

GEOCHEMISTRY OF IGNEOUS ROCKS ASSOCIATED WITH THE MMH PORPHYRY
COPPER DEPOSIT, CHUQUICAMATA DISTRICT, CHILE

Jessica Wilson

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Department of Earth Sciences
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Department of Earth Sciences

Halifax, Nova Scotia

Canada B3H 4J1

(902) 494-2358

FAX (902) 494-6889

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AUTHOR: Jessica Wilson

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Abstract

The MMH deposit is located 7 km to the south of the Chuquicamata mine within the Chuquicamata District in northern Chile, in the Domeyko Cordillera. The two deposits are located on opposite sides of the West Fault, a section of a regional (Domeyko) orogen-parallel fault system with ca. 35 km of strike-slip displacement. The MMH has a tabular shape, parallel to and adjoining the West Fault. MMH, held and operated by the state-owned company CODELCO, contains resources of more than one billion tonnes of copper ore, and is being prepared for open pit mining.

The focus of this thesis is the igneous geochemistry of the MMH deposit. The ore at MMH is hosted by the MM Granodiorite (Triassic; U-Pb in zircon 237-222 Ma), which is intruded by the MM Porphyry (Eocene; U-Pb in zircon 39 Ma) and the MM Quartz Porphyry (Eocene; U-Pb in zircon 36 Ma). The deposit contains Cu-(Mo) porphyry type mineralization (Re-Os in molybdenite ≤ 37.3 Ma) and potassic alteration at depth, overprinted at higher levels by a younger (32-31 Ma), high sulphidation Cu-(Ag-As) ore characterized by high-grade hydrothermal breccia bodies. The genetic relationship between the Cu-(Mo) and Cu-(Ag-As) mineralization and Tertiary porphyries is still to be proven.

Trace element geochemistry reveals that the Triassic MM Granodiorite and the Eocene MM Porphyry have nearly identical trace-element geochemical signatures. Both are the equivalent of volcanic or sub-volcanic andesite, whereas the Eocene MM Quartz Porphyry is closer to a rhyodacite in composition. All have calc-alkaline - to some extent adakitic - characteristics. Electron microprobe analyses show that whereas the plagioclase in MM Granodiorite has been pervasively albitized, the plagioclase in MM Porphyry is oligoclase-andesine.

The surprising geochemical similarity between Triassic and Eocene host rocks suggests that tectonic conditions and magma generation in the Triassic was not very different to the Tertiary for this region of northern Chile. These results are in conflict with the commonly held thinking that the Triassic was a time of tectonic extension and relaxation. One implication is that the Triassic and Paleozoic basement in the Domeyko Cordillera may be more prospective for porphyry copper type deposits than previously thought. Another implication is that the composition of the mantle and lower crust in the Domeyko Cordillera may have more to do with the formation of giant porphyry type copper-molybdenum deposits than previously envisaged.

The MMH igneous host rocks have geochemical similarities with the Fortuna Igneous Complex, which is juxtaposed to the Chuquicamata deposit west of the West Fault, and with the El Abra Porphyry Complex, which is 35 km north, on the east side of the regional fault.

TABLE OF CONTENTS

Abstract	i
Table of Contents	ii
Table of Figures	v
Table of Tables	vii
Acknowledgements	viii
CHAPTER 1: INTRODUCTION	
1.1 General Statement	1
1.2 Overview of Cu Porphyry Systems	3
1.3 Chuquicamata District	4
1.4 Previous Work	5
1.5 Methods	7
CHAPTER 2: IGNEOUS ROCKS OF PORPHYRY COPPER DEPOSITS	
2.1 Igneous Petrology	9
2.1.1 Magmatic Processes	10
2.1.2 Hydrothermal Alteration Processes	11
2.1.3 Alteration Zones	13
2.1.4 Veining	15
2.1.5 Second Boiling	16
2.1.6 Supergene Enrichment	18
2.2 Tectonic Setting	19
2.2.1 Origin of Cu	20
2.2.2 Structural Control of Porphyry Deposits	20
2.2.3 Adakites	21

2.2.1 Origin of Cu	20
2.2.2 Structural Control of Porphyry Deposits	20
2.2.3 Adakites	21
CHAPTER 3: GEOLOGY OF THE CHUQUICAMATA DISTRICT	
3.1 Geology of the Chuquicamata Open Pit Mine	22
3.1.1 Chuquicamata Porphyry Complex	22
3.1.2 Fortuna Complex	25
3.1.3 Pre-Chuquicamata Granodiorites	26
3.1.4 Hydrothermal Alteration of the Chuquicamata Porphyry Complex	26
3.1.5 Supergene Enrichment	27
3.2 Geology of the MMH Deposit	27
3.2.1 Eastern Portion of the MMH Mine	28
3.2.2 MM Granodiorite	28
3.2.3 MM Porphyry	29
3.2.4 MM Quartz Porphyry	29
3.2.5 Other Geologic Units	30
3.2.6 Hypogene Mineralization	30
3.2.7 Ages of Hypogene Phases	31
3.2.8 Supergene Mineralization	31
3.3 The West Fault	32
3.3.1 Horizontal Displacement of the West Fault	32
3.3.2 Vertical Displacement of the West Fault System	33
CHAPTER 4: PETROGRAPHY OF THE MMH DEPOSIT	
4.1 Optical Microscopy	35

4.1.1 MM Porphyry	36
4.1.2 MM Granodiorite	40
4.1.3 MM Quartz Porphyry	42
4.2 Electron Microprobe Analysis	44
4.2.1 MM Porphyry	46
4.2.2 MM Granodiorite	47
4.2.3 MM Quartz Porphyry	48
4.2.4 Sulphides	48
CHAPTER 5: GEOCHEMISTRY	
5.1 Analytical Methods	51
5.2 Major Elements	54
5.3 Major Element Geochemistry	57
5.4 Comparisons with other Intrusive Complexes in the Region	64
5.5 Discussion	67
CHAPTER 6: DISCUSSION AND CONCLUSIONS	70
CHAPTER 7: FUTURE WORK	77
REFERENCES	78
APPENDIX 1	86
APPENDIX 2	129
APPENDIX 3	151
APPENDIX 4	154
APPENDIX 5	162
APPENDIX 6	174

TABLE OF FIGURES

Figure 1.1 Chuquicamata District	2
Figure 1.2 Major Porphyry Deposits of northern Chile	2
Figure 1.3 Regional Map of Chuquicamata District	8
Figure 2.1 Parental Batholith	9
Figure 2.2 Model of Porphyry Deposits	13
Figure 2.3 Second Boiling	17
Figure 2.4 Oceanic-Continental Subduction Environment	19
Figure 3.1 Lithologic Units within the Chuquicamata Open Pit	23
Figure 3.2 Satellite Image of the MMH Deposit	28
Figure 4.1 Core Chip Photo of the MM Porphyry	38
Figure 4.2 Optical Microscopy Photo of the MM Porphyry	38
Figure 4.3 Core Chip Photo of the MM Granodiorite	41
Figure 4.4 Optical Microscopy Photo of the MM Granodiorite	41
Figure 4.5 Core Chip Photo of the MM Quartz Porphyry	43
Figure 4.6 Optical Microscopy Photo of the MM Quartz Porphyry	43
Figure 4.7 Feldspar Ternary Diagram with MMH Data	45
Figure 4.8 Sulphide Ternary Diagram with MMH Data	49
Figure 5.1 FUS-MS vs. PP-XRF Analytical Results	52
Figure 5.2 Le Maitre (1989) and Rickwood (1989) Classification Diagram	54
Figure 5.3 Major Elements vs. SiO ₂	56
Figure 5.4 Winchester and Floyd (1977) Volcanic Rock Classification Diagram	58
Figure 5.5 Rock/Chondrite Plot	59
Figure 5.6 Baldwin and Pearce Classification Diagram	61
Figure 5.7 Trace Elements vs SiO ₂	63

Figure 5.8 Trace Element Comparison between MMH Lithologic Units	64
Figure 5.9 Trace Element Comparison between MMH and Chuquicamata	65
Figure 5.10 Trace Element Comparison between MMH and El Abra	66
Figure 5.11 Trace Element Comparison between MMH and Fortuna	67

TABLE OF TABLES

Table 1.1 Previous Work	7
Table 3.1 Dates of the Chuquicamata Porphyries	24
Table 4.1 Drillcore Samples from MMH	36
Table 4.2 Proportions of Feldspar End Members in MM Porphyry	46
Table 4.3 Proportions of Feldspar End Member in MM Granodiorite	47
Table 4.4 Proportions of Feldspar End Members in MM Quartz Porphyry	48
Table 4.5 Barium Analysis of Feldspars in MM Quartz Porphyry	48
Table 4.6 Proportions of Sulphides in MMH Lithologic Units	50
Table 5.1 MMH Adakite Characteristics	60

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CHAPTER 1: INTRODUCTION

1.1 General Statement

The Chuquicamata mining district, site of one of the three largest open pit mines in the world, is located northeast of the port city of Antofagasta in northern Chile (Fig. 1.2). Mined continuously since 1915, the Chuquicamata open pit has a length of 5 km, width of 3 km and depth of >1 km, and has copper resources of over 10 billion metric tonnes of ore (e.g. Ossandón et al. 2001). Besides its extraordinary size, what makes this giant copper deposit all the more interesting is that it is truncated on its west side by the important north-south regional structure denominated Falla Oeste or West Fault (Fig. 1.1) with a strike-slip displacement of ~35 km. This circumstance has led to speculation and an intense search for the “missing half” of Chuquicamata across the fault, identifying several mineralized bodies, but so far none that meets the qualifications. One of these mineralized bodies is the MMH (Mina Ministro Hales; originally known as *Mansa Mina* or MM) which is 7 km to the south of Chuquicamata on the opposite side of the West Fault (see Fig. 1.1). The MMH deposit, the subject of this thesis, has been classified as another supergiant Cu-Mo type porphyry deposit within the Chuquicamata District. The deposit has a tabular shape, parallel to and adjoining the West Fault and involves numerous fault slices (Boric et al. 2009). This thesis attempts to shed some light onto the question of the relationship of the MMH deposit with other copper-molybdenum deposits in the Chuquicamata District.

Open pit mining of MMH is scheduled to begin in 2013 (Boric et al. 2009), to be followed by underground mining in the next decade. As part of the preparation for mining, a full review of the geological model of MMH is underway, and one of the aspects being considered is the better definition and characterization of the host rocks.

This thesis provides the first systematic, internally-consistent geochemical analysis of characteristic rock units as shown in the geological maps and sections.

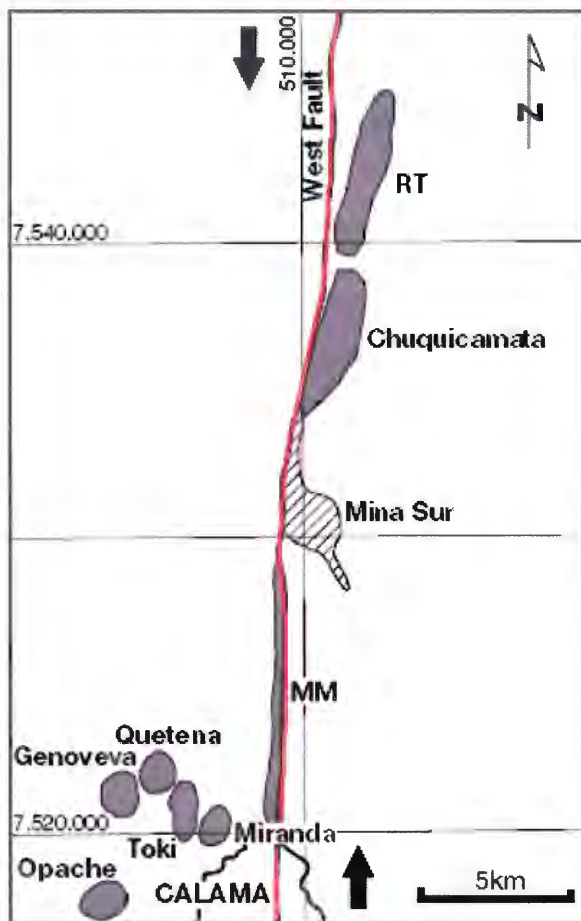


Figure 1.1 Areal map of the Chuquicamata District, northern Chile. This map shows the spatial relationship between the MM (MMH) ore deposit and Chuquicamata ore deposit. The distance between the two is approx 7 km. This figure also indicates the location of RT or Radomiro Tomic and Mina Sur, both of which are mines with mineralization related to the Chuquicamata Intrusive Complex. The Opache, Genoveva, Toki and Quetena deposits are satellite porphyry deposits that are currently being explored. (modified from Sillitoe 2010).



Figure 1.2 map showing the arc-parallel distribution of Cu porphyries from central to northern Chile (Sillitoe 2010).

1.2 Overview of Cu Porphyry Systems

The conceptual models of copper porphyry systems have evolved over the last forty years and the consensus today is that the majority of the most significant deposits formed at the roots of subduction-related stratovolcanoes (Sillitoe 2010, Guilbert & Park 1986, Sillitoe 1973). They are comprised of hydrothermally and tectonically altered sub-volcanic rock centered around stock systems. These stock “porphyry” systems are emplaced 1.5-4 km beneath the surface from subaerial volcanic eruptions (Sillitoe, 1973). The stock systems are the upper extensions of plutonic intrusive bodies at depth. These plutonic bodies are result of calc-alkaline arc magmatism above active continental subduction zones. On the large scale, this activity results in a high density of linearly disposed arc parallel deposits such as in northern Chile (Fig. 1.2). On the small scale, this magmatism results in multiple porphyry and stock systems originating from a single massive batholith at depth (Sillitoe 2010). Not only are these porphyry systems spatially related, they evolve within a time span of roughly 10-20 million years also known as “metallogenic epochs” (Sillitoe 2010). The tight-knit temporal and spatial relationships of porphyry deposits proves to be advantageous in terms of exploration methods and often results in districts of Cu mines such as the Chuquicamata Mining District.

For an up-to-date description and discussion of porphyry copper type deposits, the reader is referred to a comprehensive review by Sillitoe (2010).

