	3.13	2.06	1.17	1.23	0.71	1.53	0.76	1.96	1.02	0.61
ZrO ₂	b.d.	0.07	0.02	b.d.	b.d.	0.01	0.05	0.01	0.02	0.03	b.d.
ThO ₂	b.d.	3.11	0.26	0.23	0.33	1.77	0.26	0.33	0.36	0.12	0.16
\mathbf{UO}_2	0.08	0.23	b.d.	0.02	0.04	0.07	b.d.	0.16	0.10	0.19	b.d.
Al ₂ O ₃	0.67	0.65	0.51	0.40	0.47	0.34	0.64	0.11	1.26	0.39	0.23
Y_2O_3	b.d.	1.31	1.99	2.25	2.06	2.34	2.08	2.48	2.27	7.44	4.92
La_2O_3	12.90	15.40	12.76	8.53	10.55	11.32	8.81	11.47	9.67	8.19	8.63
Pr_2O_3	2.21	2.17	2.33	1.94	2.26	2.27	2.07	2.49	2.15	1.67	2.01
Ce2O3	24.20	25.11	26.53	15.65	18.15	17.08	16.23	17.14	17.29	18.07	19.99
Nd ₂ O ₃	8.59	9.77	9.10	8.85	10.05	10.40	8.43	11.35	9.13	8.05	9.15
Sm2O3	0.53	1.15	1.33	1.74	1.88	1.84	1.57	2.18	1.65	1.78	1.88
Eu2O3	0.59	0.76	0.72	0.72	0.82	0.78	0.65	0.86	0.62	0.53	0.58
Gd_2O_3	0.17	0.68	0.78	1.38	1.27	1.23	1.21	1.68	1.18	2.12	1.79
Dy ₂ O ₃	0.17	0.36	0.42	0.67	0.58	0.59	0.35	0.59	0.59	1.12	0.84
$H_{02}O_3$	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.33	0.12
Er_2O_3	0.07	0.05	0.20	b.d.	0.02	0.12	0.20	0.11	0.10	0.19	0.20
Yb ₂ O ₃	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	0.13	b.d.
FeOT	11.16	2.24	5.58	11.34	9.10	8.18	14.75	2.74	10.69	0.48	0.91
MgO	0.57	0.01	0.13	b.d.	b.d.	0.03	0.11	b.d.	0.01	b.d.	0.25
MnO	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.	b.d.
CaO	6.95	4.41	7.86	15.02	12.20	11.21	12.41	13.73	10.79	17.95	17.85
$\mathbf{K_{2}O}$	0.16	0.22	0.08	0.15	0.11	0.06	0.08	0.06	0.07	0.32	0.07
Ĩ	0.14	3.81	3.99	2.75	3.14	2.89	2.78	2.77	2.83	2.55	2.68
CO ₂ calc	27.92	23.08	22.85	27.61	27.26	26.23	28.44	25.53	27.96	30.22	29.67
0- F	-0.06	-1.60	-1.68	-1.16	-1.32	-1.22	-1.17	-1.16	-1.19	-1.07	-1.13
Total	104.00	96.58	97.86	99.22	100.23	98.34	101.55	95.49	99.50	101.81	101.52

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Sample	6497	4106	1779	4829	4829	4829	4829	4829	4829	1779	1779
Mineral	bastnäsite	bastnäsite	bastnäsite	parisite	parisite	parisite	parisite	parisite	parisite	synchysite	synchysite
C =	1	1	1	ŝ	ŝ	ŝ	ŝ	ŝ	ŝ	6	7
Structur	al formula										
Р	0.000	0.009	0.000	0.000	0.000	0.004	0.000	0.007	0.000	0.000	0.003
qN	0.000	0.000	0.000	0.000	0.002	0.001	0.002	0.000	0.000	0.000	0.001
\mathbf{As}	0.000	0.001	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000
Si	0.160	0.095	0.056	0.081	0.092	0.055	0.105	0.062	0.144	0.050	0.030
Zr	0.000	0.001	0.000	0.000	0.000	0.000	0.002	0.000	0.001	0.001	0.000
ЧТ	0.000	0.022	0.002	0.004	0.006	0.032	0.004	0.006	0.006	0.001	0.002
Ŋ	0.000	0.002	0.000	0.000	0.001	0.001	0.000	0.003	0.002	0.002	0.000
N	0.018	0.023	0.016	0.033	0.041	0.032	0.052	0.011	0.110	0.022	0.013
Υ	0.000	0.021	0.029	0.084	0.082	0.098	0.076	0.108	0.089	0.194	0.130
La	0.109	0.173	0.128	0.220	0.290	0.327	0.223	0.345	0.263	0.148	0.158
Pr	0.018	0.024	0.023	0.049	0.061	0.065	0.052	0.074	0.058	0.030	0.036
Ce	0.203	0.280	0.263	0.401	0.495	0.489	0.408	0.512	0.466	0.325	0.364
PN	0.070	0.106	0.088	0.221	0.267	0.291	0.207	0.331	0.240	0.141	0.162
Sm	0.004	0.012	0.012	0.042	0.048	0.050	0.037	0.061	0.042	0.030	0.032
Eu	0.005	0.008	0.007	0.017	0.021	0.021	0.015	0.024	0.016	0.009	0.010
Gd	0.001	0.007	0.007	0.032	0.031	0.032	0.028	0.046	0.029	0.034	0.029
Dy	0.001	0.004	0.004	0.015	0.014	0.015	0.008	0.016	0.014	0.018	0.013
Er	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.005	0.002
Ho	0.000	0.000	0.002	0.000	0.001	0.003	0.004	0.003	0.002	0.003	0.003
ЧÞ	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.002	0.000
Fe	0.214	0.057	0.126	0.663	0.566	0.535	0.847	0.187	0.659	0.020	0.038
Mg	0.020	0.000	0.005	0.000	0.000	0.003	0.011	0.000	0.002	0.000	0.018
Mn	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000	0.000
Ca	0.171	0.144	0.228	1.125	0.973	0.940	0.913	1.200	0.852	0.943	0.950
K	0.005	0.008	0.003	0.013	0.011	0.006	0.007	0.006	0.007	0.020	0.004
Notes:	CO ₂ calculate	ed through ch	arge-balance,	the oxygen	equivalent f	or fluorine i	s calculated	on the basis	of two F at	oms for every	O atom.

Sampla	6407	1871	1770	6280	6280	6280	4106	4106
Mineral	Thorite	Thorite	Thorite	Thorite	Chernovite	Chernovite	Cerianite	Cerianite
	0.08	0.65	0.47	3 04	1.62	2 64	2 04	1 28
As2O5	0.01	1.24	0.01	1.88	35.03	33.73	3.36	2.21
SiO ₂	22.26	13.14	17.01	15 72	2 73	3.02	13.24	16.68
	0.56	h d	h d	0.03	b.d	b.d	0.03	h d
	1.25	2.40	0.10	0.94	0.08	0.07	b.d.	b.d.
ThO ₂	42.14	26.48	61.07	32.81	4.48	2.00	0.69	2.39
	1.46	1.77	8.43	1.68	0.48	0.30	0.03	0.69
Al ₂ O ₃	4.82	0.39	0.13	1.17	b.d.	b.d.	2.00	1.68
Y2O3	0.93	9.18	0.63	9.75	27.42	30.91	8.74	8.29
	1.29	2.52	b.d.	0.44	b.d.	0.06	0.10	0.04
Ce ₂ O ₃	3.57	6.34	0.20	14.07	7.00	4.55	47.90	43.80
Pr ₂ O ₃	0.41	0.72	b.d.	0.31	0.32	0.18	0.13	0.08
Nd ₂ O ₃	1.71	2.76	0.06	1.22	1.59	1.26	0.68	0.70
Sm ₂ O ₃	0.52	0.74	b.d.	0.69	1.78	1.57	0.30	0.30
Eu ₂ O ₃	0.13	0.30	b.d.	0.06	0.03	0.06	b.d.	b.d.
Gd ₂ O ₃	0.28	1.64	0.12	1.71	5.04	5.17	0.92	1.30
Dy ₂ O ₃	0.29	1.65	0.28	1.99	5.02	5.49	1.85	1.53
Ho ₂ O ₃	b.d.	0.54	0.18	0.75	2.29	2.44	0.44	0.51
Er ₂ O ₃	0.12	1.19	0.34	0.98	2.29	2.75	0.71	0.64
Yb ₂ O ₃	0.01	0.64	0.06	0.88	1.74	1.67	0.42	0.64
FeO	7.05	5.02	2.34	0.89	1.43	0.96	4.73	3.57
MgO	0.20	b.d.	0.01	b.d.	b.d.	b.d.	b.d.	b.d.
MnO	b.d.	b.d.	0.09	0.06	0.17	b.d.	1.75	0.19
CaO	6.63	4.44	1.33	0.54	0.52	0.39	0.24	0.69
K ₂ O	0.06	0.05	b.d.	0.04	0.03	0.02	0.03	0.11
F	0.08	1.14	0.29	0.13	b.d.	b.d.	b.d.	b.d.
Total	95.83	83.06	92.68	91.73	101.09	99.22	90.39	87.40
Structural	l formula							
Р	0.002	0.026	0.021	0.064	0.029	0.047	0.035	0.023
As	0.000	0.030	0.000	0.024	0.387	0.370	0.036	0.024
Si	0.795	0.611	0.904	0.782	0.116	0.127	0.270	0.348
Ti	0.015	0.000	0.000	0.001	0.000	0.000	0.001	0.000
Zr	0.022	0.054	0.003	0.023	0.002	0.002	0.000	0.000
Th	0.343	0.280	0.739	0.371	0.043	0.019	0.003	0.011
U	0.012	0.018	0.100	0.019	0.005	0.003	0.000	0.003
Al	0.203	0.021	0.008	0.034	0.000	0.000	0.048	0.041
Y	0.018	0.227	0.018	0.129	0.309	0.345	0.095	0.092
La	0.017	0.043	0.000	0.004	0.000	0.000	0.001	0.000
Ce	0.047	0.108	0.004	0.003	0.002	0.001	0.358	0.334
Pr	0.005	0.012	0.000	0.128	0.054	0.035	0.001	0.001
Nd	0.022	0.046	0.001	0.011	0.012	0.009	0.005	0.005
Sm	0.006	0.012	0.000	0.006	0.013	0.011	0.002	0.002
Eu	0.002	0.005	0.000	0.001	0.000	0.000	0.000	0.000
Gd	0.003	0.025	0.002	0.014	0.035	0.036	0.006	0.009
Dy	0.003	0.025	0.005	0.016	0.034	0.037	0.012	0.010
Er	0.000	0.008	0.003	0.006	0.015	0.016	0.003	0.003
Н0 УЬ	0.001	0.01 /	0.000	0.008	0.015	0.018	0.005	0.004
YD Ee	0.000	0.009	0.001	0.007	0.011	0.011	0.003	0.004
re Ma	0.211	0.195	0.104	0.03/	0.001	0.034	0.081	0.002
Mg	0.010	0.000	0.000	0.000	0.000	0.000	0.000	0.000
	0.000	0.000	0.004	0.002	0.000	0.000	0.030	0.005
Ca K	0.234	0.221	0.070	0.029	0.023	0.018	0.005	0.013
л Г	0.005	0.005	0.000	0.001	0.001	0.000	0.001	0.003
r Tatel	0.005 2 005	2 024	2 025	1 710	1 165	1 1/0	1 001	1 007
Notes: Th	Lives	ernovita for	2.023 nula calcul	1./17	hasis of 1 or	1.140	a formula no	rmalized
to 1 and			nula calcula		00315 01 4 0Xy	gens, certaille		manzeu
io i catio	1.							

Appendix C. 5 Microprobe analyses of hydrothermal thorite, chernovite and cerianite.

Sample	6497	6497	6497	6497
Mineral	Hingganite	Hingganite	Hingganite	Hingganite
P	0.06	h d	0.05	0.05
SiO ₂	29.43	28 77	29.07	34.88
$\frac{5102}{2r\Omega_2}$	0.01	0.10	b d	0.03
	0.01	0.18	0.12	0.09
	0.23	0.09	0.12	0.29
	0.01	0.09	0.15	1.53
	27.84	26.70	27.62	20.57
	27.04 h d	20.79 h d	0.01	20.37
	0.34	0.38	0.01	0.09
Ce_2O_3	0.34	0.38	0.23	0.40
Nd.O.	0.17	0.12	0.09	0.18
	0.84	1.14	0.80	1.07
Sm_2O_3	1.08	1.12	1.06	0.74
Eu_2O_3	0.12	0.11	0.09	0.13
Gd2O3	4.69	4.15	4.4/	2.78
Dy_2O_3	5.11	4.23	5.06	3.00
H0 ₂ O ₃	2.28	1.99	2.18	1.39
Er ₂ O ₃	3.01	2.64	2.83	2.27
Yb ₂ O ₃	1.78	1.73	1.72	1.44
FeO	7.73	7.09	6.21	10.61
MgO	b.d.	0.13	b.d.	1.14
MnO	0.08	0.12	0.08	0.34
CaO	5.34	5.47	5.87	6.35
K2O	0.03	0.07	0.03	0.29
F	0.14	0.09	0.13	0.29
Total	90.54	86.82	87.98	90.14
Structura	l formula on	the basis of 10	oxygens	
Р	0.003	0.000	0.003	0.002
Si	1.862	1.874	1.886	1.983
Zr	0.000	0.003	0.000	0.001
Th	0.004	0.003	0.002	0.004
U	0.000	0.001	0.002	0.003
Al	0.013	0.022	0.011	0.102
Y	0.937	0.929	0.953	0.622
La	0.000	0.000	0.000	0.002
Ce	0.008	0.009	0.006	0.010
Pr	0.004	0.003	0.002	0.004
Nd	0.019	0.026	0.018	0.022
Sm	0.024	0.025	0.024	0.014
Eu	0.003	0.003	0.002	0.002
Gd	0.098	0.090	0.096	0.052
Dv	0.104	0.089	0.106	0.055
Er	0.046	0.041	0.045	0.025
Ho	0.060	0.054	0.058	0.041
Yh	0.034	0.034	0.034	0.025
Fe	0 409	0 386	0 337	0.504
Μσ	0.000	0.013	0.000	0.096
Mn	0.000	0.013	0.005	0.016
Co	0.004	0.387	0.005	0.387
Ca K	0.302	0.362	0.400	0.007
к Г	0.002	0.000	0.005	0.021
r Tatal	0.028	0.009	0.013	0.020
lotal	4.028	4.009	4.013	4.026

Appendix C. 6 Representative microprobe analyses of hydrothermal hingganite-(Y).

APPENDIX D

Euler rotation parameters used as an input for G-Plates in order to produce the paleoreconstructions presented in Chapter 5.

Moving plate	Age (Ma)	Latitude	Longitude	Angle	Fixed plate
Florida	250	63.19	-13.87	79.87	Congo
Florida	490	63.06	-11.94	80.86	Congo
Florida	540	63.19	-13.87	79.87	Congo
S. China	250	53.52	-99.21	37.90	Congo
S. China	260	41.85	-103.27	40.51	Congo
S. China	270	60.25	-175.17	39.00	Congo
S. China	280	60.92	164.36	30.78	Congo
S. China	290	56.92	168.31	31.11	Congo
S. China	300	53.62	164.62	29.94	Congo
S. China	310	49.96	161.26	28.91	Congo
S. China	320	45.23	141 21	25.42	Congo
S. China	330	42.25	140.57	25.53	Congo
S. China	340	39.29	140.00	25.55	Congo
S. China	350	38.76	128 78	23.61	Congo
S. China	360	37.38	120.70	22.81	Congo
S. China	370	35.40	113.01	22.05	Congo
S. China	380	24.03	126.78	22.13	Congo
S. China	390	11 30	132.00	22.10	Congo
S. China	400	13.94	140.65	25.56	Congo
S. China S. China	410	19.94	152.09	29.30	Congo
S. China	420	1 97	-20.64	-19.18	Congo
S. China S. China	430	14 35	-17 50	-19.10	Congo
S. China S. China	430	29.81	-15.31	-17.97	Congo
S. China S. China	450	29.01 44 95	-12.35	-18.80	Congo
S. China S. China	460	57.92	-8.17	-20.79	Congo
S. China S. China	400	53.67	62.99	-39.62	Congo
S. China S. China	470	45 57	77 14	-64 30	Congo
S. China	540	45 57	77.17	-64 30	Congo
S. China	640	0.00	-117.80	90.20	Congo
S. China	650	-2.19	-92 99	108.88	Congo
S. China	750	-16 75	-94 25	144 23	Congo
Oiangtang	250	25.38	68 23	-30.37	Congo
Qiangtang	260	36.14	44.61	-35.20	Congo
Qiangtang	270	34.53	37.26	-43.97	Congo
Qiangtang	280	30.73	41.85	-54.31	Congo
Qiangtang	290	29.19	43.42	-59.60	Congo
Qiangtang	540	29.19	43.42	-59.60	Congo
Laurentia	250	63.19	-13.87	79.87	Congo
Laurentia	320	63.19	-13.87	79.87	Congo
Laurentia	330	52.34	-20.01	81.57	Congo
Laurentia	, ,,,,			01101	001180
Laurentia	330 340	38.73	-15.87	86.18	Congo
Laurentia	340 350	38.73 29.94	-15.87 -14.54	86.18 91.50	Congo Congo
	340 350 360	38.73 29.94 17.48	-15.87 -14.54 -18.96	86.18 91.50 111.39	Congo Congo Congo
Laurentia	340 350 360 370	38.73 29.94 17.48 8.90	-15.87 -14.54 -18.96 -21.96	86.18 91.50 111.39 140.86	Congo Congo Congo Congo
Laurentia Laurentia	340 350 360 370 380	38.73 29.94 17.48 8.90 11.83	-15.87 -14.54 -18.96 -21.96 -25.00	86.18 91.50 111.39 140.86 141.71	Congo Congo Congo Congo Congo
Laurentia Laurentia Laurentia	340 350 360 370 380 390	38.73 29.94 17.48 8.90 11.83 9.60	-15.87 -14.54 -18.96 -21.96 -25.00 -25.92	86.18 91.50 111.39 140.86 141.71 142.30	Congo Congo Congo Congo Congo Congo

Appendix D. 1 Euler rotation parameters used for the reconstructions presented in Chapter 5.

			_		
Moving plate	Age (Ma)	Latitude	Longitude	Angle	Fixed plate
Laurentia	410	4.11	-23.77	122.42	Congo
Laurentia	420	4.12	-22.16	97.99	Congo
Laurentia	430	5.17	-27.94	81.20	Congo
Laurentia	440	7.75	-35.54	97.98	Congo
Laurentia	450	10.43	-38.78	108.64	Congo
Laurentia	460	14.83	-39.68	110.99	Congo
Laurentia	470	21.50	-44.25	93.69	Congo
Laurentia	480	28.05	-46.77	79.70	Congo
Laurentia	490	36.43	-59.04	63.74	Congo
Laurentia	500	39.32	-88.75	43.90	Congo
Laurentia	510	42.98	-66.44	25.65	Congo
Laurentia	520	11.65	-17.25	22.38	Congo
Laurentia	530	9.54	171.92	-43.53	Congo
Laurentia	540	17.95	166.47	-58.80	Congo
Laurentia	540.1	17.95	166.47	-58.80	Mantle
Laurentia	541	38.59	-159.21	-128.18	Mantle
Laurentia	550	49.80	-178.74	-158.23	Mantle
Laurentia	600	39.27	-158.20	-132.54	Mantle
Laurentia	620	33.62	-149.02	-123.65	Mantle
Laurentia	630	28.57	-142.05	-115.10	Mantle
Laurentia	650	18.00	-130.4 -	104.30	Mantle
Laurentia	750	-1.80	-108.72	-85.95	Mantle
Greenland	2.50	60.48	1.42	69.40	Congo
Greenland	320	60.48	1.42	69 40	Congo
Greenland	330	49.11	-7.55	72 39	Congo
Greenland	340	34 84	-5.18	79 73	Congo
Greenland	350	26.21	-4 67	86 48	Congo
Greenland	360	15.62	-10 41	108 92	Congo
Greenland	370	9.20	-14 45	139.77	Congo
Greenland	380	12 19	-17 49	139.67	Congo
Greenland	390	10.00	-18 45	140.48	Congo
Greenland	400	8 05	_17 33	133.48	Congo
Greenland	410	3 10	-15.82	122.40	Congo
Greenland	420	1 13	-13.32	08 30	Congo
Greenland	430	0.26	-19.39	81 80	Congo
Greenland	4 <u>4</u> 0	0.20 4 75	-26 71	06.33	Congo
Greenland	450	ч.75 8.47	-30.38	105.95	Congo
Greenland	460	13.06	-30.30	105.55	Congo
Greenland	400 470	18 21	-31.33	88 20	Congo
Greenland	4/0	10.31 23.40	-55.25	72 21	Congo
Greenland	400 /00	23.49 30.42	-30.07	12.31 51 60	Congo
Greenland	470 500	20.43 28 00	-40.17	27.04	Congo
Greenland	510	20.90 18 70	-/0.43	52.94 16.64	Congo
Greenland	520	16.70	-43.13 179 15	24.04	Congo
Greeniana Greeniana	520	10.91	-1/8.43	-24.90	Congo
Greenland	550	21.03	-1/0.10	-49.28	Congo
Greenland	540	∠3.04 (92	1//.14	-04.38	Congo
Greenland	541	-0.85	104.33	-65.98	Congo
Greenland	550	29.88	-8.48	69./ <i>3</i>	Congo
Greenland	560	34.10	-16.29	68.88	Congo