1.3 Chuquicamata District

The Chuquicamata Mining District lies 2.8 km above sea level in the Atacama Desert of northern Chile (22° 17.5' S, 68 ° 54.5' W). The copper oxide outcrops were first exploited by the Incas in pre-colonial times and subsequently by the Spanish conquistadors (Ossandón et al. 2001, Guilbert & Park 1986). Modern operations began in 1879 with Chilean and British companies. In 1915 the Chuquicamata open-pit mine was established in order to extract disseminated copper oxide ore. The South Mine (Mina Sur) (see Fig. 1.1) which is comprised of *exotic* (oxides laterally displaced from their original source) copper ore related to the Chuquicamata porphyry system was discovered in 1957. The Radomiro Tomic (RT) mine (see Fig. 1.1), which is north of Chuquicamata but part of the same porphyry system, was initiated in the 1960s. The MMH deposit was discovered in 1967; its geology is discussed in more detail below in section 1.4. To date, the region has produced 2.155 billion metric tons of ore averaging 1.52% Cu, equating to 87.02 Million tonnes (Mt) of Cu metal. Projected resources for the district predict another 11.45 billion tonnes at 0.76% Cu still in the ground (Ossandón et al 2001). As result of the phenomenally large concentration of Cu ore, Chuquicamata ranks in the top three current copper producing districts in the world, and was for the majority of the 20th century, the largest copper mine in the world. The Chuquicamata District is owned and operated by the Corporación del Cobre de Chile or CODELCO which is a state-owned mining company.

The Chuquicamata Intrusive Complex contains the vast majority of ore mineralization. It occurs in contact with the eastern side of the West Fault and extends 9 km from its contact with the West Fault into the Radomiro Tomic Mine to the north. The

Chuquicamata Intrusive Complex is composed of three porphyry systems, East Porphyry, West Porphyry and Banco Porphyry, which are temporally and texturally separate but compositionally similar (Ossandón et al. 2001, McInnes et al. 1999). The East porphyry is the earliest intrusion and the largest in volume and ore mineralization (Ossandón et al. 2001, McInnes et al. 1999). McInnes et al. (1999) described all three units as monzogranite porphyries. The Chuquicamata Intrusive Complex ore body terminates against the West Fault, juxtaposing the unit against the economically barren Fortuna Intrusive Complex.

The West Fault is part of the intricate and extensive Domeyko Fault system. It truncates many of the porphyry deposits in the region including Chuquicamata. The truncation of these major Cu porphyry deposits indicates that there may be “missing portions” somewhere along the adjacent side of the West Fault. Although it is generally agreed upon that the movement since the time of emplacement has been between 35 and 37 km sinistral, there have been reversals in movement throughout its existence (e.g. Tomlinson & Blanco 1997). There is also evidence of vertical displacement within the system (e.g. McInnes et al. 1999). The post mineralization displacement along the West Fault is significant in that it has determined where the “missing half” of Chuquicamata has gone and whether or not the potential ore exists and is extractable.

1.4 Previous Work

Due to its economic importance, the Chuquicamata district has been extensively studied. The most recent and comprehensive review of the Chuquicamata Mining District was written by Ossandón et al. in 2001. For a more in depth look into the entirety of the district, the reader is encouraged to review that publication.

MMH was discovered by the Chilean Exploration Company, a part of the American Anaconda Company, in 1967, by accident (Sillitoe et al. 1996). The company was planning to use the area as a waste site for Mina Sur and drilling was conducted to ensure the gravel covered area was not underlain by economic mineralization, resulting in the breakthrough. The deposit was originally named MM for “Mansa Mina”. That name, which has a double meaning in Chilean Spanish, still persists in the literature and everyday usage, but the name of the mine was recently changed by the company to Mina Ministro Hales after a former Minister of Mines Mr. Alejandro Hales (1923-2001) thus MMH. In 1972 the Corporación Nacional del Cobre de Chile or CODELCO took over the Chuquicamata district and in 1978 conducted a geophysical survey and shallow drilling program. Through this work, a vein of copper mineralization was discovered to the west of the West Fault. From 1990-1992 approximately 35,000m of deep drilling was completed and from 1993 to 1995, a mine shaft and tunnels were also constructed. This investigation of MMH resulted in the recognition that it was a giant porphyry copper deposit (Sillitoe et al. 1996). As of 2009, there had been 170,000 m of core drilled and 7.3 km of underground workings (Boric et al. 2009). The most significant events and studies are listed in Table 1.1.

<u>Event</u>	<u>Company or Person Responsible</u>	<u>Date</u>
Discovery of MMH	Chilean Exploration Company	1967
Investigation into potential of deposit	CODELCO	1972
First abstract written on MMH	Alvarez & Miranda	1991
Structural, mineralogical and dating study	Zentilli et al.	1994
First comprehensive review of MMH and comparison with Chuquicamata	Sillitoe et al.	1996
Economic model of MMH	Muller & Quiroga	2003
Mineralogical study of sulphides	Tobey	2005
Review and update of the geology	Boric et al.	2009

Table 1.1 Significant studies and benchmarks of MMH.

1.5 Methods

The emphasis of this study is on petrology of the sub-volcanic porphyritic rocks. This thesis uses whole-rock geochemical analysis to compare samples from MMH with those of other deposits in the district that could possibly be connected to MMH. This objective is carried out by the evaluation of the major and trace element geochemical signatures of the intrusive and host rock units of MMH and comparing that with work previously done on other copper porphyry deposits of interest such as Chuquicamata, Fortuna and El Abra (see Fig. 1.3). This thesis also uses petrographic analysis done by optical microscopy and an electron-microprobe analysis to identify and characterize the silicate minerals as well as the sulphide minerals.

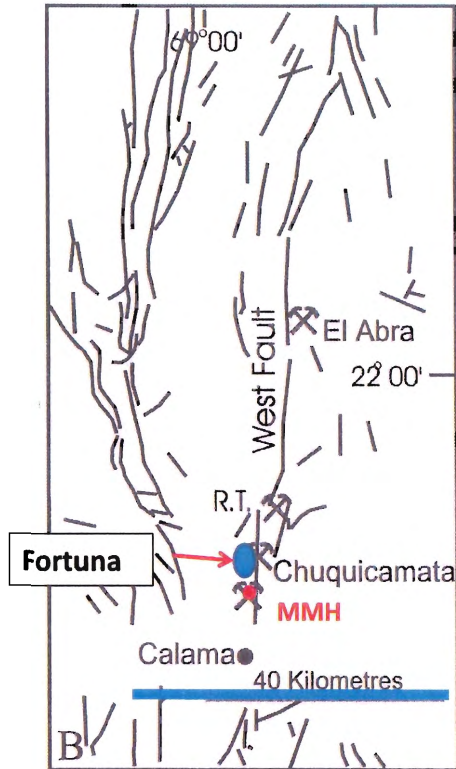


Figure 1.3 Location of MMH, Chuquicamata, Fortuna igneous complex and El Abra. (modified from Sillitoe 2010)

The data are then discussed in the context of other geochemical, mineralogical and geochronological data available.

CHAPTER 2: IGNEOUS ROCKS OF PORPHYRY COPPER DEPOSITS

2.1 Igneous Petrology

Porphyry copper deposits (PCD) are known to develop preferentially within calc-alkaline subvolcanic systems associated with subduction zones. The reader is directed to a comprehensive review by Richard Sillitoe, recognized authority on the subject and who has worked on MMH (Sillitoe 2010). According to the current models, porphyry systems, which develop at depths of ca. 4 km, are the shallower expressions of large batholiths at paleodepths of 5-15km. These batholiths transfer magma, fluids and heat through pipes and chambers to the overlying porphyry systems causing the host rock to fracture and thus allowing for magmatic-hydrothermal mineralization (see Fig. 2.1). This results in the development of pervasive veining and stock systems above the roof of the batholith, the geometry and placement of which are controlled by structural factors of the host rock.

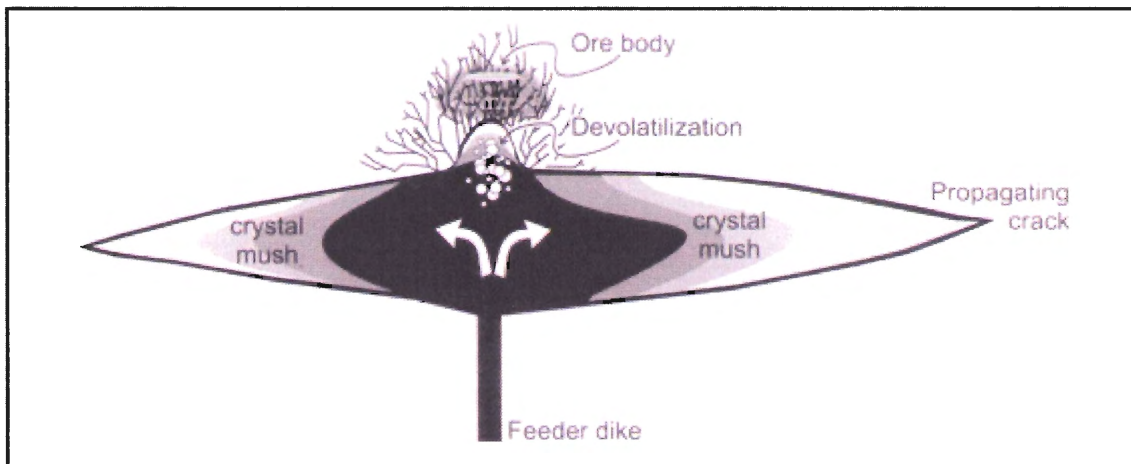


Figure 2.1 Illustration showing a magma chamber that been saturated with volatiles which has led to devolatilization and fracturing of the host rock. (Candela & Picolli 2005)

During the time of porphyry generation, the batholiths are suspended at neutral buoyancy (Sillitoe 2010). When new magma is injected into the batholith from beneath the more buoyant components of the magma (e.g. volatiles and silicates) rise to the top of

the batholith. This will continue until the top of the magma chamber is saturated in volatiles. The growth of feldspars and quartz, which are volatile free, supersaturate the remaining magma with supercritical fluid. At some point a fluid phase is separated that occupies more space than in solution; fluid pressure exerted by the volatiles becomes greater than the lithostatic pressure exerted by the host rock plus the tensional strength of the rocks, and the host rock will fracture (see Fig. 2.1). In the absence of tectonic stresses, fractures are random and stockworks irregular; where there is an active shear system (e.g. Chuquicamata, MMH, Lindsay 1997), fractures show preferred orientation and form dyke-like bodies. These fractures will be filled with the buoyant components of the magma as well as the devolatilized magmatic fluid. This process often repeats itself over an extended period of time (up to 5 million years in the case of Andean porphyry deposits (Sillitoe 2010), and creates successive generations of porphyries. The accumulation of these successive porphyries is what constitutes a porphyry system.

2.1.1 Magmatic Processes

Porphyry Cu deposits (PCD) are exclusively related to I-type, metaluminous calc-alkaline arc magma. The composition can range from calc-alkaline diorite, quartz diorite, and granodiorite to quartz monzonite (monzogranite).

As the name indicates, an inherent characteristic of PCD is that the ores are associated with porphyritic rocks. Porphyry intrusions generally contain phenocrysts of feldspar, quartz, hornblende and biotite with a fine-grained aplitic groundmass. The phenocrysts result from differential crystallization that occurs within the relatively stagnant magma chamber, prior to devolatilization. At deeper levels, within the magma chamber, the high Fe and Mg concentrations impede nucleation of felsic minerals such as

quartz and feldspars (Candela & Picolli 2005). As result, a relatively small number of phenocrysts form, also known as. Many of the feldspar phenocrysts show zoning which is result of slow growth under variable conditions. The aplitic groundmass forms from pressure quenching and devolatilization during the rapid ascent of the magma. The magma under these conditions becomes more felsic as well as supersaturated with volatiles. The combination of this rapid decrease in pressure and temperature facilitates the nucleation of small crystals which constitutes the groundmass of the porphyry.

Quartz in typical granites is anhedral and interstitial whereas in porphyries it is often phenocrystic. Many shallow porphyries with hydrothermal magmatic ore are called “quartz porphyries” due to their high proportion of “quartz eyes”. Quartz eyes are spheroidal quartz phenocrysts that are prevalent in quartz porphyries, even after alteration. Quartz eyes form due to the ease at which quartz nucleates in supercooled granite dykes. Many quartz phenocrysts show resorbtion features, such as reentrants representing dissolution of already formed phenocrysts (e.g. Candela & Picolli 2005). These rocks also typically contain some phenocrystic feldspar.

2.1.2 Hydrothermal Alteration Processes

Alteration-mineralization zoning in PCD affects several cubic kilometres of rock. The formation of the hydrothermal alteration zones in PCD is due to a two-phase fluid (which originates from a supercritical gas) that is emitted from the magma chamber. This two phase fluid is composed mainly of a low-density vapour as well as hypersaline liquid or brine. The brine has a salinity between 35-70 wt% (NaCl equivalent) whereas the vapour is composed mainly of volatiles such as SO₂, H₂S, CO₂, HCl, and HF. The Cu atoms reside in the vapour phase whereas the Mo resides in the brine. Sulfur ligands in

the vapour phase act as Cu transporting agents whereas the Mo is transported by Cl ligands in the brine (e.g. Candela & Picolli 2005).

PCD are characteristically developed in a stockwork, which is a complex system of structurally controlled or randomly oriented veins and veinlets. The stockwork system controls the ascent of the metal-rich magma fluid and thus, the Cu and Mo sulphides precipitate in the veins. Gustafson and Hunt (1975) have described and categorized these veins which will be discussed later in this chapter. As the magmatic fluid ascends, the temperature decreases which increases the S fugacity (e.g. Sillitoe 2010). As the S fugacity increases the precipitation of Cu sulphides changes as well as the alteration mineral assemblages. This change in mineral assemblages is what defines the various alteration zones. In stratigraphic (or temporal) order from lowest to highest, the alteration zones that occur are: sodic-calcic, potassic, chloritic-sericite, sericitic and advanced argillic (see Fig. 2.2). A propylitic zone can occur around the outer edges of the porphyry system (see Fig. 2.2). The deeper zones of the porphyry system are the first to be affected by hydrothermal alteration with the shallower hydrothermal zones occurring later. This creates overprinting of older zones by younger (i.e. chlorite-sericite zone overprints potassic zone) (Sillitoe 2010). It should be noted that the terms sericitic and phyllic are synonymous.