Moving plate	Age (Ma)	Latitude	Longitude	Angle	Fixed plate
Greenland	570	26.78	-22.87	79.29	Congo
Greenland	580	19.68	-28.83	91.21	Congo
Greenland	590	16.24	-31.96	102.93	Congo
Greenland	600	13.22	-33.93	115.19	Congo
Greenland	650	-0.83	-46.10	176.67	Congo
Greenland	750	-17.65	-56.37	194.67	Congo
W. Avalonia	250	63.19	-13.87	79.87	Congo
W. Avalonia	320	63.19	-13.87	79.87	Congo
W. Avalonia	330	52.34	-20.01	81.57	Congo
W. Avalonia	340	38.73	-15.87	86.18	Congo
W. Avalonia	350	29.94	-14.54	91.50	Congo
W. Avalonia	360	17.48	-18.96	111.39	Congo
W. Avalonia	370	8.90	-21.96	140.86	Congo
W. Avalonia	380	11.90	-24.83	141.42	Congo
W. Avalonia	390	9.60	-25.92	142.03	Congo
W. Avalonia	400	8.13	-24.99	134.63	Congo
W. Avalonia	410	4.11	-23.77	122.42	Congo
W. Avalonia	420	4.12	-22.16	97.99	Congo
W. Avalonia	430	5.17	-27.94	81.20	Congo
W. Avalonia	440	14.37	-34.41	81.19	Congo
W. Avalonia	450	23.05	-36.98	90.70	Congo
W. Avalonia	460	27.35	-39.43	116.94	Congo
W. Avalonia	470	41.05	-35.10	113.70	Congo
W. Avalonia	471	42.82	-28.83	92.21	Congo
W Avalonia	480	43 67	-22.37	103 39	Congo
W Avalonia	489	46 36	-15 33	93.16	Congo
W Avalonia	490	46.24	-15.19	94 36	Congo
W Avalonia	500	41.37	-9.58	86.86	Congo
W Avalonia	510	36.22	-4 24	82.97	Congo
W Avalonia	540	31.02	-6.35	89.86	Congo
W Avalonia	550	28.56	-10 59	101.28	Congo
W Avalonia	560	24.91	-16.57	111 74	Congo
W. Avalonia	570	21.93	-20.98	120 44	Congo
W Avalonia	580	19 10	-24 72	129.44	Congo
W Avalonia	590	16.68	-28.48	138 78	Congo
W Avalonia	600	14 17	-33 19	151.61	Congo
W Avalonia	610	11 39	-35.81	161.01	Congo
W Avalonia	620	9.23	-37 74	171 68	Congo
W Avalonia	640	5.15	-42.36	191.85	Congo
W Avalonia	650	3 3 2	_44 02	201 75	Congo
W Avalonia	660	0.58	-47.67	201.75	Congo
W Avelonie	670	-0.57	_47.80	200.14	Congo
w. Avalonia W. Avalonia	680	-0.57	-77.07	205.07	Congo
w. Avalonia W. Avalonia	600	_/ 02	-57.21	210.11	Congo
w. Avalonia W. Avalonia	700	8 00	-50.00	210.04	Congo
w. Avalonia W. Avalonia	710	-0.90	-37.03	227.93 227.80	Congo
w. Avalonia W. Avalonia	720	-10.34 8 70	-02.07	237.00	Congo
w. Avalonia W. Avalonia	730	-0.70	-01.31	240.47	Congo
w. Avalonia	740	-7.14	-04.49	243.31	Congo
w. Avaionia	/50	-8.90	-07.55	231.49	Congo

Moving plate	Age (Ma)	Latitude	Longitude	Angle	Fixed plate
Florida	250	63.19	-13.87	79.87	Congo
Florida	540	63.19	-13.87	79.87	Congo
Florida	550	54.01	-21.25	90.94	Congo
Florida	560	43.75	-35.11	103.27	Congo
Florida	570	38.53	-39.27	107.55	Congo
Florida	580	33.06	-42.35	110.89	Congo
Florida	590	28.60	-44.07	113.92	Congo
Florida	600	22.01	-50.76	131.64	Congo
Florida	650	4.12	-55.54	151.60	Congo
Florida	750	-17.27	-65.60	163.13	Congo
Amazonia	250	50.00	-32.50	55.08	Congo
Amazonia	540	50.00	-32.50	55.08	Congo
Amazonia	550	44.26	-29.27	59.51	Congo
Amazonia	650	-10.67	-44.66	158.15	Congo
Amazonia	750	-30.25	-49.22	170.16	Congo
Colorado	250	47.50	-33.30	57.30	Congo
Colorado	540	47.50	-33.30	57.30	Congo
Colorado	550	23.18	-26.69	61.42	Congo
Colorado	560	37.48	-28.94	52.71	Congo
Colorado	570	26.12	-32.14	59.22	Congo
Colorado	580	16.42	-33.46	66.28	Congo
Colorado	590	7.93	-32.25	72 23	Congo
Colorado	600	-9.30	-37.42	103 70	Congo
Colorado	610	-2.87	-41.80	97.21	Congo
Colorado	620	-5.29	-40.64	110.10	Congo
Colorado	630	-3.2)	40.04	122.60	Congo
Colorado	640	-7.22	-40.00	122.00	Congo
Colorado	650	-10.40	-39.47	133.95	Congo
Colorado	660	-12.11	-36.72	144.03	Congo
Colorado	670	-14.55	-30.83	149.94	Congo
Colorado	690	18.02	-37.22	162.63	Congo
Colorado	600	-16.02	-34.31	166.09	Congo
Colorado	090 700	-21.24	-31.00	100.98	Congo
Colorado	700	-22.40	-29.80	1/3.48	Congo
Colorado	710	-24.08	-27.33	160.12	Congo
Colorado Colorado	720	-15.85	-34.35	151.51	Congo
Colorado	/30	-17.02	-32.82	150.50	Congo
Colorado	740	-10.04	-32.34	155.57	Congo
Colorado	/50	-14.98	-32.65	155.82	Congo
Patagonia	250	47.50	-33.30	63.00	Congo
Patagonia	260	49.12	-30.35	62.28	Congo
Patagonia	270	50.68	-27.15	61.66	Congo
Patagonia	540	50.68	-27.15	61.66	Congo
Patagonia	550	0.43	-24.86	72.53	Congo
Patagonia	560	20.61	-28.71	58.90	Congo
Patagonia	570	9.19	-30.17	66.51	Congo
Patagonia	580	0.63	-28.83	75.31	Congo
Patagonia	590	-5.51	-27.66	79.73	Congo
Patagonia	600	-24.95	-31.65	112.61	Congo
Patagonia	610	-6.30	-40.82	106.62	Congo

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Moving plate	Age (Ma)	Latitude	Longitude	Angle	Fixed plate
Patagonia	620	-13.26	-36.69	114.78	Congo
Patagonia	630	-18.43	-34.27	130.70	Congo
Patagonia	640	-21.43	-32.47	138.96	Congo
Patagonia	650	-25.04	-30.28	158.70	Congo
Patagonia	660	-28.50	-30.73	161.69	Congo
Patagonia	670	-28.59	-30.56	164.28	Congo
Patagonia	680	-29.42	-29.69	164.77	Congo
Patagonia	690	-30.99	-28.77	166.42	Congo
Patagonia	700	-29.81	-27.56	170.17	Congo
Patagonia	710	-31.72	-26.62	169.15	Congo
Patagonia	720	-31.70	-24.44	171.49	Congo
Patagonia	730	-33.29	-23.76	171.79	Congo
Patagonia	740	-32.83	-22.98	175.90	Congo
Patagonia	750	-32.06	-22.54	179.48	Congo
Baltica	250	46.04	3.89	58.20	Congo
Baltica	320	46.04	3.89	58.20	Congo
Baltica	330	34.10	-2.57	64.85	Congo
Baltica	340	20.86	0.32	76 70	Congo
Baltica	350	13 46	1 19	85.23	Congo
Baltica	360	6.41	-3 38	110.96	Congo
Baltica	370	3 38	-6.50	143.63	Congo
Baltica	380	5.50 6.45	0.30	142.05	Congo
Baltica	300	0.43	-9.52	142.80	Congo
Daliica	400	1.82	-10.34	128 70	Congo
Daliica Daltiaa	400	1.62	-9.42	130.70	Congo
	410	4.01 9.72	171.02	-120.39	Congo
	420	0./J	1/5.11	-105.51	Congo
Baltica	430	11.72	167.67	-89.36	Congo
Baltica	440	14.27	168.23	-84.70	Congo
Baltica	450	19.32	167.39	-/2.49	Congo
Baltica	460	28.57	168.91	-58.33	Congo
Baltica	470	46.54	174.95	-47.21	Congo
Baltica	480	75.11	-170.20	-47.40	Congo
Baltica	490	77.58	37.73	-69.40	Congo
Baltica	500	65.26	33.54	-105.39	Congo
Baltica	510	64.93	10.76	-113.77	Congo
Baltica	520	65.74	-22.11	-118.37	Congo
Baltica	530	63.24	-35.90	-135.10	Congo
Baltica	540	67.21	-74.31	-112.84	Congo
Baltica	550	16.46	-151.21	-59.70	Congo
Baltica	560	13.60	-163.59	-53.94	Congo
Baltica	570	12.48	-169.65	-66.90	Congo
Baltica	580	11.30	-175.83	-81.25	Congo
Baltica	590	8.50	-177.81	-93.28	Congo
Baltica	650	1.00	167.26	-169.16	Congo
Baltica	750	7.87	157.43	-202.74	Congo
Scotland	250	46.04	3.89	58.20	Congo
Scotland	320	46.04	3.89	58.20	Congo
Scotland	330	34.10	-2.57	64.85	Congo
Scotland	340	20.86	0.32	76.70	Congo

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Moving plate	Age (Ma)	Latitude	Longitude	Angle	Fixed plate
Scotland	350	13.46	1.19	85.23	Congo
Scotland	360	6.41	-3.38	110.96	Congo
Scotland	370	3.38	-6.50	143.63	Congo
Scotland	380	6.45	-9.32	142.86	Congo
Scotland	390	4.42	-10.34	144.34	Congo
Scotland	400	1.82	-9.42	138.70	Congo
Scotland	410	4.01	171.62	-128.59	Congo
Scotland	420	8.73	173.11	-105.51	Congo
Scotland	430	11.72	167.67	-89.36	Congo
Scotland	440	4.92	160.65	-103.44	Congo
Scotland	450	0.07	157.66	-111.91	Congo
Scotland	460	4.54	-22.97	111.71	Congo
Scotland	470	7.08	-26.86	91.69	Congo
Scotland	480	8.92	-28.65	74.60	Congo
Scotland	490	10.71	-38.96	55.83	Congo
Scotland	500	1.50	115.95	-36.25	Congo
Scotland	510	28.48	139.21	-27.30	Congo
Scotland	520	43.27	175.34	-38.79	Congo
Scotland	530	37.42	-175.86	-62.13	Congo
Scotland	540	37.20	179.43	-78.39	Congo
Iberia	250	51.77	67.88	3.42	Congo
Iberia	320	51.77	67.88	3.42	Congo
Iberia	330	10.76	-166.02	-15.90	Congo
Iberia	340	14.04	-154.13	-33.63	Congo
Iberia	350	12.65	-153.64	-41.98	Congo
Iberia	360	9.43	-164.23	-64.97	Congo
Iberia	370	28.20	-144.20	-64.10	Congo
Iberia	380	-48.10	83.30	58.90	Congo
Iberia	390	47.30	-44.60	-88.90	Congo
Theria	400	44.80	-26.3 -	101.80	Congo
Iberia	410	40.50	-11.6 -	113.90	Congo
Theria	420	39.51	-6.66	-116.57	Congo
Theria	520	38.41	-3.36	-163.12	Congo
Iberia	540	39.29	-5.98	-141.27	Congo
Iberia	550	40.25	15.98	-270.20	Congo
Iberia	650	10.14	-36.29	-215 73	Congo
Iberia	750	26.76	-76.07	-151 41	Congo
Armorica	250	46.04	3 89	58 20	Congo
Armorica	320	46.04	3.89	58.20	Congo
Armorica	330	34 10	-2 57	64.85	Congo
Armorica	340	20.86	0.32	76 70	Congo
Armorico	350	17.03	0.32	82.86	Congo
Armorico	360	11.58	5.17	104.02	Congo
Armorica	370	2.60	-5.17	00.80	Congo
Armorico	380	-2.00	174 10	-22.00 82.70	Congo
Armorico	200	-7.20	1/4.10	-05.70	Congo
Ai morico	370 400	16.00	-4.00	20.50	Congo
Ai morico	400	20.40	2.30 155 50	22.20	Congo
Ai morica Anmorica	410	-37.40	-155.50	-23.30	Congo
Armorica	420	54.59	38.40	21.32	Congo

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Moving plate	Age (Ma)	Latitude	Longitude	Angle	Fixed plate
Armorica	540	34.42	119.45	7.45	Congo
Armorica	550	0.00	-110.10	95.80	Congo
Armorica	650	50.53	-89.22	-100.24	Congo
Armorica	750	-38.95	56.23	128.47	Congo
E. Avalonia	250	46.04	3.89	58.20	Congo
E. Avalonia	320	46.04	3.89	58.20	Congo
E. Avalonia	330	34.10	-2.57	64.85	Congo
E. Avalonia	340	20.86	0.32	76.70	Congo
E. Avalonia	350	13.46	1.19	85.23	Congo
E. Avalonia	360	6.41	-3.38	110.96	Congo
E. Avalonia	370	3.38	-6.50	143.63	Congo
E. Avalonia	380	6.45	-9.32	142.86	Congo
E. Avalonia	390	4.42	-10.34	144.34	Congo
E. Avalonia	400	1.82	-9.42	138.70	Congo
E. Avalonia	410	4.01	171.62	-128.59	Congo
E. Avalonia	420	8.73	173.11	-105.51	Congo
E. Avalonia	430	11.72	167.67	-89.36	Congo
E. Avalonia	440	14.27	168.23	-84.70	Congo
E. Avalonia	450	-3.47	166.10	-93.75	Congo
E. Avalonia	460	-13.69	167.80	-90.77	Congo
E. Avalonia	470	20.79	-6.83	81.19	Congo
E. Avalonia	480	18.61	6.77	69.86	Congo
E. Avalonia	481	17.91	7.59	69.50	Congo
E. Avalonia	490	22.53	10.22	63.58	Congo
E. Avalonia	510	3.59	26.79	56.18	Congo
E. Avalonia	540	0.90	27.08	59.49	Congo
E. Avalonia	550	-0.34	28.94	67.57	Congo
E. Avalonia	560	5.16	16.38	83.17	Congo
E. Avalonia	570	5.18	11.58	91.37	Congo
E. Avalonia	580	6.77	3.97	105.24	Congo
E. Avalonia	590	2.30	5.80	105.97	Congo
E. Avalonia	600	4.09	0.80	116.72	Congo
E. Avalonia	650	6.02	-13.12	179.96	Congo
E. Avalonia	660	4.17	-11.81	178.15	Congo
E. Avalonia	670	2.65	-13.10	180.50	Congo
E. Avalonia	680	3.16	-20.16	198.91	Congo
E. Avalonia	700	3.51	-24.38	211.20	Congo
E. Avalonia	710	3.74	-29.99	224.32	Congo
E. Avalonia	720	4.07	-28.06	222.52	Congo
E. Avalonia	730	6 4 9	-32.20	233 30	Congo
E. Avalonia	740	6 4 4	-37.22	239.34	Congo
E. Avalonia	750	7 48	-43.04	239.31	Congo
Novava	250	49 08	2.28	60.60	Congo
Novaya	320	49.08	2.28	60.60	Congo
Novava	330	37 14	-4 34	66 50	Congo
Novava	340	23 59	-1 47	76.85	Congo
Novava	350	15.87	-0.56	85 51	Congo
Novava	360	8 01	-5 21	110.69	Congo
1.0vaya Novaya	370	4.25	-3.21 -8.41	143 10	Congo
1 waya	510	7.20	0.71	1-1,10	Congo

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Noving plate	Age (Ma)		Longitude	Angle	Fixed plate
INOVAYA	380	/.51	-11.26	142.52	Congo
Novaya	390	5.25	-12.27	143.85	Congo
Novaya	400	2.77	-11.31	137.41	Congo
Novaya	410	2.88	169.79	-127.55	Congo
Novaya	420	7.08	171.38	-104.19	Congo
Novaya	430	9.66	165.98	-87.82	Congo
Novaya	440	12.06	166.58	-82.38	Congo
Novaya	450	16.76	165.82	-70.49	Congo
Novaya	460	25.55	167.54	-55.82	Congo
Novaya	470	43.70	174.15	-43.90	Congo
Novaya	480	74.27	-168.07	-43.47	Congo
Novaya	490	77.14	32.21	-65.66	Congo
Novaya	500	64.91	30.45	-102.00	Congo
Novaya	510	64.49	7.64	-110.52	Congo
Novaya	520	65.09	-25.16	-115.19	Congo
Novaya	530	62.67	-38.71	-131.95	Congo
Novaya	540	59.19	-40.36	-147.88	Congo
Novaya	550	-18.86	24.61	147.95	Mantle
Novaya	650	-9.92	64.93	145.54	Mantle
Novaya	750	-8.59	86.72	135.34	Mantle
Siberia	250	47.59	3.10	59.38	Congo
Siberia	260	48.22	2.61	60.90	Congo
Siberia	270	48.84	2.13	60.82	Congo
Siberia	280	49.44	1.64	61.55	Congo
Siberia	290	50.23	1.07	62.42	Congo
Siberia	300	50.99	0.49	63.30	Congo
Siberia	310	51.74	-0.09	64.20	Congo
Siberia	320	52.46	-0.67	65.10	Congo
Siberia	330	39.38	-3.08	62.59	Congo
Siberia	340	22.43	4.98	67.10	Congo
Siberia	350	21.45	3.04	68.86	Congo
Siberia	360	20.90	-4.62	76.42	Congo
Siberia	370	20.47	-9.82	98.57	Congo
Siberia	380	28.43	-13.96	96.64	Congo
Siberia	390	35.05	-11.14	91.83	Congo
Siberia	400	36.84	-6.25	92.28	Congo
Siberia	410	41.82	2.43	86.66	Congo
Siberia	420	57.21	22.75	75.91	Congo
Siberia	430	68.26	50.31	64.91	Congo
Siberia	440	76.32	16.57	52.80	Congo
Siberia	450	70.32	-42.78	43.22	Congo
Siberia	460	52.14	-37 70	46.13	Congo
Siberia	470	60.25	-20.27	48.70	Congo
Siberia	480	76.95	11.23	52.64	Congo
Siberia	490	71.96	149 46	47.95	Congo
Siberia	500	61.81	145 39	43 87	Congo
Siberia	510	56.69	100 51	46.80	Congo
Siberia	520	46.16	63 47	47 50	Congo
Siberia	530	27.04	50.10	47.50	Congo