2.1.3 Alteration Zones

The sodic-calcic zone occurs at the base of the porphyry system and is typically not well defined (hence it is not represented in Fig. 2.2). Key minerals of this zone are albite and/or oligoclase, actinolite, magnetite \pm diopside \pm epidote \pm garnet. Veins that occur in this zone are comprised of magnetite and actinolite. The sodic-calcic zone is typically devoid of sulphides and thus, is of no economic interest (Sillitoe 2010)

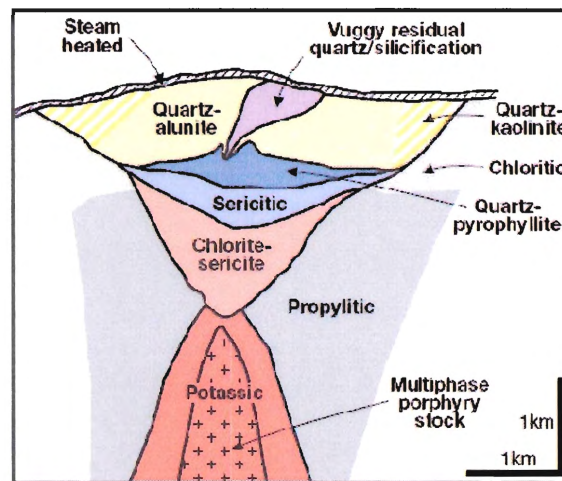


Figure 2.2 Vertical successions of porphyry alteration zones. It should be noted that the indicated quartz-kaolinite and quartz-alunite regions constitute the advanced argillic zone and the sericite and chlorite-sericite regions comprise the phyllic zone. Another point this figure displays is the vertical gap between the potassic and advanced argillic economic zones. (Sillitoe 2010)

The potassic zone constitutes the majority of the lower part of the Cu porphyry systems, especially in the Cu porphyry deposits of northern Chile (Fig. 2.2). Typical gangue mineral assemblages include: biotite, K-feldspar \pm actinolite \pm epidote \pm sericite \pm andalusite \pm albite \pm carbonate \pm tourmaline \pm magnetite. Biotite is the predominant alteration mineral in the more mafic porphyry intrusions whereas K-feldspar is more dominant in the more felsic porphyry intrusions with both types containing sodic plagioclase (Sillitoe 2010). Typical sulphide mineral assemblages that occur are: pyrite-

chalcopyrite, chalcopyrite ± bornite, bornite ± digenite ± chalcocite. Bornite signifies a high sulphidation state and occurs in the deeper, more central part of Cu ore deposits. The bornite-rich assemblages grade into chalcopyrite-bornite with decreasing sulphidation. As the potassic zone grades into the propylitic zone, increasing amounts of pyrite will occur as halos around the chalcopyrite, bornite and any other sulphides that may occur.

The chloritic alteration that occurs within the propylitic zone is a succession of potassic alteration and thus is a result of the same process (Fig. 2.2). This alteration is caused by chlorite replacing biotite and remaining amphiboles. The unit is characterized by chlorite-magnetite with occasional sericite and hematite (Lindsay 1997).

The phyllic zone is divided by Sillitoe (2010) into two categories: the chlorite-sericite and the sericitic zones. The chlorite-sericite is distinguishable by its pale green colour. It occurs in the stratigraphically higher portions of the Cu deposit (see Fig. 2.2) and often overprints pre-existing potassic alteration assemblages. A distinguishing feature of this zone is the alteration of biotite to chlorite, plagioclase to sericite and magnetite to hematite (Sillitoe 2010). Key gangue minerals that occur in this zone are: chlorite, sericite/illite and hematite. The main sulphide assemblage that occurs within this zone is chalcopyrite-pyrite (Sillitoe 2010). The sericitic alteration zone can partially or wholly overprint potassic and chlorite-sericite alteration assemblages. Partial overprint occurs as zoned halos around veins of sericitic alteration. There sericitic alteration can produce either a greenish-grey colour (such as in Chuquicamata) and host chalcopyrite-bornite assemblages or produce a white colour and contain only pyrite. Key minerals assemblages include: quartz, sericite ± prophyllite ± carbonate ± tourmaline ± specularite (Sillitoe 2010).

The advanced argillic zone forms a lithocap in the upper regions of the Cu porphyry environment. It is formed by low temperature and high sulphidation conditions as well as interactions with meteoric water. The development of advanced argillic lithocaps is more developed in lithologies with low acid-buffering capacities (such as felsic igneous rocks) than in those with high acid-buffering capacities (such as mafic igneous rocks) (Sillitoe 2010). Sillitoe (2010) divided the advanced argillic zone into two categories: the quartz-pyrophyllite zone and the quartz \pm kaolinite \pm alunite zone. The quartz-pyrophyllite (pyrophyllite is a hydrated alumina silicate, $\text{Al}_2\text{Si}_4\text{O}_{10}(\text{OH})_2$) is the stratigraphically lower unit of the advanced argillic lithocap and as such, it represents the higher temperature, higher sulphidation, lower pH region of the advanced argillic lithocap. The quartz \pm kaolinite \pm alunite (kaolinite is a type of clay, $\text{Al}_2\text{Si}_2\text{O}_2(\text{OH})_4$ and alunite is a type of sulphate $\text{KAl}_2(\text{SO}_4)_2(\text{OH})_6$) is the stratigraphically upper region of the advanced argillic zone. It forms under lower temperature, lower sulphidation and higher pH conditions.

2.1.4 Veining

Gustafson and Hunt (1975) identified three generations of vein types in the El Salvador Cu porphyry complex and their classification system is still accepted and used today. These vein categories in order of age are: A-type quartz veins, B-type quartz veins and D-type quartz veins.

A-type quartz veins contain silicate assemblages of: quartz + K-feldspar + anhydrite \pm biotite. The sulphide assemblage is comprised of: chalcopyrite + bornite \pm molybdenite. These veins range from 50 to 95% quartz. Occasionally, halos of K-

feldspar develop around these veins. As these veins are the first veins to form, they are typically cut by later B-type and D-type veins.

B-type veins contain silicate assemblages of: quartz + anhydrite with no K-feldspar. The sulphide assemblage is comprised of: molybdenite + chalcopyrite ± pyrite ± bornite. These veins typically have coarse grained quartz and lack alteration halos. These veins types typically truncate older A-type veins but are truncated by younger D-type veins.

D-type veins contain silicate assemblages of: anhydrite + quartz ± carbonate. The sulphide assemblage is comprised of: pyrite ± chalcopyrite ± bornite ± enargite ± sphalerite ± galena. These veins typically contain open spaces and truncate the older A- and B-type veins.

Sulphide mineralization associated with veining is not necessarily contained within the vein boundaries. Often time regions that are pervasively affected by veining will contain sulphide assemblages typical of the vein type(s) disseminated throughout the host rock (Corbett 2009)

2.1.5 Second Boiling

As mentioned above, second boiling occurs at the top of a magma chamber that is saturated with respect to volatiles (Fig. 2.3). Once the hydrostatic pressure of this silicate-volatile phase exceeds the confining pressure of the wall rock (~1 kb), the volatile phase will undergo an increase in volume. The magnitude of the pressure being exerted by this increase in volume will cause the wall rock to fracture, given this process is occurring in the upper limits of the crust (i.e. < 4km) (Burnham 1979). These fractures will occur in the direction of the less stress which generally is in an upward direction. In zones with no

active faulting or tectonic stress, this creates a concentric system of stocks and veins that originate at the top of the magma chamber and splay upward in a conical fashion (Fig. 2.3). The stocks and veins created by second boiling are filled with the buoyant crystalline magma that are mostly comprised of felsic minerals. This magma coexists with the volatile phase at the top of the magma chamber and with the two-phase magmatic fluid mentioned in the above (section 2.1.2). As hydrostatic pressures and temperatures decrease, the metals will exsolve from the two-phase fluid and precipitate within the veins and stocks.

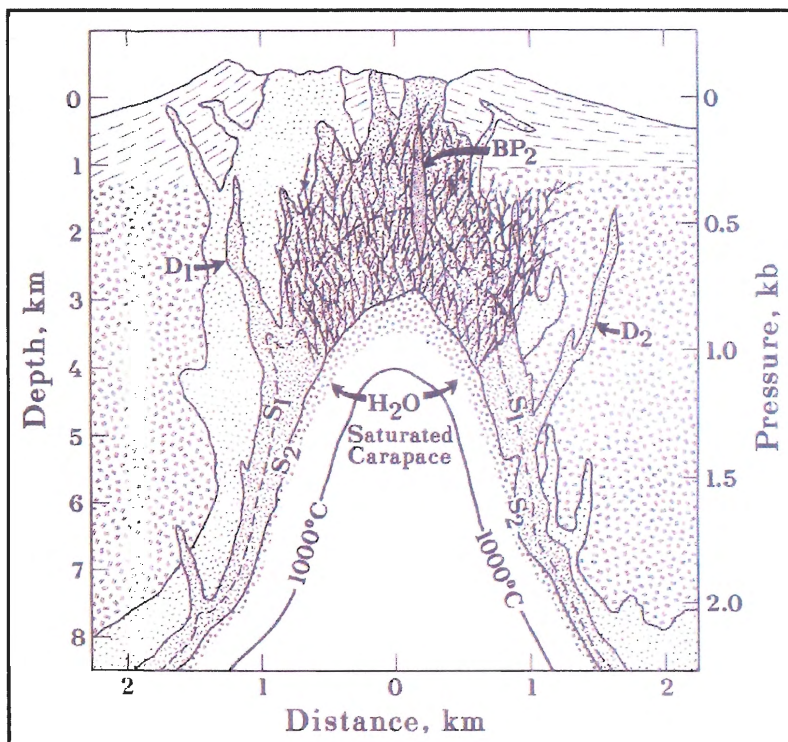


Figure 2.3 A magma chamber undergoing second boiling. S_2 represents the H_2O saturated solvus and S_1 represents the devolatilization boundary. D_1 , D_2 represent dykes that occur as result of second boiling and BP_2 represents a breccia pipe. The fractures illustrated above the magma chamber represent the veins and dykes that result from the increase in volume of the magmatic fluid. (Burnham 1979)

2.1.6 Supergene Enrichment

Supergene (meaning genesis from above) enrichment occurs by descending meteoric waters oxidizing and dissolving hypogene (primary) ore and redistributing it in a more horizontal manner, creating a supergene (secondary) enrichment cap. Sulphuric acid is created by oxidizing pyrite and thus, the more pyrite that is available, the more effective the supergene enrichment will be. This process occurs when the porphyry system is exhumed and is considered to be a form of weathering. Faulting can also facilitate supergene enrichment as it allows the groundwater to penetrate a greater surface area; thus, increasing oxidation capabilities. These oxides are typically not economical; however, from this supergene enrichment cap, metals leach from the oxidized ore and are carried down by groundwater and react with the hypogene sulphides at the supergene-hypogene boundary near the paleo-water-table. This process produces secondary sulphides with higher metal content than primary hypogene sulphides, hence “supergene enrichment”. This enhances the economic potential of the Cu ore. Porphyry deposits that have not undergone supergene enrichment and depend on their primary or hypogene Cu grade of ca. 0.5% may not be economical unless the metal prices are anomalously high.

When the supergene solutions move laterally down the water table, “exotic” deposits are formed, such as those mined at Mina Sur in Chuquicamata (Fig. 1.1.) and encountered east of the West Fault at MMH.

2.2 Tectonic Setting

Porphyry copper deposits are associated with convergent plate boundaries and the transfer of heat from the Earth's interior to the outer crust. Under oceanic-continental convergent conditions, serpentinized mantle along with oceanic crust, ocean-floor sediments and ocean water are subducted beneath the overlying continental lithosphere (Fig. 2.4). At depths which satisfy certain pressure and temperature conditions for various components of the descending slab, the buoyant, hydrous components will rise and metasomatize the overlying mantle. This results in hydrous arc magmas which form volcanic arc systems such as the Andes. Porphyries are a part of the arc systems and can be thought of as subvolcanic rocks.

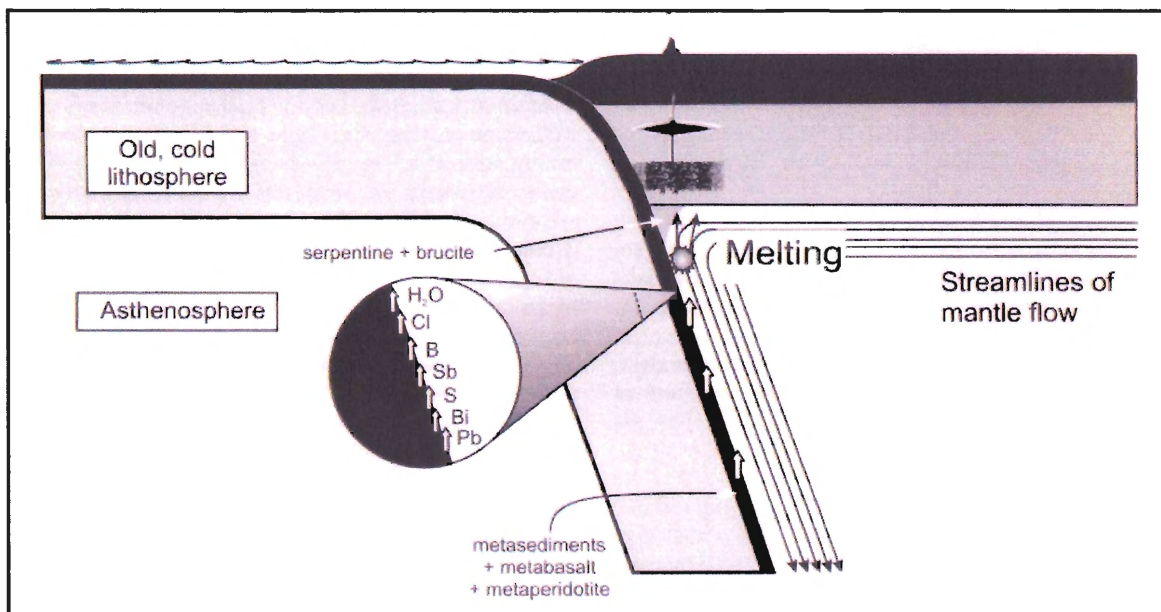


Figure 2.4 The oceanic-continental convergent environment under which porphyries form. This illustration emphasizes the input from the subducting slab, the mantle and the lower portions of the continental lithosphere. (Candela & Picolli 2005)

2.2.1 Origin of Cu

Although porphyries occur along all convergent margins, not all porphyries contain economic amounts of Cu. The reason for this is not known and can only be speculated. The origin of Cu in a porphyry system can come from a combination of places. It is possible for the Cu to come from the subducting slab itself if the ocean floor had additional Cu added to it through oceanic hydrothermal alteration processes. It is also possible to have Cu contributed from mantle sources. These sources include calc-alkaline magmas and basaltic andesites mixing with the ascending magma in an open-system environment beneath the crust (see Fig. 2.4). There is also evidence that lower crustal sources could be a contributor to Cu concentrations such as the Bagdad Cu-Mo porphyry deposit in northern Arizona (Barra et al. 2002). It is likely that many of the economical Cu deposits have contributions from all the sources mentioned above. Porphyries that form in environments that are devoid of such mechanisms will likely not be economical.