Moving plate Age (Ma) Latitude Longitude Angle Fixed plate Siberia 550 70.10 -75.49 161.90 Congo Siberia 550 70.10 -75.49 161.90 Congo Siberia 560 72.76 81.25 159.62 Congo Siberia 570 65.92 -85.41 156.72 Congo Siberia 750 -7.97 -98.78 152.43 Congo India 540 29.93 42.29 -60.47 Congo India 550 40.11 31.71 -50.76 Congo India 550 47.11 17.11 -8.86 Congo Arabia 250 27.12 47.50 136.16 Congo Arabia 550 27.12 47.50 136.16 Congo Arabia 560 26.39 52.04 128.63 Congo Arabia 560 27.06 52.45 122.21						
Siberia 540 3.59 45.81 55.49 Congo Siberia 550 70.10 -75.49 161.90 Congo Siberia 570 65.92 -85.41 156.72 Congo Siberia 580 58.29 -90.50 151.97 Congo Siberia 750 -7.97 -98.78 152.43 Congo Siberia 750 -7.97 -98.78 152.43 Congo India 250 29.93 42.29 -60.47 Congo India 550 40.11 31.71 -50.76 Congo India 650 -8.70 54.96 -84.75 Congo Arabia 250 37.11 17.11 -8.86 Congo Arabia 560 27.12 47.50 136.16 Congo Arabia 570 24.03 50.55 133.97 Congo Arabia 590 27.06 52.45 122.21 Congo <th>Moving plate</th> <th>Age (Ma)</th> <th>Latitude</th> <th>Longitude</th> <th>Angle</th> <th>Fixed plate</th>	Moving plate	Age (Ma)	Latitude	Longitude	Angle	Fixed plate
Siberia 550 70.10 -75.49 161.90 Congo Siberia 560 72.76 -81.25 159.62 Congo Siberia 570 65.92 -85.41 156.72 Congo Siberia 650 13.31 -94.77 155.76 Congo India 250 29.93 42.29 -60.47 Congo India 550 40.11 31.71 -50.76 Congo India 650 -8.70 54.96 -84.75 Congo India 750 19.29 45.50 -77.42 Congo Arabia 540 37.11 17.11 -8.86 Congo Arabia 550 27.12 47.50 136.16 Congo Arabia 560 23.03 49.90 129.48 Congo Arabia 570 24.03 50.55 133.97 Congo Arabia 590 27.06 52.45 122.21 Congo	Siberia	540	3.59	45.81	55.49	Congo
Siberia 560 72.76 -81.23 159.62 Congo Siberia 570 65.92 -85.41 156.72 Congo Siberia 650 13.31 -94.77 155.76 Congo Siberia 750 -7.97 -98.78 152.43 Congo India 250 29.93 42.29 -60.47 Congo India 550 40.11 31.71 -50.76 Congo India 650 -8.70 54.96 -84.75 Congo India 750 19.29 45.50 -77.42 Congo Arabia 250 37.11 17.11 -8.86 Congo Arabia 550 27.12 47.50 133.17 Congo Arabia 560 26.30 49.90 129.48 Congo Arabia 590 27.06 52.45 135.70 Congo Arabia 600 5.62 54.00 134.20 Congo	Siberia	550	70.10	-75.49	161.90	Congo
Siberia 570 65.92 -85.41 156.72 Congo Siberia 580 58.29 -90.50 151.97 Congo Siberia 750 -7.97 -98.78 152.43 Congo India 250 29.93 42.29 -60.47 Congo India 550 40.11 31.71 -50.76 Congo India 750 19.29 45.50 -77.42 Congo Arabia 250 37.11 17.11 -8.86 Congo Arabia 550 27.12 47.50 136.16 Congo Arabia 560 26.30 49.90 129.48 Congo Arabia 590 27.06 52.45 122.21 Congo Arabia 591 26.39 52.04 128.63 Congo Arabia 602 19.06 53.78 134.70 Congo Arabia 620 10.25 62.24 106.19 Congo	Siberia	560	72.76	-81.25	159.62	Congo
Siberia 580 58.29 -90.50 151.97 Congo Siberia 750 7.97 98.78 152.43 Congo India 250 29.93 42.29 -60.47 Congo India 550 40.11 31.71 -50.76 Congo India 650 -8.70 54.96 -84.75 Congo India 750 7.92 45.50 -77.42 Congo India 750 19.29 45.50 -77.42 Congo Arabia 540 37.11 17.11 -8.56 Congo Arabia 550 27.12 47.50 136.16 Congo Arabia 570 24.03 50.55 133.97 Congo Arabia 590 27.06 52.45 122.21 Congo Arabia 602 19.06 53.78 134.70 Congo Arabia 610 12.36 56.24 134.25 Congo	Siberia	570	65.92	-85.41	156.72	Congo
Siberia 650 13.31 -94.77 155.76 Congo Siberia 750 -7.97 -98.78 152.43 Congo India 250 29.93 42.29 -60.47 Congo India 550 40.11 31.71 -50.76 Congo India 650 -8.70 54.96 -84.75 Congo Arabia 250 37.11 17.11 -8.86 Congo Arabia 540 37.11 17.11 -8.86 Congo Arabia 550 27.12 47.50 136.16 Congo Arabia 570 24.03 50.55 133.97 Congo Arabia 590 27.06 52.45 122.21 Congo Arabia 600 5.62 54.00 134.20 Congo Arabia 610 12.36 Congo Arabia 620 10.25 62.24 106.19 Congo Arabia 620 10.25	Siberia	580	58.29	-90.50	151.97	Congo
Siberia 750 -7.97 -98.78 152.43 Congo India 250 29.93 42.29 -60.47 Congo India 540 29.93 42.29 -60.47 Congo India 550 40.11 31.71 -50.76 Congo India 750 19.29 45.50 -7.42 Congo Arabia 250 37.11 17.11 -8.86 Congo Arabia 560 26.30 49.90 129.48 Congo Arabia 570 27.12 47.50 136.16 Congo Arabia 570 27.06 52.45 122.21 Congo Arabia 590 27.06 52.45 122.21 Congo Arabia 602 19.06 53.78 134.70 Congo Arabia 610 12.36 56.24 134.25 Congo Arabia 610 12.36 56.24 134.25 Congo <tr< th=""><th>Siberia</th><th>650</th><th>13.31</th><th>-94.77</th><th>155.76</th><th>Congo</th></tr<>	Siberia	650	13.31	-94.77	155.76	Congo
India 250 29.93 42.29 -60.47 Congo India 540 29.93 42.29 -60.47 Congo India 550 40.11 31.71 -50.76 Congo India 750 19.29 45.50 -77.42 Congo Arabia 250 37.11 17.11 -8.86 Congo Arabia 560 26.30 49.90 129.48 Congo Arabia 570 24.03 50.55 133.97 Congo Arabia 590 27.06 52.45 122.21 Congo Arabia 591 26.39 52.04 128.63 Congo Arabia 600 5.62 40.01 124.63 Congo Arabia 610 12.36 56.24 134.20 Congo Arabia 610 12.36 56.24 134.25 Congo Arabia 630 9.43 64.66 100.28 Congo	Siberia	750	-7.97	-98.78	152.43	Congo
India 540 29.93 42.29 -60.47 Congo India 550 40.11 31.71 -50.76 Congo India 650 -8.70 54.96 -84.75 Congo Arabia 250 37.11 17.11 -8.56 Congo Arabia 550 27.12 47.50 136.16 Congo Arabia 570 24.03 50.55 133.97 Congo Arabia 580 25.36 51.53 135.70 Congo Arabia 590 27.06 52.45 122.48 Congo Arabia 600 5.62 54.00 134.20 Congo Arabia 610 12.36 56.24 106.19 Congo Arabia 610 12.36 56.24 106.19 Congo Arabia 630 9.43 64.66 100.28 Congo Arabia 650 9.37 63.71 124.70 Congo	India	250	29.93	42.29	-60.47	Congo
India 550 40.11 31.71 -50.76 Congo India 650 -8.70 54.96 -84.75 Congo Arabia 250 37.11 17.11 -8.86 Congo Arabia 540 37.11 17.11 -8.86 Congo Arabia 550 27.12 47.50 136.16 Congo Arabia 570 24.03 50.55 133.97 Congo Arabia 570 24.03 50.55 133.97 Congo Arabia 590 27.06 52.45 122.21 Congo Arabia 600 5.62 54.00 134.20 Congo Arabia 610 12.36 56.24 134.25 Congo Arabia 610 12.36 56.24 134.25 Congo Arabia 650 9.37 63.71 124.70 Congo Arabia 650 9.37 63.71 124.70 Congo	India	540	29.93	42.29	-60.47	Congo
India 650 -8.70 54.96 -84.75 Congo India 750 19.29 45.50 -77.42 Congo Arabia 250 37.11 17.11 -8.86 Congo Arabia 540 37.11 17.11 -8.86 Congo Arabia 550 27.12 47.50 136.16 Congo Arabia 560 26.30 49.90 129.48 Congo Arabia 590 27.06 52.45 122.21 Congo Arabia 591 26.39 52.04 128.63 Congo Arabia 600 5.62 54.00 134.20 Congo Arabia 610 12.36 56.24 134.25 Congo Arabia 630 9.43 64.66 100.28 Congo Arabia 650 9.37 63.71 124.70 Congo Arabia 650 9.37 63.71 724.70 Congo	India	550	40.11	31.71	-50.76	Congo
India 750 19.29 45.50 -77.42 Congo Arabia 250 37.11 17.11 -8.56 Congo Arabia 550 27.12 47.50 136.16 Congo Arabia 560 26.30 49.90 129.48 Congo Arabia 570 24.03 50.55 133.97 Congo Arabia 590 27.06 52.45 122.21 Congo Arabia 600 5.62 54.00 134.20 Congo Arabia 602 19.06 53.78 134.70 Congo Arabia 610 12.36 56.24 106.19 Congo Arabia 620 10.25 62.24 106.19 Congo Arabia 650 9.37 63.71 124.70 Congo Arabia 650 9.37 63.71 124.70 Congo Arabia 750 35.38 -29.84 -99.90 Congo <tr< th=""><th>India</th><th>650</th><th>-8.70</th><th>54.96</th><th>-84.75</th><th>Congo</th></tr<>	India	650	-8.70	54.96	-84.75	Congo
Arabia 250 37.11 17.11 -8.56 Congo Arabia 540 37.11 17.11 -8.86 Congo Arabia 560 26.30 49.90 129.48 Congo Arabia 570 24.03 50.55 133.97 Congo Arabia 580 25.36 51.53 135.70 Congo Arabia 590 27.06 52.45 122.21 Congo Arabia 600 5.62 54.00 134.20 Congo Arabia 610 12.36 56.24 134.25 Congo Arabia 610 12.36 56.24 134.25 Congo Arabia 620 10.25 62.24 106.19 Congo Arabia 650 9.37 63.71 124.70 Congo Arabia 750 35.38 -29.84 -99.90 Congo Taurides 250 52.83 -11.13 7.73 Congo <	India	750	19.29	45.50	-77.42	Congo
Arabia 540 37.11 17.11 -8.86 Congo Arabia 550 27.12 47.50 136.16 Congo Arabia 570 24.03 50.55 133.97 Congo Arabia 570 24.03 50.55 133.97 Congo Arabia 580 25.36 51.53 135.70 Congo Arabia 590 27.06 52.45 122.21 Congo Arabia 600 5.62 54.00 134.20 Congo Arabia 610 12.36 56.24 134.25 Congo Arabia 610 12.36 56.24 134.25 Congo Arabia 630 9.43 64.66 100.28 Congo Arabia 650 9.37 63.71 124.70 Congo Arabia 750 35.38 -29.84 -99.90 Congo Arabia 750 35.34 17.69 -7.89 Congo <tr< th=""><th>Arabia</th><th>250</th><th>37.11</th><th>17.11</th><th>-8.56</th><th>Congo</th></tr<>	Arabia	250	37.11	17.11	-8.56	Congo
Arabia 550 27.12 47.50 136.16 Congo Arabia 560 26.30 49.90 129.48 Congo Arabia 570 24.03 50.55 133.97 Congo Arabia 590 27.06 52.45 122.21 Congo Arabia 591 26.39 52.04 128.63 Congo Arabia 600 5.62 54.00 134.20 Congo Arabia 602 19.06 53.78 134.70 Congo Arabia 610 12.36 56.24 106.19 Congo Arabia 630 9.43 64.66 100.28 Congo Arabia 650 9.37 63.71 124.70 Congo Arabia 750 35.38 -29.84 -99.90 Congo Arabia 750 35.34 17.69 -7.89 Congo Arabia 750 35.34 17.69 -7.89 Congo <tr< th=""><th>Arabia</th><th>540</th><th>37.11</th><th>17.11</th><th>-8.86</th><th>Congo</th></tr<>	Arabia	540	37.11	17.11	-8.86	Congo
Arabia 560 26.30 49.90 129.48 Congo Arabia 570 24.03 50.55 133.97 Congo Arabia 580 25.36 51.53 135.70 Congo Arabia 590 27.06 52.45 122.21 Congo Arabia 600 5.62 54.00 134.20 Congo Arabia 602 19.06 53.78 134.70 Congo Arabia 610 12.36 56.24 134.25 Congo Arabia 620 10.25 62.24 106.19 Congo Arabia 630 9.43 64.66 100.28 Congo Arabia 650 9.37 63.71 124.70 Congo Arabia 750 35.38 -29.84 -99.90 Congo Arabia 750 35.34 17.69 -7.89 Congo Alborz 250 52.83 -11.13 7.73 Congo <tr< th=""><th>Arabia</th><th>550</th><th>27.12</th><th>47.50</th><th>136.16</th><th>Congo</th></tr<>	Arabia	550	27.12	47.50	136.16	Congo
Arabia 570 24.03 50.55 133.97 Congo Arabia 580 25.36 51.53 135.70 Congo Arabia 590 27.06 52.45 122.21 Congo Arabia 600 5.62 54.00 134.20 Congo Arabia 602 19.06 53.78 134.70 Congo Arabia 610 12.36 56.24 106.19 Congo Arabia 620 10.25 62.24 106.19 Congo Arabia 630 9.43 64.66 100.28 Congo Arabia 650 9.37 63.71 124.70 Congo Arabia 750 35.38 -29.84 -99.00 Congo Taurides 540 37.07 17.13 -8.83 Congo Alborz 260 18.37 32.21 -3.18 Congo Alborz 270 35.34 17.69 -7.89 Congo <t< th=""><th>Arabia</th><th>560</th><th>26.30</th><th>49.90</th><th>129.48</th><th>Congo</th></t<>	Arabia	560	26.30	49.90	129.48	Congo
Arabia 580 25.36 51.53 135.70 Congo Arabia 590 27.06 52.45 122.21 Congo Arabia 591 26.39 52.04 128.63 Congo Arabia 600 5.62 54.00 134.20 Congo Arabia 602 19.06 53.78 134.70 Congo Arabia 610 12.36 56.24 134.25 Congo Arabia 620 10.25 62.24 106.19 Congo Arabia 650 9.37 63.71 124.70 Congo Arabia 750 35.38 -29.84 -99.90 Congo Taurides 250 37.07 17.13 -8.83 Congo Alborz 260 18.37 32.21 -3.18 Congo Alborz 270 35.34 17.69 -7.89 Congo Afghan 260 26.61 60.32 -61.61 Congo <	Arabia	570	24.03	50.55	133.97	Congo
Arabia59027.0652.45122.21CongoArabia59126.3952.04128.63CongoArabia6005.6254.00134.20CongoArabia60219.0653.78134.70CongoArabia61012.3656.24134.25CongoArabia62010.2562.24106.19CongoArabia6309.4364.66100.28CongoArabia6509.3763.71124.70CongoArabia75035.38-29.84-99.90CongoArabia75035.38-29.84-99.90CongoArabia75035.38-29.84-99.90CongoArabia75035.38-29.84-99.90CongoArabia75035.38-29.84-99.90CongoArabia75035.38-11.137.73CongoAlborz26018.3732.21-3.18CongoAlborz27035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAfghan26026.6160.32-61.61CongoAfghan26025.2369.03-55.70CongoAfghan27027.0455.58-66.30CongoAfghan29027.0952.84-68.62CongoAfghan29027.9145.2859.59Congo	Arabia	580	25.36	51.53	135.70	Congo
Arabia59126.3952.04128.63CongoArabia6005.6254.00134.20CongoArabia61012.3656.24134.70CongoArabia61012.3656.24134.25CongoArabia62010.2562.24106.19CongoArabia6309.4364.66100.28CongoArabia6509.3763.71124.70CongoArabia75035.38-29.84-99.90CongoArabia75037.0717.13-8.83CongoArabia75035.38-11.137.73CongoArabia75035.34-17.69-7.89CongoAlborz25052.83-11.137.73CongoAlborz26018.3732.21-3.18CongoAlborz27035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz54027.0853.73-66.30CongoAfghan26026.6160.32-61.61CongoAfghan28027.0853.73-67.75CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62Congo	Arabia	590	27.06	52.45	122.21	Congo
Arabia6005.6254.00134.20CongoArabia60219.0653.78134.70CongoArabia61012.3656.24134.25CongoArabia62010.2562.24106.19CongoArabia6309.4364.66100.28CongoArabia6509.3763.71124.70CongoArabia75035.38-29.84-99.90CongoArabia75035.38-29.84-99.90CongoTaurides25037.0717.13-8.83CongoAlborz25052.83-11.137.73CongoAlborz26018.3732.21-3.18CongoAlborz27035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz54025.2369.03-55.70CongoAfghan26026.6160.32-61.61CongoAfghan27027.0455.58-66.30CongoAfghan28027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62Congo<	Arabia	591	26.39	52.04	128.63	Congo
Arabia60219.0653.78134.70CongoArabia61012.3656.24134.25CongoArabia62010.2562.24106.19CongoArabia6309.4364.66100.28CongoArabia6509.3763.71124.70CongoArabia75035.38-29.84-99.90CongoArabia75035.38-29.84-99.90CongoTaurides25037.0717.13-8.83CongoTaurides54037.0717.13-8.83CongoAlborz25052.83-11.137.73CongoAlborz26018.3732.21-3.18CongoAlborz27035.3417.69-7.89CongoAlborz5025.2369.03-55.70CongoAlborz54035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlbar26026.6160.32-61.61CongoAfghan27027.0455.58-66.30CongoAfghan28027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan24027.0952.84-68.62CongoAfghan24027.0952.84-68.62Congo <th>Arabia</th> <th>600</th> <th>5.62</th> <th>54.00</th> <th>134.20</th> <th>Congo</th>	Arabia	600	5.62	54.00	134.20	Congo
Arabia61012.3656.24134.25CongoArabia62010.2562.24106.19CongoArabia6309.4364.66100.28CongoArabia6509.3763.71124.70CongoArabia75035.38-29.84-99.90CongoTaurides25037.0717.13-8.83CongoTaurides54037.0717.13-8.83CongoAlborz25052.83-11.137.73CongoAlborz26018.3732.21-3.18CongoAlborz27035.3417.69-7.89CongoAlborz27035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz26026.6160.32-61.61CongoAfghan26026.6160.32-61.61CongoAfghan28027.0853.73-67.75CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan24027.0952.84-68.62Congo<	Arabia	602	19.06	53.78	134.70	Congo
Arabia62010.2562.24106.19CongoArabia6309.4364.66100.28CongoArabia6509.3763.71124.70CongoArabia75035.38-29.84-99.90CongoTaurides25037.0717.13-8.83CongoTaurides54037.0717.13-8.83CongoAlborz25052.83-11.137.73CongoAlborz26018.3732.21-3.18CongoAlborz27035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlbar26026.6160.32-61.61CongoAfghan26026.6160.32-61.61CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan54027.0952.84-68.62CongoAfghan54027.9145.28-58.26CongoHimalayas280-28.88-136.1159.25CongoHimalayas290-29.12-136.5859.59Cong	Arabia	610	12.36	56.24	134.25	Congo
Arabia6309.4364.66100.28CongoArabia6509.3763.71124.70CongoArabia75035.38-29.84-99.90CongoTaurides25037.0717.13-8.83CongoTaurides54037.0717.13-8.83CongoAlborz25052.83-11.137.73CongoAlborz26018.3732.21-3.18CongoAlborz27035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlbar26026.6160.32-61.61CongoAfghan26026.6160.32-61.61CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan54027.0952.84-68.62CongoHimalayas280-28.28-136.1159.25CongoHimalayas290-29.12-136.5859.59CongoHimalayas290-29.12-136.5859.59CongoHimalayas290-29.12-136.5859.59CongoHimalayas290-29.12-136.5859.	Arabia	620	10.25	62.24	106.19	Congo
Arabia6509.3763.71124.70CongoArabia75035.38-29.84-99.90CongoTaurides25037.0717.13-8.83CongoTaurides54037.0717.13-8.83CongoAlborz25052.83-11.137.73CongoAlborz26018.3732.21-3.18CongoAlborz27035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlbarz54025.2369.03-55.70CongoAfghan26026.6160.32-61.61CongoAfghan27027.0455.58-66.30CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoHimalayas25027.9145.28-58.22CongoHimalayas280-28.88-136.1159.25CongoHimalayas290-29.12-136.5859.59CongoHimalayas25037.06-162.868.82CongoPontides25037.06-162.868.82CongoLut260-30.15-121.24112.63 </th <th>Arabia</th> <th>630</th> <th>9.43</th> <th>64.66</th> <th>100.28</th> <th>Congo</th>	Arabia	630	9.43	64.66	100.28	Congo
Arabia75035.38-29.84-99.90CongoTaurides25037.0717.13-8.83CongoTaurides54037.0717.13-8.83CongoAlborz25052.83-11.137.73CongoAlborz26018.3732.21-3.18CongoAlborz27035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz54025.2369.03-55.70CongoAfghan26026.6160.32-61.61CongoAfghan26027.0455.58-66.30CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.9145.28-58.22CongoHimalayas27028.2444.80-58.56CongoHimalayas290-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoPontides250-37.06-162.868.82CongoHimalayas290-29.12-136.5859.59CongoHimalayas240-37.06-162.868.82CongoJunalayas540-37.06-162.868.82CongoJunalayas540-37.06-162.	Arabia	650	9.37	63.71	124.70	Congo
Taurides25037.0717.13-8.83CongoTaurides54037.0717.13-8.83CongoAlborz25052.83-11.137.73CongoAlborz26018.3732.21-3.18CongoAlborz27035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAfghan25025.2369.03-55.70CongoAfghan26026.6160.32-61.61CongoAfghan27027.0455.58-66.30CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan25027.9145.28-58.22CongoHimalayas25027.9145.28-58.56CongoHimalayas290-29.12-136.5859.59CongoHimalayas290-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas250-37.06-162.868.82CongoHimalayas260-37.06-162.868.82CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-29.12-136.	Arabia	750	35.38	-29.84	-99.90	Congo
Taurides54037.0717.13-8.83CongoAlborz25052.83-11.137.73CongoAlborz26018.3732.21-3.18CongoAlborz27035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAfghan25025.2369.03-55.70CongoAfghan26026.6160.32-61.61CongoAfghan27027.0455.58-66.30CongoAfghan28027.0853.73-67.75CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.9145.28-58.22CongoHimalayas25027.9145.28-58.22CongoHimalayas290-29.12-136.5859.59CongoHimalayas290-29.12-136.5859.59CongoHimalayas250-37.06-162.868.82CongoPontides540-37.06-162.868.82CongoLut260-30.15-121.24112.63CongoLut260-30.15-121.24116.35Congo	Taurides	250	37.07	17.13	-8.83	Congo
Alborz25052.83-11.137.73CongoAlborz26018.3732.21-3.18CongoAlborz27035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAfghan25025.2369.03-55.70CongoAfghan26026.6160.32-61.61CongoAfghan27027.0455.58-66.30CongoAfghan28027.0853.73-67.75CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan54027.0952.84-68.62CongoHimalayas25027.9145.28-58.22CongoHimalayas290-29.12-136.5859.59CongoHimalayas290-29.12-136.5859.59CongoHimalayas540-37.06-162.868.82CongoHimalayas540-37.06-162.868.82CongoHimalayas540-37.06-162.868.82CongoHimalayas540-37.06-162.868.82CongoLut25031.56-105.38 <t< th=""><th>Taurides</th><th>540</th><th>37.07</th><th>17.13</th><th>-8.83</th><th>Congo</th></t<>	Taurides	540	37.07	17.13	-8.83	Congo
Alborz26018.3732.21-3.18CongoAlborz27035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAfghan25025.2369.03-55.70CongoAfghan26026.6160.32-61.61CongoAfghan27027.0455.58-66.30CongoAfghan28027.0853.73-67.75CongoAfghan29027.0952.84-68.62CongoAfghan29027.0952.84-68.62CongoAfghan25025.7145.28-58.22CongoAfghan29027.0952.84-68.62CongoHimalayas25027.9145.28-58.22CongoHimalayas25027.9145.28-58.22CongoHimalayas290-29.12-136.5859.59CongoHimalayas290-29.12-136.5859.59CongoHimalayas540-37.06-162.868.82CongoHimalayas540-37.06-162.868.82CongoLut25031.56-105.38-105.38CongoLut260-30.15-121.24112.63CongoLut260-30.15-121.24116.35Congo	Alborz	250	52.83	-11.13	7.73	Congo
Alborz27035.3417.69-7.89CongoAlborz54035.3417.69-7.89CongoAfghan25025.2369.03-55.70CongoAfghan26026.6160.32-61.61CongoAfghan27027.0455.58-66.30CongoAfghan28027.0853.73-67.75CongoAfghan29027.0952.84-68.62CongoAfghan54027.0952.84-68.62CongoHimalayas25027.9145.28-58.22CongoHimalayas27028.2444.80-58.56CongoHimalayas280-28.88-136.1159.25CongoHimalayas290-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoLut25031.56-105.38-105.38CongoLut260-30.15-121.24112.63CongoLut260-30.15-121.24112.63CongoLut260-30.15-121.24116.35Congo	Alborz	260	18.37	32.21	-3.18	Congo
Alborz54035.3417.69-7.89CongoAfghan25025.2369.03-55.70CongoAfghan26026.6160.32-61.61CongoAfghan27027.0455.58-66.30CongoAfghan28027.0853.73-67.75CongoAfghan29027.0952.84-68.62CongoAfghan54027.0952.84-68.62CongoAfghan54027.0952.84-68.62CongoHimalayas25027.9145.28-58.22CongoHimalayas27028.2444.80-58.56CongoHimalayas280-28.88-136.1159.25CongoHimalayas290-29.12-136.5859.59CongoPontides250-37.06-162.868.82CongoLut25031.56-105.38-105.38CongoLut260-30.15-121.24112.63CongoLut260-30.15-121.24116.35Congo	Alborz	270	35.34	17.69	-7.89	Congo
Afghan25025.2369.03-55.70CongoAfghan26026.6160.32-61.61CongoAfghan27027.0455.58-66.30CongoAfghan28027.0853.73-67.75CongoAfghan29027.0952.84-68.62CongoAfghan54027.0952.84-68.62CongoAfghan54027.0952.84-68.62CongoHimalayas25027.9145.28-58.22CongoHimalayas27028.2444.80-58.56CongoHimalayas280-28.88-136.1159.25CongoHimalayas290-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-37.06-162.868.82CongoLut25031.56-105.38-105.38CongoLut260-30.15-121.24112.63CongoLut260-30.15-121.24116.35Congo	Alborz	540	35.34	17.69	-7.89	Congo
Afghan26026.6160.32-61.61CongoAfghan27027.0455.58-66.30CongoAfghan28027.0853.73-67.75CongoAfghan29027.0952.84-68.62CongoAfghan54027.0952.84-68.62CongoHimalayas25027.9145.28-58.22CongoHimalayas27028.2444.80-58.56CongoHimalayas290-29.12-136.5859.59CongoHimalayas290-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoLut25031.56-105.38-105.38CongoLut260-30.15-121.24112.63CongoLut260-30.15-121.24116.35Congo	Afghan	250	25.23	69.03	-55.70	Congo
Afghan27027.0455.58-66.30CongoAfghan28027.0853.73-67.75CongoAfghan29027.0952.84-68.62CongoAfghan54027.0952.84-68.62CongoHimalayas25027.9145.28-58.22CongoHimalayas27028.2444.80-58.56CongoHimalayas280-28.88-136.1159.25CongoHimalayas290-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-37.06-162.868.82CongoLut25031.56-105.38-105.38CongoLut260-30.15-121.24112.63CongoLut270-29.41-123.44116.35Congo	Afghan	260	26.61	60.32	-61.61	Congo
Afghan28027.0853.73-67.75CongoAfghan29027.0952.84-68.62CongoAfghan54027.0952.84-68.62CongoHimalayas25027.9145.28-58.22CongoHimalayas27028.2444.80-58.56CongoHimalayas280-28.88-136.1159.25CongoHimalayas290-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoPontides250-37.06-162.868.82CongoLut25031.56-105.38-105.38CongoLut260-30.15-121.24112.63CongoLut270-29.41-123.44116.35Congo	Afghan	270	27.04	55.58	-66.30	Congo
Afghan29027.0952.84-68.62CongoAfghan54027.0952.84-68.62CongoHimalayas25027.9145.28-58.22CongoHimalayas27028.2444.80-58.56CongoHimalayas280-28.88-136.1159.25CongoHimalayas290-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoPontides250-37.06-162.868.82CongoLut25031.56-105.38-105.38CongoLut260-30.15-121.24112.63Congo	Afghan	280	27.08	53.73	-67.75	Congo
Afghan54027.0952.84-68.62CongoHimalayas25027.9145.28-58.22CongoHimalayas27028.2444.80-58.56CongoHimalayas280-28.88-136.1159.25CongoHimalayas290-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoPontides250-37.06-162.868.82CongoLut25031.56-105.38-105.38CongoLut260-30.15-121.24112.63CongoLut270-29.41-123.44116.35Congo	Afghan	290	27.09	52.84	-68.62	Congo
Himalayas25027.9145.28-58.22CongoHimalayas27028.2444.80-58.56CongoHimalayas280-28.88-136.1159.25CongoHimalayas290-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoPontides250-37.06-162.868.82CongoLut25031.56-105.38-105.38CongoLut260-30.15-121.24112.63Congo	Afghan	540	27.09	52.84	-68.62	Congo
Himalayas27028.2444.80-58.56CongoHimalayas280-28.88-136.1159.25CongoHimalayas290-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoPontides250-37.06-162.868.82CongoLut25031.56-105.38-105.38CongoLut260-30.15-121.24112.63Congo	Himalavas	250	27.91	45.28	-58.22	Congo
Himalayas280-28.88-136.1159.25CongoHimalayas290-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoPontides250-37.06-162.868.82CongoPontides540-37.06-162.868.82CongoLut25031.56-105.38-105.38CongoLut260-30.15-121.24112.63CongoLut270-29.41-123.44116.35Congo	Himalayas	270	28.24	44.80	-58.56	Congo
Himalayas290-29.12-136.5859.59CongoHimalayas540-29.12-136.5859.59CongoPontides250-37.06-162.868.82CongoPontides540-37.06-162.868.82CongoLut25031.56-105.38-105.38CongoLut260-30.15-121.24112.63CongoLut270-29.41-123.44116.35Congo	Himalayas	280	-28.88	-136.11	59.25	Congo
Himalayas540-29.12-136.5859.59CongoPontides250-37.06-162.868.82CongoPontides540-37.06-162.868.82CongoLut25031.56-105.38-105.38CongoLut260-30.15-121.24112.63CongoLut270-29.41-123.44116.35Congo	Himalayas	290	-29.12	-136.58	59.59	Congo
Pontides 250 -37.06 -162.86 8.82 Congo Pontides 540 -37.06 -162.86 8.82 Congo Lut 250 31.56 -105.38 -105.38 Congo Lut 260 -30.15 -121.24 112.63 Congo Lut 270 -29.41 -123.44 116.35 Congo	Himalayas	540	-29.12	-136.58	59.59	Congo
Pontides 540 -37.06 -162.86 8.82 Congo Lut 250 31.56 -105.38 -105.38 Congo Lut 260 -30.15 -121.24 112.63 Congo Lut 270 -29.41 -123.44 116.35 Congo	Pontides	250	-37.06	-162.86	8.82	Congo
Lut 250 31.56 -105.38 -105.38 Congo Lut 260 -30.15 -121.24 112.63 Congo Lut 270 -29.41 -123.44 116.35 Congo	Pontides	540	-37.06	-162.86	8.82	Congo
Lut 260 -30.15 -121.24 112.63 Congo Lut 270 -29.41 -123.44 116.35 Congo	Lut	250	31.56	-105.38	-105.38	Congo
Lut 270 -29.41 -123.44 116.35 Congo	Lut	260	-30.15	-121.24	112.63	Congo
	Lut	270	-29.41	-123.44	116.35	Congo