2.2.2 Structural Control of Porphyry Deposits

Obliquity of crustal deformation has control on the ratio of plutonism to volcanism. Strike-slip environments, such as that of the ancestral West Fault, creates local transpressional or pull-apart settings (e.g. Sibson 2001) which facilitate the ascent of magma in the lower crust and enhance permeability during the hydrothermal mineralization process. In northern Chile, the transtensional pull apart features that were caused by a reversal of dextral shear along the West Fault in the Eocene resulted in dilation and emplacement of large batholiths (Candela & Picolli 2005). These batholiths acted as the parental sources of magma, magmatic fluid and heat to the porphyries within the Chuquicamata district as well as the many other porphyries in close proximity.

2.2.3 Adakites

Term “adakite” was originally used to describe igneous rocks that were generated as result of the subduction of young (≤ 25 Ma) oceanic crust. Under these conditions, the basaltic portion of the subducting slab partially melts after dehydration (i.e. after it has undergone amphibolite to eclogite intensity metamorphism). As result of this process, adakites have the following geochemical signature: $\text{SiO}_2 \geq 56$ wt percent, $\text{Al}_2\text{O}_3 \geq 15$ wt percent, MgO normally < 3 wt percent, Mg number ≈ 0.5 , Sr ≥ 400 ppm, Y ≤ 18 ppm, Yb ≤ 1.9 ppm, Ni ≥ 20 ppm, Cr ≥ 30 ppm, Sr/Y ≥ 20 , La/Yb ≥ 20 , and $^{87}\text{Sr}/^{86}\text{Sr} \leq 0.7045$ (e.g. Richards and Kerrich, 2007). The use of the term adakite can be confusing as it implies both geochemical and genetic characteristics. Since the term was first used, rocks with adakitic geochemical characteristics have been discovered to exist in environments other than that previous described (e.g. Castillo 2006). Not only do adakites have implications for the study of crustal recycling, many of the world’s major Cu porphyry deposits have adakitic characteristics, namely high Sr/Y and La/Yb ratios (Richards & Kerrich 2007). This brings into question what the relation is between adakites and Cu porphyry deposits. This question is discussed in more detail below (see section 5.3), in the context of the geochemistry of MMH porphyries.

CHAPTER 3: GEOLOGY OF THE CHUQUICAMATA DISTRICT

3.1 Geology of the Chuquicamata Open Pit Mine

3.1.1 Chuquicamata Porphyry Complex

The Chuquicamata Porphyry Complex is comprised of three separate porphyries: the Este (East) Porphyry, the Oeste (West) Porphyry and the Banco Porphyry (Fig. 3.1), all of which were emplaced within the strike-slip fault system during the evolution of the Eocene-Oligocene magmatic arc (Arnott 2003). As mentioned in section 1.3, these porphyries are mineralogically similar but variations occur in texture and age. It should be noted that classification of this porphyry system has been variable due to intense alteration and deformation (Arnott 2003), however; it is generally accepted that three distinct porphyries exist in the Chuquicamata open pit.

The Este Porphyry is the oldest and most extensive of the group comprising greater than 80% of the exposed mineralized rock. The unit has a granular to porphyritic texture with phenocrysts phases (which comprise >65% of volume) in contact with one another (Lindsay 1997). The phenocrysts are comprised of the following minerals: euhedral plagioclase >2mm, poikilitic K-feldspar with biotite, plagioclase and quartz inclusions, biotite booklets, elongate quartz eyes and occasionally hornblende. The minerals comprising the medium-grained groundmass are deformed quartz, plagioclase and K-feldspar (Ossandón et al. 2001, Lindsay 1997). This unit has been classified by Lindsay (1997) as a matrix-poor monzogranitic porphyry. Mineralogically and compositionally, the dyke-like Banco and Oeste Porphyries are similar to Este leaving the units to be distinguished mainly by texture. The Banco Porphyry is more porphyritic and is also finer grained than the Este Porphyry (Ossandón et al. 2001). It is distinguished by having a bimodal distribution of plagioclase phenocrysts of different sizes in an

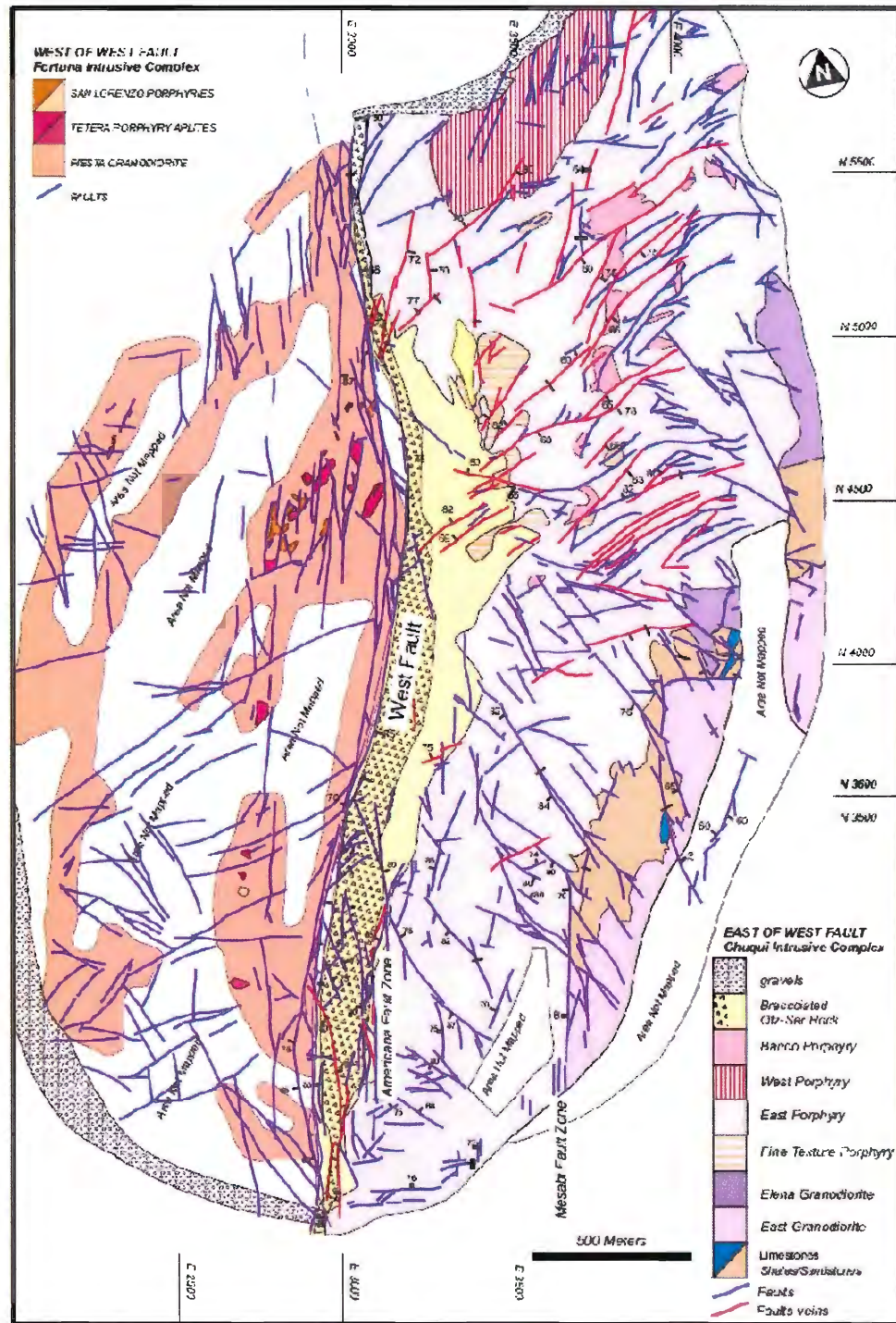


Figure 3.1 Geological map of the Chuquicamata open pit as of 1998. Local latitude and longitude coordinates along the sides in 500m intervals. Shading defines lithologic units and lines define fault and vein structures (Ossandón et al. 2001).

Rock Unit	Age range Ma	Method	Reference
Pre-mesozoic basement	300 - 200	Ar/Ar in biotite	Maksaev (1990)
Fortuna Complex	39.3 - 37.6	U-Pb zircon	Dilles et al. (1997); Ballard (2001)
Este Porphyry	34.6	ELA-ICP-MS	Ballard (2001)
Este Porphyry	33.9	Ar/Ar in biotite	Reynolds et al. (1998); Arnott (2003)
Potassic alteration	34.0 - 33.7	Ar/Ar in biotite	Reynolds et al. (1998); Arnott (2003)
Oeste Porphyry	33.5	ELA-ICP-MS	Ballard (2001)
Banco Porphyry	33.3	ELA-ICP-MS	Ballard (2001)
Banco Porphyry	33.1	Ar/Ar in K-spar	Reynolds et al. (1998); Arnott (2003)
Quartz-Sericite alteration	31.1	Ar/Ar in sericite	Zentilli et al (1994); Reynolds et al. (1998)
Supergene enrichment	19 - 15	Ar/Ar in alunite	Sillitoe & McKee (1996)

Table 3.1 Ages of the three porphyries (Este, Oeste & Banco) that occur in the Chuquicamata mine along with other relevant units, indicating geochronological methods used.

aphanitic quartz-feldspar-biotite matrix (Lindsay 1997, Ossandón et al. 2001). The contact with the Este Porphyry is sharp. The Banco Porphyry does not contact the Oeste Porphyry (Lindsay 1997). Arnott (2003) concluded that the Banco Porphyry was intruded 0.5 My after the end of potassic alteration mineralization and is thus less mineralized with Cu. Banco was affected by the quartz-molybdenum veining phase at ca. 32 Ma and the later quartz-sericite alteration-mineralization phase (ca. 31 Ma).

The Oeste Porphyry comprises <5% of the rock exposed in the Chuquicamata open pit and occurs mostly in the stratigraphically higher portions. The unit is distinguished from Este as being more matrix dominated. Geochronological dates indicate that the Banco Porphyry and the Oeste Porphyry intruded shortly after the Este Porphyry (table 3.1). Furthermore, Ossandón et al. (2001) suggest that due to the gradational boundaries between the porphyries, it is possibly that all three porphyries are in fact locally indistinguishable from each other and easy to confuse.

3.1.2 Fortuna Complex

The Fortuna Intrusive Complex occurs on the western side of the West Fault in the Chuquicamata open-pit mine (Fig. 3.1). It is 5 km wide and 22 km long (Dilles et al. 1997) and is mostly mined as waste to achieve slope stability (Arnott 2003). The proximity to the Chuquicamata Porphyry Complex is result of post-ore mineralization sinistral movement along the West Fault, which is discussed in section 3.3 (Ossandón et al. 2001). The Fortuna Intrusive Complex is comprised of 3 separate components (Fig. 3.1). The oldest is the Antena Granodiorite which Dilles et al. (1997) dated between 39.0 and 39.6 Ma. This unit contains fine to medium grained hornblende and biotite (Dilles et al. 1997). It is thought that this unit correlates with El Abra Granodiorite which is 35 km to the north on the eastern side of the west fault. The Fiesta Granodiorite is a younger unit (37.6 Ma from Dilles et al. 1997) that intrudes the Antena Granodiorite unit and is of similar composition. It is the largest unit within the Fortuna Intrusive Complex. The Fiesta Granodiorite grades into the San Lorenzo Porphyry System, including the Fortuna Leucocratic Intrusion, for which the ages are synchronous or slightly younger than Fiesta (Dilles et al. 1997).

Because the Fortuna Intrusive Complex is spatially juxtaposed to the Chuquicamata Intrusive Complex, the two were once thought to be of the same porphyry system, with the Fortuna Intrusive Complex representing the deeper roots of the Chuquicamata Intrusive Complex (e.g. Sillitoe 1973). Recent evidence (e.g. Arnott 2003) indicates that this concept is invalid. As mentioned above, Fortuna Intrusive Complex has been related to El Abra Granodiorite based on geochronological and field studies (39-38

Ma, thus 4-5 My older than Chuquicamata Intrusive Complex 34-33 Ma). Thus, Fortuna is not a cogenetic parent to Chuquicamata.

3.1.3 Pre-Chuquicamata Granodiorites

Both the Elena and East Granodiorites predate the Chuquicamata porphyries which lie to the west of them (Fig. 3.1). The East Granodiorite is the older of the two with textural evidence indicating no connection to the Chuquicamata Porphyries (Ossandón et al. 2001). The Elena Granodiorite however, is mineralogically and texturally similar to the Este Porphyry of the Chuquicamata Porphyry Complex (Ossandón et al. 2001). Radiometric dating of the Elena Granodiorite indicates a Jurassic to Early Cretaceous age (Ossandón et al. 2001). Both of the lithologies are economically barren (Ossandón et al. 2001), and form the Mesozoic host lithologies for the Chuquicamata Porphyry Complex. These Mesozoic host rocks are analogous in age and composition to the Triassic granodiorite (Granodiorite MMH) that represents the host of porphyries and mineralization at MMH (see Chapter 3.2)

3.1.4. Hydrothermal Alteration of the Chuquicamata Porphyry Complex

Chuquicamata rocks have been extensively altered by potassic alteration and high-sulphidation phyllic (quartz-sericite) alteration (Ossandón et al. 2001), but the distribution of alteration zones does not follow the classical concentric “onion” disposition of Lowell & Guilbert (1970) with potassic followed outwards by phyllic followed outwards by argillic and propylitic. Instead, the potassic alteration in Chuquicamata is followed outwards (to the east) by propylitic (chloritic) alteration, and the phyllic zone is central, adjoining the Falla Oeste (Fig. 3.2). Reynolds et al. (1998) demonstrated that the phyllic alteration is considerably younger than the potassic alteration and thus belongs to a

different cycle; the control exerted by the Falla Oeste and the age of this alteration show similarities with what is observed in MMH.

3.1.5 Supergene Enrichment

After the alteration-mineralization process was finished around 31 Ma, exhumation and erosion proceeded at around 50 m per million years (Maksaev & Zentilli 1999) with the most intense weathering occurring around 20 Ma when 450m of overlying rock had been removed (Sillitoe & McKee 1996). There is still some low grade supergene enrichment in the North pit area within the quartz-sericite zone (Ossandón et al. 2001). The Mina Sur south of Chuquicamata open pit consists of “exotic” copper oxide minerals (chrysocolla, etc.), which were derived from supergene enrichment, but were displaced a few kilometres from its source (hence exotic) into Tertiary gravels. Supergene enrichment also affected MMH at the same time as Chuquicamata, and its higher levels east of the West Fault contain some exotic ores identical to those of Mina Sur.