Moving plate	Age (Ma)	Latitude	Longitude	Angle	Fixed plate
Lut	540	-29.41	-123.44	116.35	Congo
Sibumasu	250	14.56	110.76	48.20	Congo
Sibumasu	260	5.02	105.57	68.20	Congo
Sibumasu	270	-1.95	-76.03	-79.45	Congo
Sibumasu	540	-1.95	-76.03	-79.45	Congo
North China	250	23.87	130.20	-71.99	Congo
North China	260	31.33	124.88	-76.30	Congo
North China	270	37.09	120.13	-87.45	Congo
North China	280	44.71	117.46	-98.11	Congo
North China	290	43.96	112.10	-103.25	Congo
North China	300	43.20	109.46	-106.39	Congo
North China	310	47.21	108.32	-110.85	Congo
North China	320	50.76	111.26	-121.28	Congo
North China	330	52.88	111.21	-122.52	Congo
North China	340	-58.02	-61.02	123.54	Congo
North China	350	57.23	122.22	-121.37	Congo
North China	360	54.77	130.04	-122.23	Congo
North China	370	54.81	149.28	-130.31	Congo
North China	380	50.98	146.36	-130.91	Congo
North China	390	52.34	145.65	-128.95	Congo
North China	400	53.08	139.49	-121.55	Congo
North China	410	55.94	130.09	-118.91	Congo
North China	420	50.14	118.18	-118.36	Congo
North China	430	44.63	110.81	-127.58	Congo
North China	440	45.01	110.02	-144.79	Congo
North China	450	46.54	113.30	-158.29	Congo
North China	460	-49.64	-58.18	161.48	Congo
North China	470	50.05	125.75	-153.16	Congo
North China	480	46.76	120.48	-144.40	Congo
North China	490	36.08	109.76	-148.18	Congo
North China	500	32.54	105.78	-151.47	Congo
North China	510	40.37	104 68	-146.81	Congo
North China	520	50.59	100.50	-153.98	Congo
North China	530	54.17	88.76	-168.61	Congo
North China	540	58.81	81.38	-175.66	Congo
North China	650	-7.58	-81.79	118.20	Congo
North China	750	24.86	87.86	-125.15	Congo
Lhasa	250	25.38	68 23	-30 37	Congo
Lhasa	250	36.14	44 61	-35 20	Congo
Lhasa	200	34 53	37.26	-43 97	Congo
Lhasa	280	30.73	41.85	-54 31	Congo
Lhasa	200	29.19	43.42	-59.60	Congo
Lhasa	540	29.19	43.47	-59.60	Congo
Lhasa	750	_4 17	-85.61	-92.00	Congo
Lilasa Madagasaar	250	- - 1.1/ 14 71	137.67	-15.64	Congo
Madagasoar	230 540	14.74	137.02	-15.04	Congo
Madagascar	540	3/14	137.02	-13.04	Month
Madagassar	J41 550	-34.10	+1.00	201.20	Montha
Madagassar	55U 520	-43./0	44./9	201.29	Iviantie
waagascar	560	-38.12	41.30	188.97	iviantie