3.2 Geology of the MMH Deposit

The MMH deposit is located at 22°22'45" S, 68°54'50" W at an elevation of 2400 m above sea level in the Atacama Desert of northern Chile. It is 7 km south of the Chuquicamata open pit mine and 7 km north of the town of Calama (see fig 1.1). It runs parallel to West Fault on its western side and extends 7 km N-S and 200-300 m E-W with a depth of 1200 m below the bedrock surface (Boric et al. 2009) (Fig. 3.2). It is estimated to have 12.6 Mt (million metric tonnes) of Cu metal, qualifying it as a supergiant porphyry Cu-type deposit (Boric et al. 2009).



Figure 3.2 satellite image of the MMH deposit with ore grades superimposed where purple is high Cu grade (from MM mine records).

3.2.1 Eastern Portion of MMH Mine

On the eastern side of the West Fault, directly adjacent to the porphyritic complex, the basement rock is comprised of unmineralized Paleozoic metadiorite and amphibolites (Sillitoe et al. 1996). Due to the presence of a narrow fault-bounded basin, alluvial gravels, which overlie the bedrock, are 700-1000 m thick (Sillitoe et al. 1996, Boric et al. 2009). These gravels contain an exotic oxide blanket (Boric et al. 2009).

3.2.2 MM Granodiorite

The MMH deposit occurs on the western portion of the West Fault. The MM Granodiorite hosts the majority of the hypogene ore. Sillitoe et al. (1996) describes it to be a medium-grained hypidiomorphic-granular granodiorite and Boric et al. (2009)

describes it as pervasively altered. According to Proffett (2008), the MM Granodiorite has a quartz-rich-mafic poor portion with up to 35% quartz and 4-7% mafics as well as a quartz poor-mafic rich portion with up to 25% quartz and 10-12% mafics. Both Boric et al. (2009) and Proffett (2008) recognize the granodiorite to be Triassic in age. This Granodiorite is cut by two north-striking porphyritic dikes which will be discussed in the following sections.

3.2.3 MM Porphyry

The MM Porphyry lies deep in the MMH deposit adjacent to the West Fault (Boric et al. 2009). It contains euhedral phenocrysts of plagioclase, biotite and biotitized hornblende throughout the unit with some locations containing anhedral interstitial quartz and K-feldspar between the phenocrysts, and others with a fine grained aplitic groundmass (Proffett 2008). The unit was dated with U/Pb zircon methods which provided an age of 38.9 ± 0.4 Ma (Boric et al. 2009). The unit has undergone potassic alteration so that secondary K-feldspar appears as rims and veinlets within K-feldspar (Boric et al. 2009). Opaque minerals that occur within the unit are bornite, chalcopyrite, chalcocite and minor digenite.

3.2.4 MM Quartz Porphyry

The MM Quartz Porphyry is distinguished by its abundance of quartz eyes (spheroidal quartz crystals that are millimetres in diameter) (Boric et al. 2009). It has been affected by potassic alteration and is noted to have blue-green sericite, K-feldspar as well as anhydrite (Boric et al. 2009). This unit cuts the MM Porphyry and is dated by U/Pb methods to be 35.5 ± 0.6 Ma (Boric et al. 2009).

3.2.5 Other Geologic Units

A rock mapped as Hornblende Porphyry was sampled in DHH 3173 at 888.5 m, but there is insufficient data to characterize it; dark patches consist of chlorite; its geochemistry is similar to MM Porphyry.

Dacite dykes cut all of the units mentioned above and are classified by their assemblage of hornblende and plagioclase with minor quartz (Boric et al. 2009). They are cut by the hydrothermal breccias and some are considered Mesozoic in age (Boric et al. 2009).

The hydrothermal breccias are spatially “rooted” in the MM Porphyry (Boric et al. 2009). Areas that contain rotated quartz clasts show the highest Cu grades in the MMH deposit along with As, Ag and Zn (Boric et al. 2009). The hydrothermal breccia unit shows argillic alteration in places and represents a transition between typical Cu-Mo porphyry and epithermal high sulphidation mineralization (Boric et al. 2009).

3.2.6 Hypogene Mineralization

The earliest phase of hypogene mineralization was the potassic alteration. The mineral assemblages associated with this phase include K-feldspar and quartz altered from plagioclase as well as chloritized remnants of hydrothermal biotite and hornblende (Sillitoe et al. 1996). This unit contains opaque assemblages such as bornite-chalcopyrite nuclei with chalcopyrite halos (Boric et al. 2009).

The next sequence in hypogene alteration was sericitic alteration whereby previous textures were obliterated and minerals were replaced with quartz and sericite. This phase contains the majority of Cu-bearing minerals (Sillitoe et al. 1996). This mineralization

produced sulphides such as pyrite and enargite and is found in the upper western portions of the deposit (Sillitoe et al. 1996).

The last stage of hypogene mineralization occurred as late stage high sulphidation Cu (Ag + As) with advanced argillic alteration and the formation of hydrothermal breccias discussed in section 3.2.5 (Boric et al. 2009). This zone is comprised of quartz, alunite, pyrophyllite and sericite with minor kaolinite (Sillitoe et al. 1996). This mineralization zone is contained in the easternmost portions of the MMH unit in the top 400m of the deposit (Sillitoe et al. 1996). As indicated above, supergene and exotic copper minerals occur in the near-surface of the MMH deposit.

3.2.7 Ages of Hypogene Phases

Boric et al. (2009) report Re/Os ages on molybdenite that suggest the first hypogene mineralization phase occurred at ca. 37.3 Ma. This phase may be associated with the recognized Eocene porphyries, but no intrusions have been identified as the parents of the high sulfidation bodies with hypogene alunite that yields $^{40}\text{Ar}/^{39}\text{Ar}$ ages of 31.4 to 32.2 Ma (Boric et al. 2009).

3.2.8 Supergene Mineralization

Supergene mineralization and weathering has affected the upper 60 to 150 m of sulphide bearing country rock in the MMH deposit (Sillitoe et al. 1996). The supergene leached cap composes the top 50 to 60m which is underlain by partially oxidized sulphides and then chalcocite enrichment underlies this (Sillitoe et al. 1996). The alunite produced from supergene enrichment in the central MMH body was dated using K/Ar methods to produce dates between 20.8 +/- 0.6 Ma and 20.4 +/-0.6 Ma (Sillitoe et al. 1996).

3.3 The West Fault

Structural events within the Chuquicamata District have been controlled by the West Fault which is the northern most section of the Domeyko Fault System (Tomlinson & Blanco 1997). The fault is by far the most prominent structural feature in the area as it is a 3m thick gouge and brecciated zone that cuts mineralization of the Chuquicamata porphyries. A dextral precursor to the West Fault is represented in the Chuquicamata deposit by a high-temperature ductile shear zone (Lindsay 1997). Transtensional domains associated with it may have facilitated the emplacement and mineralization of super giant Cu porphyries within the Chuquicamata District; a weak domain in the crust that allowed magma to rise rapidly to the near surface. The brittle equivalent of this ancestral fault, now the West Fault, was responsible for the post mineralization displacement of those same porphyries it helped create (e.g. Tomlinson & Blanco 1997). This section examines both the horizontal and vertical displacement of the West Fault and its evolution over time.

3.3.1 Horizontal Displacement of the West Fault

The first tectonic event that occurred in the region was a shortening event which occurred between 44 and 38.5 Ma (Eocene) (Tomlinson & Blanco 1997). This age correlates with the emplacement of the El Abra Mine to the north of Chuquicamata as well as the Fortuna Intrusive Complex and the MMH deposit. With the knowledge that El Abra and Fortuna are of the same origin, this puts a lower constraint on the age of strike-slip movement within the West Fault to 37 Ma which is the age of final cooling of the Fortuna-El Abra complex (Tomlinson & Blanco 1997). The West Fault must have formed by 31 Ma as it cut the sericitic alteration within Chuquicamata which has been dated by

Zentilli (1995) as 31 Ma (Tomlinson & Blanco 1997). If the Chuquicamata Porphyry System was in fact emplaced as result of the formation of the West Fault system, then the dates from those porphyries indicate the fault began to develop around 34-33 Ma (Tomlinson & Blanco 1997). Dextral mylonitic shears within these porphyries indicate that the fault's movement was dextral from its initiation until 31 Ma (Lindsay 1997; Tomlinson & Blanco 1997).

At 31 Ma there was a reversal in movement along the West Fault due to a shift in the tectonic nature of the Arica Bend (refer to Fig. 1.2) to the north (Tomlinson & Blanco 1997). This new sinistral strike-slip motion along the West Fault began post-mineralization with respect to the Chuquicamata Porphyry System. This movement continued until around 19 Ma as the upper age constraint is provided by the Huasco Ignimbrite which overlies the West Fault and shows no signs of sinistral displacement (Tomlinson & Blanco 1997). Field correlations collected by Tomlinson & Blanco (1997) indicate that the amount of net sinistral displacement within the Chuquicamata District is equivalent to 37 km for the Chuquicamata Porphyry System and 35 km for the Fortuna-El Abra Complex.

3.3.2 Vertical Displacement of the West Fault System

McInnes et al. (1999) used the apatite (U-Th)/He thermochronometry method to measure the vertical displacement along the West Fault system. This method is effective for dating tectonic events that occur in the uppermost crust as it measures the cooling history between 45°C and 90°C. Through this method, they determined there has been a minimum of 600 m \pm 100 m of post mineral vertical displacement along the West Fault with the western block uplifted relative to the eastern block. Since Chuquicamata is at an

elevation of ~3000 m above sea level, these findings would imply that the western half of Chuquicamata must be at an elevation of ~3600m somewhere along the western flank of the West Fault. MMH is located at 2400 m elevation and thus, this data indicates that MMH cannot be the other half of Chuquicamata. The eastern half of MMH should be at an elevation of roughly 1800m somewhere along the east side of the West Fault (McInnes et al. 1999). Part of the interpretation of McInnes et al. (1999) was challenged by Tomlinson et al. (2001) who argue that using their extensive field evidence for 35-37 Km strike-slip displacement, the missing portion of the Chuquicamata deposit may have escaped exhumation, depending on the full displacement path, and may lie immediately below the surface between Calama and Sierra Limón Verde to the south, “waiting to be discovered and exploited”.

CHAPTER 4: PETROGRAPHY OF THE MMH DEPOSIT

Petrography of the MMH samples was determined using both optical microscopy methods and electron microprobe analysis (EMP). Both methods were conducted at Dalhousie University laboratories.

4.1. Optical Microscopy

Optical microscopy was applied to 15 core chip samples from MMH that were selected as representative samples by M.Zentilli in 2010 (Table 4.1). From these samples, both thin sections and polished sections were made and analyzed. The thin sections were used to identify silicate minerals and textures using plain polarized light (PPL) and cross polarized light (XPL). Polished sections were observed using reflected light and were used to identify sulphide minerals and textures.

Samples of MMH analyzed by Activation Labs Inc				
Sample Q series	Diamond Drillhole	From m	To m	Rock Unit in Log
ZMMH-04	3173	888.50	888.63	Hb Porphyry
ZMMH-05	3173	853.75	853.86	MM Grd
ZMMH-07	4903	1012.00	1012.25	PMM
ZMMH-09	4884	158.90	159.00	PMM
ZMMH-10	5014	161.00	161.20	MM Grd
ZMMH-11	4976	897.87	898.00	POQ
ZMMH-12	4976	1128.00	1128.12	POQ
ZMMH-13	4976	1123.78	1123.90	POQ
ZMMH-14	4976	1241.00	1241.12	PMM
ZMMH-15	4976	1286.47	1286.60	PMM
ZMMH-17	4976	855.40	855.50	MM Grd
ZMMH-18	4976	845.85	845.95	MM Grd
ZMMH-20	4976	870.90	871.05	MM Grd
ZMMH-21	4976	883.20	883.24	PMM
ZMMH-23	4976	1064.00	1064.15	MM Grd
Cu 1333	Chuquicamata pit			Banco Porphyry
Cu 2022	Chuquicamata pit			Este Porphyry

Table 4.1 Drillcore samples from MMH used for petrographic and whole-rock geochemical analysis. Included in the whole-rock geochemical analysis were two samples from Chuquicamata, one from the Banco Porphyry unit and one from the Este Porphyry unit.

4.1.1. MM Porphyry

Observations

The PMM samples contain feldspar phenocrysts, biotite, quartz \pm feldspar groundmass, veins and sulphides.

The feldspar phenocrysts range in length from 5 mm to 2 mm and are generally subhedral prismatic (Fig. 4.1). Most of these feldspar phenocrysts are plagioclase with a few K-feldspar. Primary zoning is observed in some of the feldspar phenocrysts as many have alteration rims and penetrative alteration veins (Fig. 4.2). The samples are variably affected by sericitization whereby some of the samples show little or no remnant feldspar, whereas others only contain sericite along cleavage planes within feldspar.

The biotite that is observed occurs within clusters of variable size where the micaceous crystals are aligned sub-parallel to one another and look to be “shredded”. The clusters are irregularly shaped and vary in size between 1.5mm and 0.5mm. Within some of the larger biotite crystals, microlath inclusions of rutile are observed. In sample ZMMH-14 (DDH 4976) these rutile inclusions are euhedral and aligned in two directions at angles of 60°-120° (see photo within appendix 2). Within some samples, there are poikilitic biotites that contain inclusions of quartz, anhedral rutile (probably exsolution from biotite or recrystallization from titanite), apatites, zircons and sulphides (eg ZMMH 7). Much of the biotite has undergone chloritization.

The groundmass ranges from medium to fine grained quartz and occasionally contains sericite, presumably from altered feldspars (Fig. 4.2).

The veins are mostly comprised of quartz along with sulphides such as chalcopyrite, pyrite and chalcocite. The veins range in width from >0.5mm to 1.5 mm and contain vugs along with iron oxide.

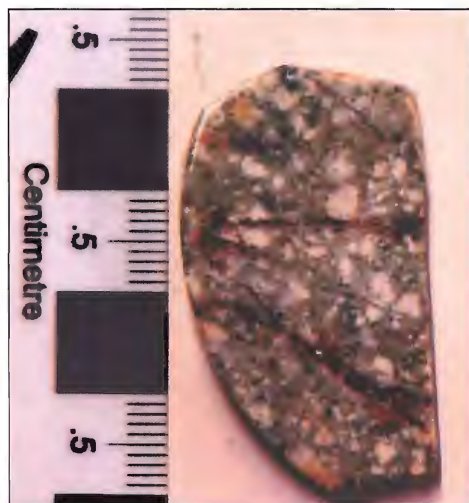


Figure 4.1 Photograph of a core chip sample of the MM Porphyry

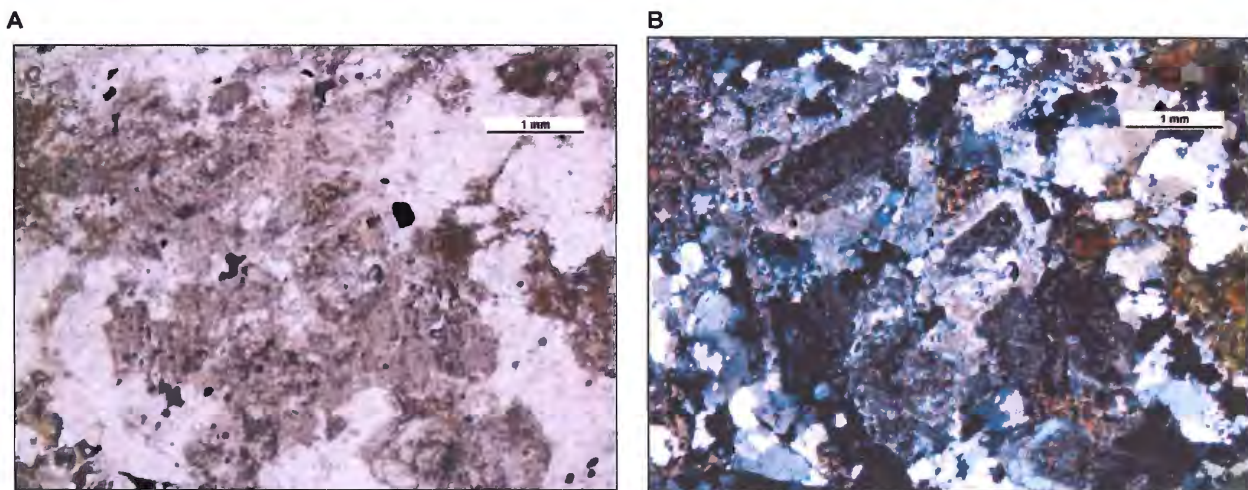


Figure 4.2 Microphotographs of the MM Porphyry. **A)** was taken in plain polarized light (PPL) and shows how the altered feldspars have accumulated a dingy brownish-grey colour; **B)** was taken in cross polarized light (XPL) and shows zoning of the feldspar phenocrysts.

Discussion

The size and zoning of feldspar phenocrysts is typical of porphyry-type rock. The alteration rims and penetrative alteration veins observed in most samples are characteristic of textural overprint by potassic and/or phyllic alteration phases (Fig. 4.2). The variable amount of sericitization that occurs within the feldspar phenocrysts in each sample is indicative of the level of alteration in that sample. Many of the samples contain evidence of initially undergoing potassic alteration, then subsequent phyllic alteration. See appendix 2 for description of secondary alteration of individual samples

The biotite is characteristic of hydrothermal mineralization. The preferred alignment of the elongate crystals indicates that they formed under a stress field such as would occur during hydrothermal alteration. The poikilitic texture observed within the biotite is also indication of the secondary nature of the mineral (see photograph of ZMMH-14 in appendix 2). Furthermore, Tobey (2005) has attributed a “shredded” texture (whereby the edges of the biotite crystals are rough and uneven, see appendix 2) to biotite in MMH that is of secondary nature (see photograph of ZMMH-14). The exsolution of the rutile was likely caused by alteration of the primary (igneous) biotite or titanite. The “immobile” behaviour of titanium in the rocks, where Ti changes mineralogy but remains in the same volume of rock, is significant when the geochemistry of the rocks is discussed in Chapter 5.

Due to the incompatible chemical nature of titanium, the element was exsolved during the alteration process to form rutile. The presence of rutile poikiloblasts within biotite crystals indicates that the biotite overprinted primary mafic minerals (possibly

amphibole). The chloritization of biotite that occurs in many samples is likely a result of chlorite-sericite alteration.

The occurrence of veins within the samples observed is controlled by alteration levels as well as structural controls. There is no characteristic vein type that is associated with a certain lithologic unit. Using Gustafson & Hunt's (1975) vein classification guide as described in section 2.1.4, all three types of veins (A, B, D) were observed to be present within the MM Porphyry samples. The sulphides that are present in the sample are likely a reflection of these vein types as well as secondary alteration. Veining and sulphide assemblages are discussed in more depth with respect to the individual samples in appendix 2.

4.1.2 MM Granodiorite

Observations

Granodiorite MM is a hypidiomorphic, equigranular assemblage of feldspar, quartz and chloritized biotite (Fig. 4.4). It is not a "porphyry". The feldspar is intergrown, zoned and commonly shows alteration rims with a sericitized center. Many of the samples have been so pervasively sericitized that there is no remnant feldspar. The crystal size ranges from 1.5mm-3mm along the length.

The quartz grains are anhedral and range in size depending on the amount of alteration that the sample has undergone. Samples that have undergone more alteration have larger quartz grains and a higher proportion of quartz in the sample.

The chloritized biotite occurs in irregularly shaped clusters ranging in size from 2 mm to 0.5 mm this remnant biotite is often poikiloblastic with inclusions of quartz, rutile, sulphides and apatites.

The sulphides are disseminated and typically occur within close proximity to the biotite remnants or within veins. Veining occurs irregularly throughout the samples observed with thickness of >0.5mm to 2.5mm.



Figure 4.3 This is an example photo of a core chip sample of the MM Granodiorite

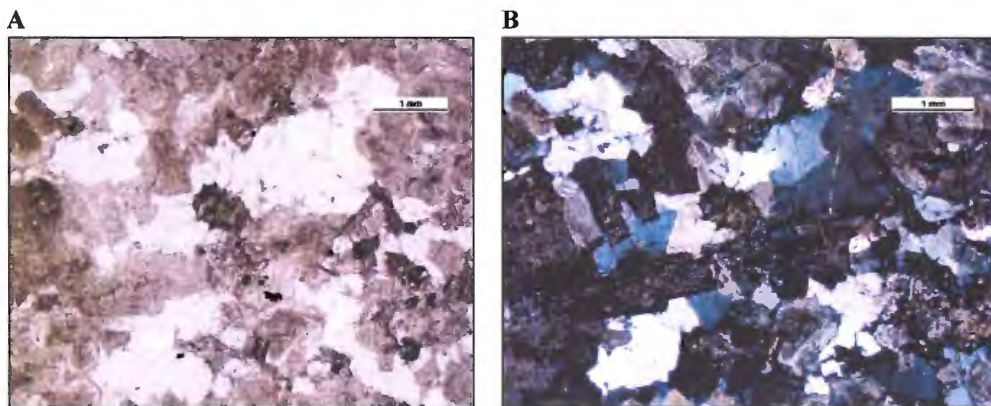


Figure 4.4 Microphotographs taken of the MM Granodiorite. **A)** was taken in plain polarized light (PPL) whereas **B)** was taken in cross polarized light (XPL). Both of these photos display the hypidomorphic texture observed in the MM Granodiorite. The PPL photo shows the brownish-grey colour that the feldspars accumulated from secondary alteration.

Discussion

The hypidomorphic texture observed as well as the silicate mineral assemblage of feldspar + quartz + biotite is typical of a granodiorite.

The hydrothermal alteration of Granodiorite MM greatly affected the texture of the samples. The pervasive alteration of feldspars which has resulted in sericitization as well as alteration rims has altered the original shape and character of the feldspars and has in some cases left them unrecognizable.

The correlation of the proportion of quartz and alteration is likely the result of deterioration of feldspar to sericite + quartz under acidic alteration conditions. This indicates that the quartz within the MM Granodiorite could have originated anywhere between primary igneous conditions to sometime during the subsequent secondary alteration processes.

Further evidence for pervasive alteration of the MM Granodiorite samples comes from the chloritization of biotite in all of the obtained samples. The poikiloblastic texture observed indicates that the biotite was likely secondary before it was chloritized.

The veining that occurs is a function of the alteration that the individual sample has undergone and has more to do with the location of the sample than the primary igneous lithology. The sulphides that are present in the samples were emplaced by secondary alteration rather than primary igneous processes. For a further discussion of the vein types and sulphides that occur in each sample see appendix 2.

4.1.3 MM Quartz Porphyry

Observations

The MM Quartz Porphyry samples are characterized by large rounded quartz phenocrysts with diameters >6mm (Fig. 4.6). There are also feldspar phenocrysts which have been completely sericitized in two of the three samples. The remnant feldspar that does occur is zoned and is affected by sericitization along its cleavage planes. These

feldspars are between 4.25 mm and 0.5 mm long. The groundmass is aplitic and is comprised of quartz with sericite. The unit contains pervasive veining with some of the veins containing vugs. Sulphides generally occur within or in close proximity to the veins.



Figure 4.5 This is an example photo of a core chip sample of the MM Quartz Porphyry

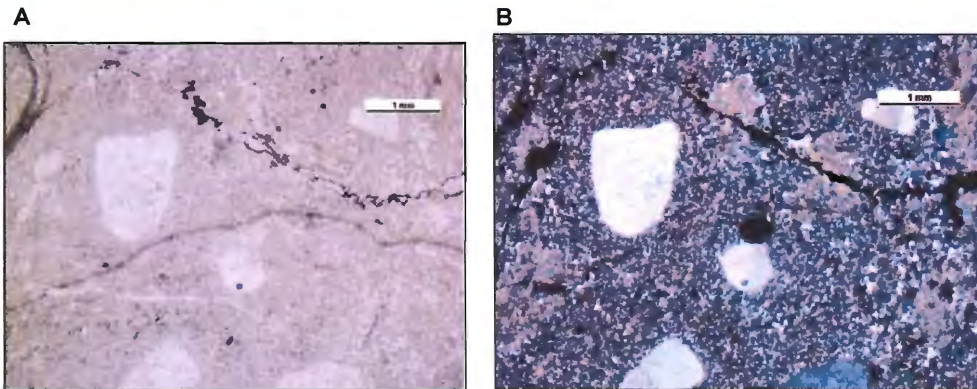


Figure 4.6 Microphotographs photos taken of the MM Quartz Porphyry. **A)** was taken in plain polarized light (PPL) **B)** was taken in cross polarized light (XPL). Both of these photos show the characteristic quartz phenocrysts known as “quartz eyes”

Discussion

The bimodal grain size is what classifies this unit as porphyry. The MM Quartz Porphyry is characterized by the large, spheroidal quartz phenocrysts which are commonly referred to as “quartz eyes” (Fig. 4.6).

The more felsic composition of this unit means that it contains a high proportion of feldspar and thus, it is more susceptible to sericitization when undergoing acidic alteration phases. This is likely why the majority of the samples are so pervasively altered. Another implication of the felsic composition of this unit is that it would have formed after the more mafic MM Porphyry which contains biotite. This is supported by the U/Pb dating cited in Boric et al. (2009).

4.2 Electron Microprobe Analysis

The electron microprobe analysis works by bombarding a carbon-coated polished section with an electron beam. This causes x-rays to be emitted from the point where the electron beam was shot. Having calibrated the microprobe with known standards, the emitted x-ray levels can be quantitatively measured and used to identify the proportion of elements present in the spot being analyzed to an accuracy and precision approaching 1% (Goldstein et al. 2003). The elements that were analyzed for in the silicates were: K, Cr, Mn, Na, Al, Ca, Ti, Mg, Si, Fe, P, Cl, Ni, F, S and Ba. 150 points were analyzed.

In this study, three polished sections were analyzed in the electron microprobe, each one representing one of the three lithologies identified at MMH. Sample ZMMH-14 (DDH 4976 1241.00-1241.12m) was used to represent the MM Porphyry, ZMMH-10 (DDH 5014 161.0-161.2m) was used to represent the MM Granodiorite and ZMMH-11

(DDH 4976 897.87 – 898.00 m) was used to represent the MM Quartz Porphyry (see table 4.1). These samples were selected because they appeared to be the least altered samples.

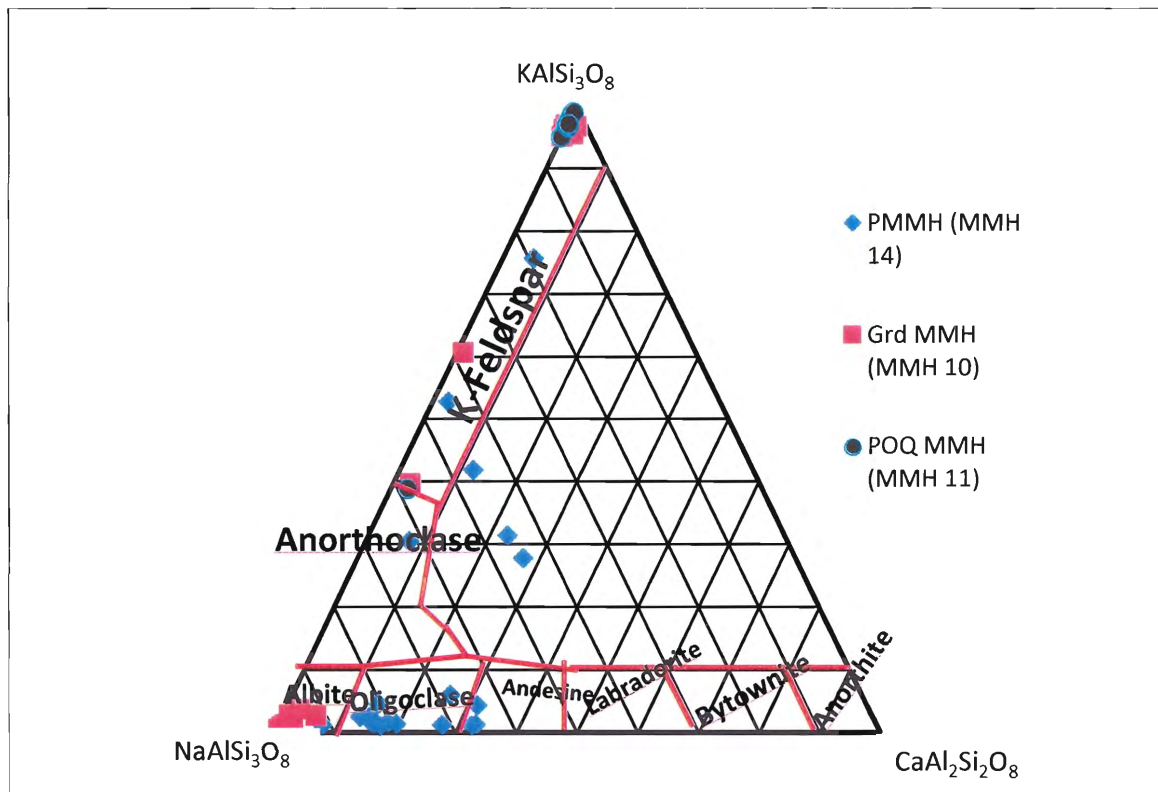


Figure 4.7 A ternary diagram of the feldspar end member classification. The diagram shows that the majority of the MM Porphyry data points (PMMH) fall into the oligoclase field, indicating that albitization has not been intense in this rock unit. The plagioclase in the Granodiorite MM (Grd MMH) has been completely albitized. Thus a distinction between the two rocks can be made using the feldspar composition. The MM Quartz Porphyry (POQ MM) mostly falls into the K-feldspar field.