Moving plate	Age (Ma)	Latitude	Longitude	Angle	Fixed plate
Madagascar	570	-34.18	50.71	143.45	Mantle
Madagascar	580	-24.07	53.12	130.87	Mantle
Madagascar	590	-12.87	55.41	122.46	Mantle
Madagascar	591	-49.79	88.69	-68.37	Mantle
Madagascar	600	-0.34	57.81	120.14	Mantle
Madagascar	610	14.31	61.48	120.62	Mantle
Madagascar	620	27.25	61.89	127.78	Mantle
Madagascar	621	35.43	-65.68	117.76	Mantle
Madagascar	630	42.21	61.99	139.77	Mantle
Madagascar	631	26.41	-52.94	127.39	Mantle
Madagascar	640	50.30	70.62	163.53	Mantle
Madagascar	650	54.34	80.73	188.77	Mantle
Madagascar	660	-54.96	-96.88	-192.12	Mantle
Madagascar	670	58.24	74.81	195.96	Mantle
Madagascar	680	56.21	83.63	199.90	Mantle
Madagascar	681	15.96	-28.81	179.30	Mantle
Madagascar	690	-55.63	-91.42	154.57	Mantle
Madagascar	700	57.32	79.84	-151.39	Mantle
Madagascar	701	14.74	-24.09	-176.11	Mantle
Madagascar	710	55.91	86.13	-146.73	Mantle
Madagascar	720	-55.33	-89.82	142.98	Mantle
Madagascar	722	13.85	-18.53	-174.15	Mantle
Madagascar	730	-54.44	-86.03	-221.76	Mantle
Madagascar	740	-56.22	-92.28	-225.19	Mantle
Madagascar	750	53.74	95.13	230.58	Mantle
Somalia	250	9.89	143.00	-0.22	Congo
Somalia	540	9.89	142.99	0.22	Congo
Somalia	550	-9.33	60.90	23.30	Congo
Somalia	600	-80.98	-67.41	-19.76	Congo
Somalia	650	-63.56	29.46	-48.70	Congo
Somalia	740	-25.55	58.75	-33.79	Congo
Somalia	750	-21.29	60.78	-34.35	Congo
Lake Victoria Block	250	9.89	143.00	-0.22	Congo
Lake Victoria Block	540	9.89	142.99	0.22	Congo
NW Africa	250	33.65	26.02	2.34	Congo
NW Africa	540	33.65	26.02	2.34	Congo
NW Africa	550	28.48	0.95	-19.50	Congo
NW Africa	650	-36.59	-7.31	100.67	Congo
NW Africa	750	-35.33	-16.21	156.75	Congo
NE Africa	250	40.45	-61.40	-0.70	Congo
NE Africa	540	40.41	-61.40	-0.70	Congo
NE Africa	550	5.36	-139.78	9.10	Congo
NE Africa	650	-0.62	-156.52	65.66	Congo
NE Africa	750	-3.37	-155.50	55.27	Congo
Australia	250	19.57	117.90	-56.42	Congo
Australia	540	19.57	117.90	-56.42	Congo
Australia	550	16.33	121.83	-54.50	Congo
Australia	650	-29.67	106.96	-76.75	Congo
Australia	750	10.10	104.46	-62.70	Congo
					0