4.2.1 MM Porphyry

Feldspar end member	Proportion
K-feldspar	39%
Albite	0%
Andesine	13%
Oligoclase	35%
Anorthoclase	4%
Unstable	9%

Table 4.2 The proportion of each feldspar end member with respect to the population of feldspar phenocrysts within the MM Porphyry. Taken from sample ZMMH-14 (DDH 4976 1241.00-1241.12m) with EMP analysis. Total number of feldspar points that were analyzed is 23.

Discussion

The data indicates that K-feldspar phenocrysts have a uniform composition whereas the plagioclase phenocrysts that are predominantly oligoclase contain portions that are andesine, anorthoclase, K-feldspar or unstable (see appendix 1). The phenocrysts with a K-feldspar composition could have originated under primary conditions or under potassic alteration conditions. Barium was not tested for in this sample so it is not possible to tell which of the two situations has occurred. It is likely that within the sample there is both primary and secondary K-feldspar. The oligoclase phenocrysts are likely relatively unaltered phenocrysts with compositions that reflect primary igneous conditions. The majority of the data points in plagioclase from the MM Porphyry are categorized as Oligoclase (see Fig. 4.7 and table 4.2). This is an important distinction because plagioclase in the Triassic MM Granodiorite is completely albitized.

The electron microprobe analysis of feldspars is therefore very useful in distinguishing between MM Porphyry and MM Granodiorite.

4.2.2 MM Granodiorite

Feldspar end member	Proportion
K-Feldspar	39%
Albite	52%
Andesine	0%
Oligoclase	0%
Anorthoclase	0%
Unstable	9%

Table 4.3 The proportion of each feldspar end member with respect to the population of feldspar phenocrysts within the MM Granodiorite. Taken from sample ZMMH-10 (DDH 5014 161.0-161.2m) with EMP analysis. Total number of feldspar points that were analyzed is 23.

Discussion

EMP data shows that K-feldspar and albite often occurred together in the same phenocryst (see appendix 1). This secondary mineral assemblage is typical of potassic alteration as the alkalis are very mobile under acidic conditions. The exclusive presence of albite and K-feldspar (see Fig. 4.7 and table 4.3) indicates that the MM Granodiorite samples have been pervasively altered, which is very different from MM Porphyry.

4.2.3 MM Quartz Porphyry

Feldspar end member	Proportion
K-feldspar	96%
Unstable	1%

Table 4.4 The proportion of each feldspar end member with respect to the population of feldspar phenocrysts within the MM Granodiorite. Taken from sample ZMMH-11 (DDH 4976 897.87 – 898.00 m) with EMP analysis. Total number of feldspar points that were analyzed is 26.

Analysis that contain Ba	38%
Analysis that do not contain Ba	62%

Table 4.5 The number of feldspar analyses in sample ZMMH-11 that contain barium compared to those that do not.

Discussion

The results from the EMP analysis indicate that all analyzed feldspar phenocrysts were K-feldspar (see Fig. 4.7 and table 4.4). This agrees with the mineralogically evolved composition of the MM Quartz Porphyry. Arnott (2003) stated that primary K-feldspars tend to contain barium whereas secondary K-feldspars do not. Therefore, the EMP analysis indicates that the phenocrysts in the MM Quartz Porphyry were likely primary K-feldspar. When the samples underwent potassic alteration, portions of the primary K-feldspar was replaced by secondary K-feldspar (table 4.5).

4.2.4 Sulphides

As previously mentioned, sulphide mineralization is more dependent on veining and secondary alteration processes than it is on primary lithology. For this reason, the results are more indicative of alteration processes rather than reflective of primary igneous lithology. Since the samples were taken at various depths from different cores, these EMP

results are likely more reflective of the local sulphide assemblages rather than that of the lithologic units. The elements analyzed using the EMP for the sulphides were: Mo, S, As, Fe, Pb, Ag, Co, Ti, Ca, Cu, Sn, Sb, Mn, Zn and W. 97 points were analyzed.

Results

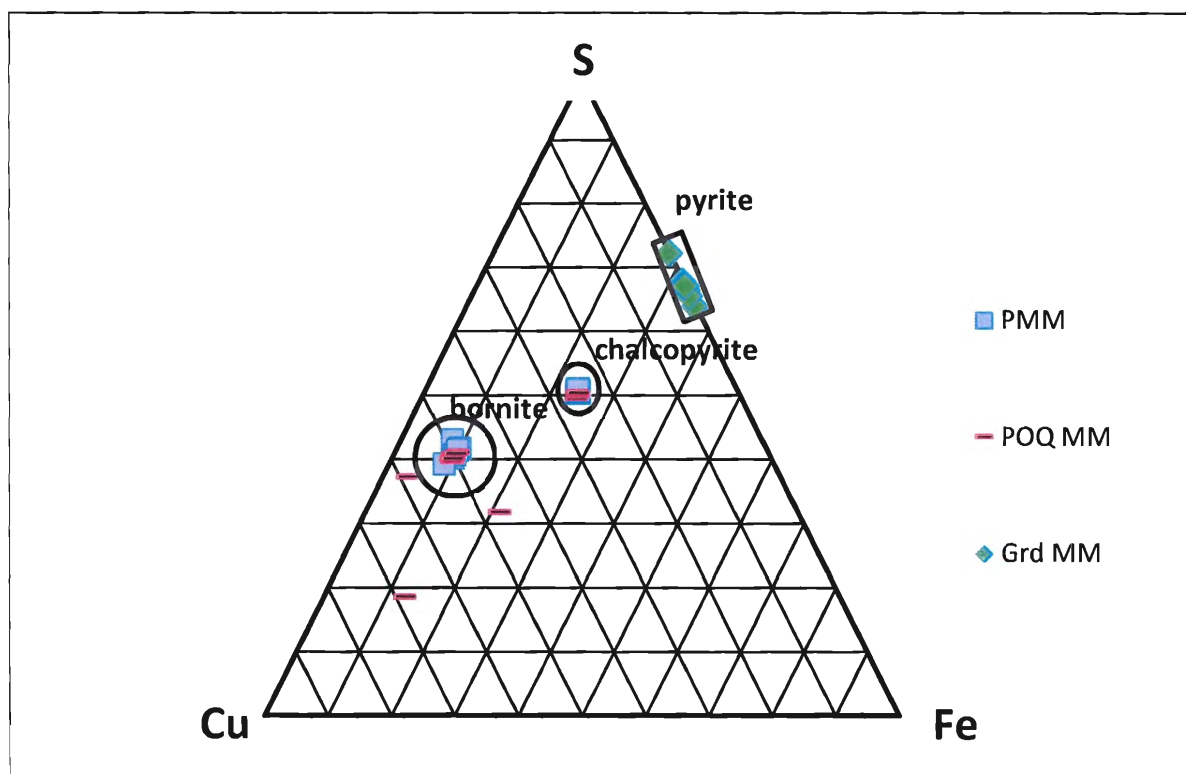


Figure 4.8 Geochemical fields of three sulphide minerals were superimposed on this ternary diagram in order to classify the result from EMP analysis.

Sulphide Mineral	Proportion in MM Porphyry	Proportion in MM Granodiorite	Proportion in MM Quartz Porphyry
Chalcopyrite	71%	0%	37%
Bornite	29%	0%	63%
Pyrite	0%	100%	0%

Table 4.6 The proportions of chalcopyrite, bornite and pyrite found in the EMP analysis of the MM Porphyry, the MM Granodiorite and the MM Quartz Porphyry.

Discussion

The proportion of chalcopyrite and bornite in the MM Quartz Porphyry and in the MM Porphyry is typical of potassic alteration as well as A-type and B-type veining. It is interesting that the MM Granodiorite contains only pyrite and no chalcopyrite or bornite. Optical microscopic observations confirm this finding. The EMP results of the feldspars in Chapter 4.2.2. indicate that the unit has undergone secondary alteration and thus, the pyrite is likely a result of that alteration. It is probable that the MM Granodiorite has undergone chloritic alteration, distal to mineralization (Propylitic; Sillitoe 2010), and is consistent with a passive role of the MM Granodiorite as host to mineralizing porphyries.

CHAPTER 5: GEOCHEMISTRY

A whole-rock geochemical analysis was done on the MMH deposit in order to gain insight into the primary geochemical signatures of the lithological units within the deposit as well as the entire MMH deposit itself. The elements and compounds that were analyzed for were: MnO, Al₂O₃, Fe₂O₃, MgO, CaO, Na₂O, TiO₂, P₂O₅, Au, Ag, As, Ba, Be, Bi, Br, Cd, Co, Cr, Cs, Cu, Ga, Ge, Hf, Hg, In, Ir, Mo, Nb, Ni, Pb, Rb, S, Sb, Sc, Se, Sn, Nb, Sr, Ta, Th, U, Y, Zr, La, Ce, Pr, Nd, Sm, Eu, Gd, Dy, Tb, Dy, Ho, Yb, Er, Tl, Tm and Lu. This geochemical analysis also allows for comparisons between the MMH deposit and the other Eocene-Oligocene porphyry deposits within the Chuquicamata district and northern Chile.

5.1 Analytical Methods

Samples from drillcore (Table 4.1) were selected as representative by consultation with MMH personnel, and were collected by M. Zentilli at the core repositories of CODELCO Norte in 2010. All samples were cleaned to avoid oxidation or staining, and trimmed using a diamond saw to eliminate excessive veining or alteration. Saw marks were then removed using inert grinding materials on a glass plate, washed with distilled water, dried, and packed in individual plastic bags.

Samples were sent by courier to Activation Laboratories Ltd. in 1336 Sandhill Drive, Ancaster, Ontario, L9G 4V5 Canada. The rocks were all prepared according to Code RX2: crushed and pulverized with mild steel, considered best for low contamination. Analyses were performed following Codes 4E-XRF and 4E-ICP/MS using research grade analytical procedures, plus S was analyzed by infrared methods.

Analytical results are shown in appendix 4. The elements Ga, Pb, Sn, Nb and Rb were analyzed using pressed pellet XRF (X-ray fluorescence) to achieve best detection limits and accuracy. Some elements, such as Rb, have been analyzed by 2 methods (see Fig. 5.1, vertical axis PP-XRF (pressed-pellet X-ray fluorescence) horizontal axis FUS-MS (fusion mass spectrometry). In general the results are very similar, except for values at the low and high end of compositions. In the case of Rb, the method preferred is XRF.

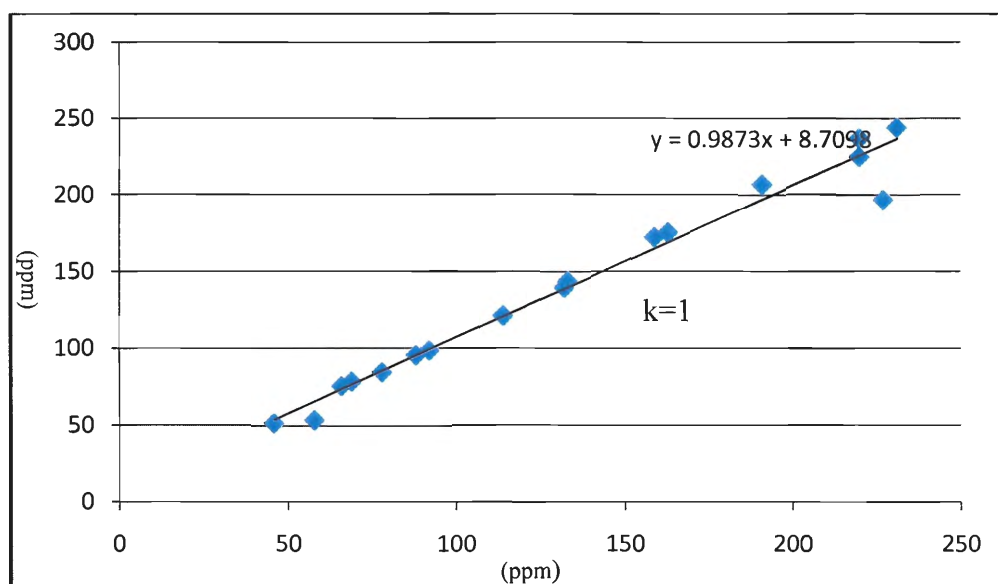


Fig 5.1 This table shows the results of two different analytical methods performed on the same samples for one element. The X-axis shows the FUS-MS results and the Y-axis shows the PP-XRF results. The resulting linear correlation indicates that the two methods gave similar results.

The results obtained for international standards analyzed are indicated in appendix 3 and can be used to assess accuracy, and duplicates were run (ZMMH 15Q and 17Q) to give information on precision (reproducibility).

The following information about the preparation of samples for whole-rock geochemical analysis was taken from the ACTLAB website:

<http://www.actlabs.com/page.aspx?menu=74&app=244&cat1=595&tp=2&lk=no>

SiO₂, Al₂O₃, Fe₂O₃, MgO, MnO, CaO, TiO₂, Na₂O, K₂O, P₂O₅, Ba, Be, Bi, Cs, Ga, Ge, Hf, In, Mo, Nb, Rb, Sn, Sr, Ta, Th, U, V, Y, Zr, La, Ce, Pr, Nd, Sm, Eu, Gd, Tb, Dy, Ho, Er, Tl, Tm, Yb and Lu were analyzed by the MS-ICP/FUS-MS method. Samples were prepared in batches and mixed with a flux of lithium metaborate and lithium tetraborate. The batches were then fused in an induction furnace. Afterwards, the melt was poured into a solution of 5% nitric acid and mixed until dissolved. Desired major oxides and trace elements were tested for using a Thermo Jarrell-Ash ENVIRO II ICP (www.actlabs.com).