Moving plate	Age (Ma)	Latitude	Longitude	Angle	Fixed plate
Antarctica	250	10.44	148.74	-58.41	Congo
Antarctica	540	10.44	148.74	-58.41	Congo
Antarctica	550	11.16	158.76	-58.76	Congo
Antarctica	650	-21.63	152.90	-42.17	Congo
Antarctica	750	12.13	142.01	-51.91	Congo
African craton	250	-2.78	150.08	45.34	Mantle
African craton	260	-3.81	148.98	45.87	Mantle
African craton	270	7.40	-35.85	-51.90	Mantle
African craton	280	-7.21	143.43	58.41	Mantle
African craton	290	-14.93	137.70	60.97	Mantle
African craton	300	20.57	-47.73	-64.52	Mantle
African craton	310	19.53	-53.66	-67.86	Mantle
African craton	320	11.50	-60.73	-74.46	Mantle
African craton	330	7.86	-66.19	-73.18	Mantle
African craton	340	13.52	-74.58	-76.26	Mantle
African craton	350	1.55	-69.58	-72.48	Mantle
African craton	360	8.81	115.68	77.64	Mantle
African craton	370	18.89	126.95	89.51	Mantle
African craton	380	15.33	120.65	88.91	Mantle
African craton	390	-15.29	-62.94	-81.53	Mantle
African craton	400	-14.06	-67.19	-72.25	Mantle
African craton	410	-9.35	-75.71	-64.60	Mantle
African craton	420	5.99	-92.66	-67.78	Mantle
African craton	430	14.65	-107.00	-82.48	Mantle
African craton	440	9.11	-108.91	-96.15	Mantle
African craton	450	6.94	-110.79	-111.15	Mantle
African craton	460	11.07	-118.20	-124.37	Mantle
African craton	470	12.77	-115.52	-129.29	Mantle
African craton	480	18.30	-120.15	-135.48	Mantle
African craton	490	23.06	-130.91	-148.84	Mantle
African craton	500	26.66	-140.67	-154.56	Mantle
African craton	510	31.24	-144.41	-146.64	Mantle
African craton	520	30.22	-139.72	-127.55	Mantle
African craton	530	25.12	-136.31	-109.38	Mantle
African craton	540	20.18	-135.59	-94.43	Mantle
African craton	550	40.26	-124.87	-160.88	Mantle
African craton	551	35.94	-4.16	-79.25	Laurentia
African craton	560	39.78	-17.47	-65.49	Laurentia
African craton	650	-5.21	-37.25	-164.97	Laurentia
African craton	750	-19.26	-55.95	-191.95	Laurentia
Ganderia	300	32.58	64.04	57.67	Mantle
Ganderia	320	23.26	58.54	74.37	Mantle
Ganderia	340	-4.04	30.42	73.36	Mantle
Ganderia	360	-8.83	23.43	81.51	Mantle
Ganderia	380	-12.63	11.21	88.92	Mantle
Ganderia	400	-15.02	3.41	94.22	Mantle
Ganderia	420	-32.17	6.68	92.70	Mantle
Ganderia	425	-36 57	9.86	94 74	Mantle
Ganderia	430	-40 58	12.07	97.12	Mantle
Gallucia	750	10.20	12.07	11.14	manne

Moving plate	Age (Ma)	Latitude	Longitude	Angle	Fixed plate
Ganderia	440	-38.17	12.07	101.20	Mantle
Ganderia	460	-32.45	15.13	122.20	Mantle
Ganderia	470	-27.48	16.49	123.85	Mantle
Ganderia	480	-26.98	18.25	135.29	Mantle
Ganderia	490	-29.87	2.99	143.82	Mantle
Ganderia	510	-31.24	-2.71	168.78	Mantle
Ganderia	520	-27.34	-1.11	155.96	Mantle
Ganderia	530	-22.26	-0.09	146.82	Mantle
Ganderia	540	-16.36	-0.67	142.34	Mantle
Ganderia	550	-56.32	-22.33	167.93	Mantle
Ganderia	560	-50.07	-10.88	173.61	Mantle
Ganderia	570	-46.96	0.23	169.98	Mantle
Ganderia	580	-46.11	1.81	161.98	Mantle
Ganderia	590	-45.62	2.90	158.38	Mantle
Ganderia	600	-41.16	13.91	154.52	Mantle
Ganderia	610	-39.43	16.38	149.30	Mantle
Ganderia	620	-38.09	18.46	143.31	Mantle
Ganderia	650	-26.39	25.07	109.33	Mantle
Ganderia	690	-27.73	4.90	84.26	Mantle
Ganderia	700	-26.38	22.21	84.65	Mantle
Ganderia	710	-25.74	21.48	79.89	Mantle
Ganderia	720	-23.05	39.04	82.99	Mantle
Ganderia	730	-19.44	53.00	86.10	Mantle
Ganderia	740	-16.58	64 84	92.13	Mantle
Ganderia	750	-16.34	75.00	99 74	Mantle
Dampoan	250	47.56	31.66	57.74	Congo
T ampean Dompoon	300	47.50	-31.00	55.04	Congo
T ampean Dompoon	540	47.46	-31.29	56.60	Congo
r ampean Dompoon	550	47.40	-32.29	58.46	Congo
r ampean Dompoon	560	40.40 52.56	-20.43	52.02	Congo
r ampean Dompoon	570	20.19	-30.34	57.11	Congo
Pampean	570	29.10	-33.23	102.00	Congo
Pampean	610	12.23	144.01	-102.90	Congo
Pampean	650	1 / .45	26.04	-121.70	Congo
Pampean	650	-10.45	-30.04	140.40	Congo
Pampean	700	-19.75	-33.40	145.80	Congo
Pampean	700	-16.23	-32.99	130.90	Congo
Pampean	/10	-25.02	-32.27	148.50	Congo
Pampean	720	-20.75	-31.32	155.48	Congo
Pampean	750	-22.10	-30.39	130.09	Congo
Rio de la Plata	250	47.58	-33.30	56.20	Congo
Rio de la Plata	540	4/./1	-31.93	56.46	Congo
Kio de la Plata	550	46.67	-29.38	58.63	Congo
Kio de la Plata	560	32.33	-32.38	63.29	Congo
Rio de la Plata	570	24.28	-35.32	69.35	Congo
Rio de la Plata	580	15.39	-36.14	78.58	Congo
Rio de la Plata	590	10.16	-37.21	86.66	Congo
Rio de la Plata	600	5.58	-38.14	95.00	Congo
Rio de la Plata	650	-12.87	-39.32	163.85	Congo
Rio de la Plata	660	-14.47	-38.84	154.84	Congo

Moving plate	Age (Ma)	Latitude	Longitude	Angle	Fixed plate
Rio de la Plata	740	-18.95	-29.46	154.64	Congo
Rio de la Plata	750	-18.84	-28.68	155.64	Congo
Tarim	300	48.36	-6.37	65.32	Mantle
Tarim	310	47.98	-6.40	70.96	Mantle
Tarim	320	44.52	-16.69	66.10	Mantle
Tarim	330	27.49	-24.03	54.31	Mantle
Tarim	340	33.92	-9.88	61.50	Mantle
Tarim	350	23.40	-8.43	59.56	Mantle
Tarim	360	15.19	-7.59	66.33	Mantle
Tarim	370	5.96	-7.58	94.36	Mantle
Tarim	380	10.83	-9.14	89.29	Mantle
Tarim	390	7.61	-8.15	77.45	Mantle
Tarim	400	1.23	-8.14	70.85	Mantle
Tarim	410	7.13	-10.27	56.68	Mantle
Tarim	420	20.70	-27.03	38.47	Mantle
Tarim	430	37.96	-53.98	30.65	Mantle
Tarim	440	3.96	-68.42	37.50	Mantle
Tarim	450	-5.76	-63.96	50.32	Mantle
Tarim	460	-73.26	-85.55	-87.46	Mantle
Tarim	470	72.25	88.63	97.11	Mantle
Tarim	480	-66.99	-82.82	-97.83	Mantle
Tarim	490	-53.02	-60.08	-84.57	Mantle
Tarim	500	-44.47	-61.47	-91.76	Mantle
Tarim	510	-47.98	-75.01	-90.30	Mantle
Tarim	520	52.62	74.47	73.39	Mantle
Tarim	530	45.24	60.66	75.17	Mantle
Tarim	540	39.57	46.52	72.56	Mantle
Tarim	650	-6.85	-13.20	80.50	Mantle
Tarim	750	21.02	-175.31	-70.70	Mantle
SE New England	490	-26.77	5.17	159.94	Mantle
SE New England	480	-27.60	5.51	143.80	Mantle
SE New England	460	-37.31	-4.44	128.47	Mantle
SE New England	440	-34.83	8.77	95.80	Mantle
SE New England	420	-31.92	2.92	92.38	Mantle
SE New England	400	-14.87	5.40	94.28	Mantle
SE New England	380	-12.63	11.32	88.97	Mantle
SE New England	360	-8.78	22.65	81.34	Mantle
SE New England	340	-3.87	28.95	73.60	Mantle
SE New England	320	23.23	58.64	74.39	Mantle
SE New England	300	32.48	64.07	57.63	Mantle
SE New England	430	-40.46	12.37	97.36	Mantle
SE New England	450	-31.88	12.11	108.90	Mantle
SE New England	470	-28.48	-5.14	128.30	Mantle
SE New England	490	52.69	-173.27	-214.34	Mantle
SE New England	491	27.43	-173.99	-161.88	Mantle
SE New England	500	-28.55	-1.12	174.60	Mantle
SE New England	510	29.87	173.35	-171.96	Mantle
SE New England	520	28.49	170.39	-156.32	Mantle
SE New England	530	-21.23	-3.24	148.41	Mantle
8	-				

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Moving plate	Age (Ma)	Latitude	Longitude	Angle	Fixed plate
SE New England	540	-17.36	-7.39	145.46	Mantle
SE New England	550	-54.01	-19.38	166.33	Mantle
SE New England	560	-51.07	-23.33	174.41	Mantle
SE New England	570	-51.42	-21.59	166.86	Mantle
SE New England	580	-49.96	-15.32	160.49	Mantle
SE New England	590	-47.22	-8.17	154.53	Mantle
SE New England	600	-44.84	-4.97	147.94	Mantle
SE New England	610	-43.46	-1.38	141.71	Mantle
SE New England	620	-39.37	8.16	140.41	Mantle
SE New England	630	-34.88	15.42	132.60	Mantle
SE New England	640	-29.29	25.00	128.69	Mantle
SE New England	650	-27.29	23.33	120.22	Mantle
SE New England	660	-25.75	28.75	117.47	Mantle
SE New England	670	-24.37	33.62	114.91	Mantle
SE New England	680	-22.17	38.89	111.18	Mantle
SE New England	690	-22.07	40.24	106.75	Mantle
SE New England	700	-20.38	43.77	102.93	Mantle
SE New England	710	-18.63	48.22	100.13	Mantle
SE New England	720	-16.47	54.01	99.50	Mantle
SE New England	730	-14.33	56.59	94.64	Mantle
SE New England	740	-11.95	60.88	92.41	Mantle
SE New England	750	-9.09	63.22	88.36	Mantle

APPENDIX E

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Angeliki Papoutsa Department of Earth Sciences Dalhousie University

Dear Angeliki,

Halifax, Nova Scotia Canada

11 August 2015

Permission is granted for your use of the following paper, published in the December 2014 issue of The Canadian Mineralogist, in your thesis at Dalhousie University:

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Proper citation of where the paper was originally published is all that is required.

Best regards,

Mackenzie Parker

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