Cd, Cu, Ni, Pb, S and Zn were tested for using Total Digestion ICP (TD-ICP). A 0.25 g aliquot of the sample was digested with HClO₄, HNO₃, HCl and HF at 200°C until fuming and was then diluted with aqua regia. This method is used for elements contained in magnetite, chromite, barite, spinels and massive sulphides (www.actlabs.com).

Au, Ag, As, Br, Co, Cr, Hg, Ir, Sb, Sc, Se, W and Zn were tested for using instrumental neutron activation analysis (INAA). Samples were prepared using 1 g aliquots encapsulated in a polyethylene vial. The samples were irradiated with flux wires at a thermal neutron flux and then left for 7 days to allow for the Na-24 to decay. The decay-corrected activities were then compared to a certified international reference calibration (www.actlabs.com).

Ga, Pb, Sn, Nb and Rb were tested for using PP-XRF. Analyses were done on 6 g pressed powder pellets from each sample. The mass absorption coefficient was found by measuring the Compton scatter of the Rh-tube and corrections were made for the background and mass absorption (www.actlabs.com).

5.2 Major Elements

The SiO_2 values for the MMH deposit lie between 55 and 79 wt%. Generally, the MM Porphyry is less felsic than the MM Quartz Porphyry. Using classification from Rickwood (1989) and Le Maitre et al. (1989) in figure 5.2, rock compositions for the MMH deposit plot in the medium to high K calc alkaline field (Fig. 5.2). Three samples from the MM Porphyry and one sample from each of the MM Quartz Porphyry and MM Granodiorite plot in the shoshonite series field.

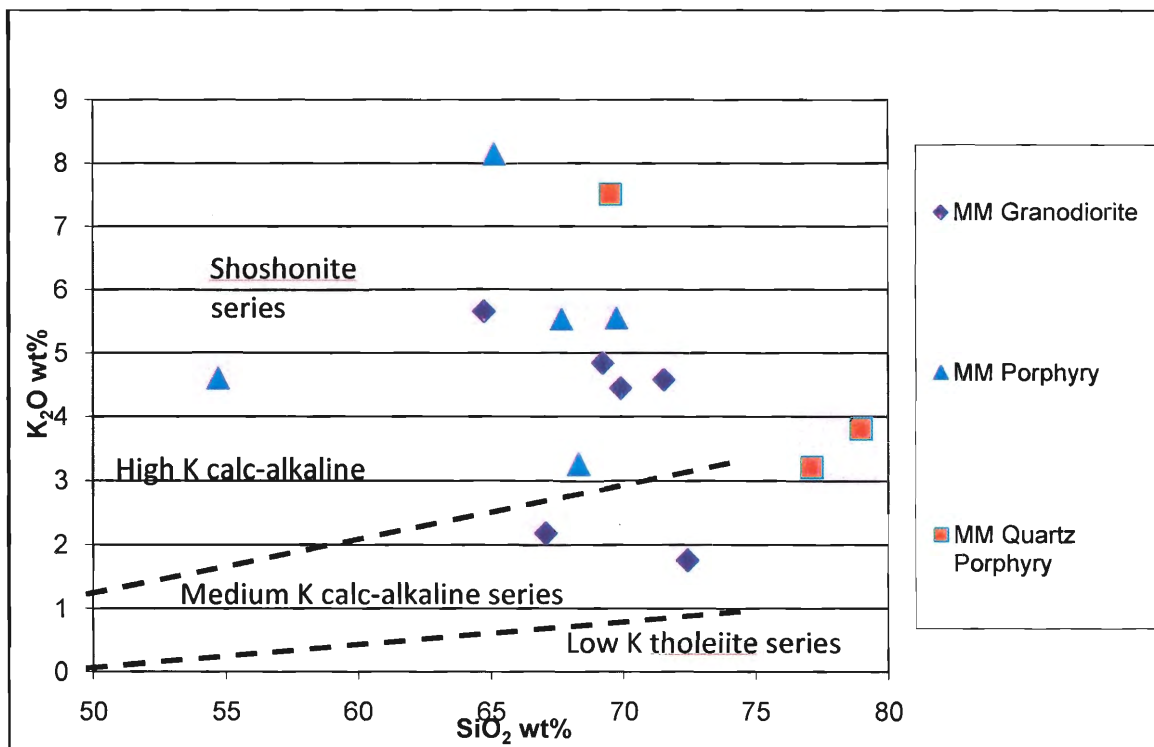
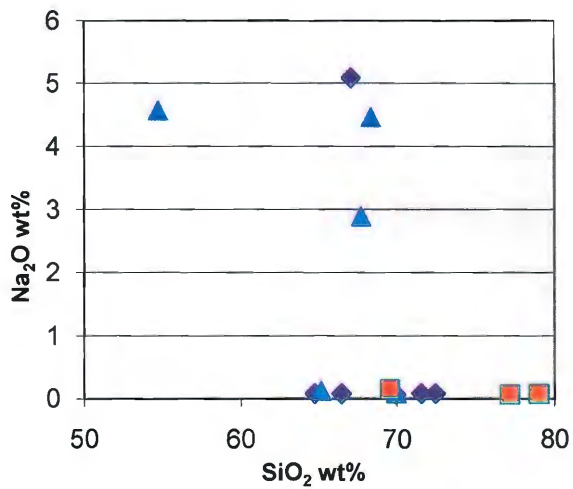
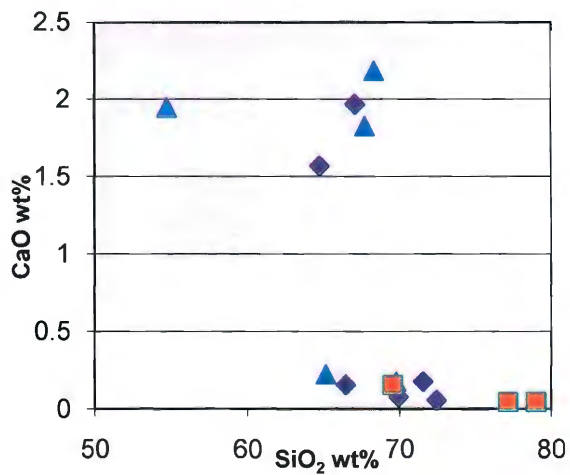
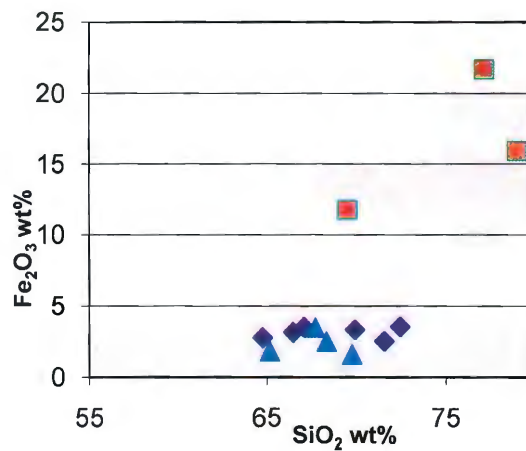
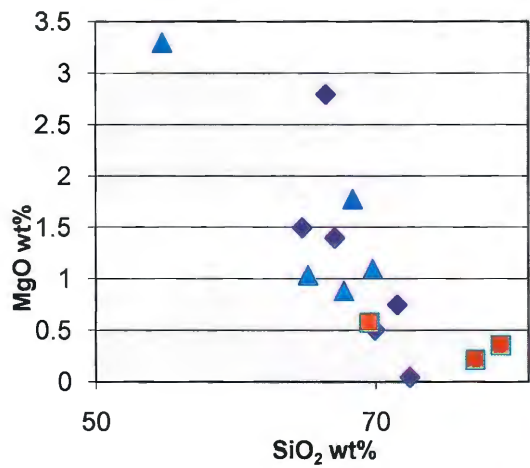
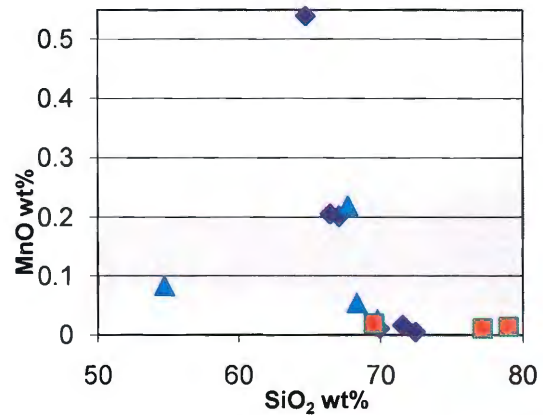
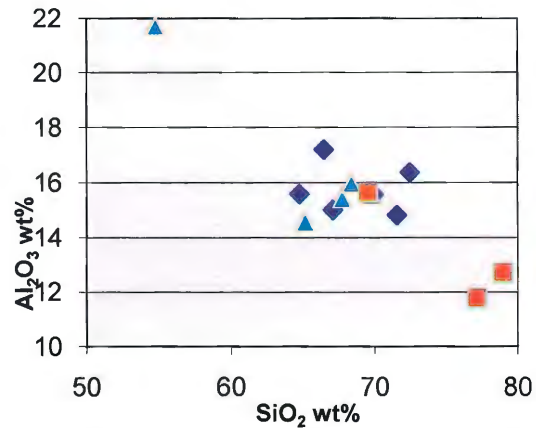


Figure 5.2 Classification of samples from MMH using Le Maitre (1989) classification for high K calc-alkaline series, medium K calc-alkaline series and low K tholeiite series as well as Rickwood (1989) classification for shoshonite series. The high K_2O values are most likely due to potassic alteration.

SiO_2 content vs. major element plots are shown in Fig. 5.3. These diagrams show that Al_2O_3 , MnO , MgO and P_2O_5 have a general trend of decreasing with increasing SiO_2 . This type of trend is indicative of major element fractionation into solid phases during the crystallization process. Deviations from a straight-line curve are likely due to secondary alteration processes that have affected concentrations of the relatively mobile elements.



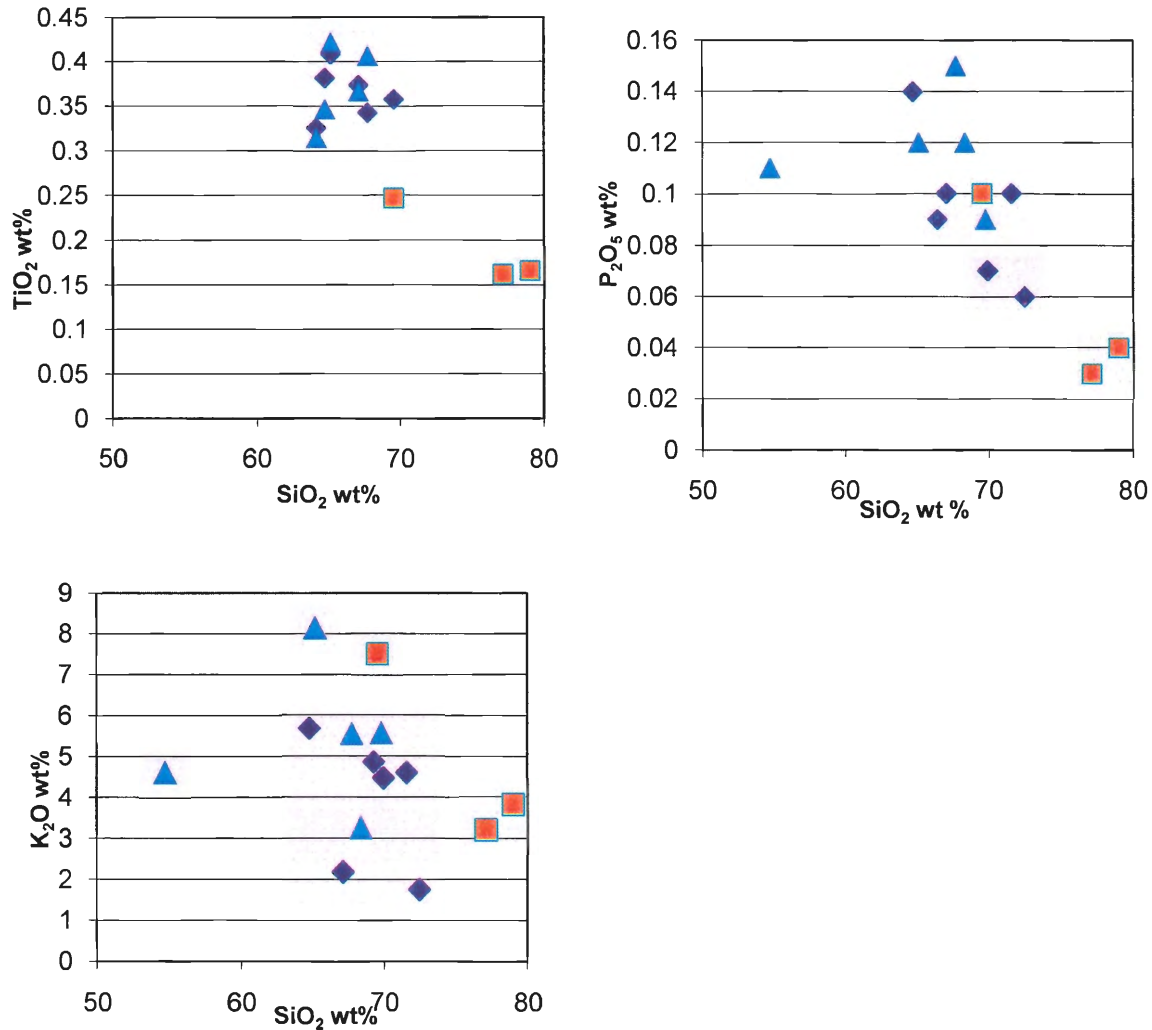


Fig 5.3 Variation diagrams of major elements plotted for MM Granodiorite, MM Porphyry and MM Quartz Porphyry. Legend same as Fig. 5.2

The alkali-oxide (Na_2O and K_2O) vs. SiO_2 and CaO vs. SiO_2 show irregular trends.

This is likely due to redistribution of these elements during the highly acidic hydrothermal alteration processes that affected the MMH deposit such as potassic and phyllic alteration.

Like the Fe_2O_3 vs. SiO_2 plot, the TiO_2 vs. SiO_2 plot shows that the MM Granodiorite unit and the MM Porphyry fall into almost identical geochemical fields. Ti is a very immobile element and thus, the nearly identical geochemical field of MM

Granodiorite and MM Porphyry is indicative that the two units formed under similar geochemical conditions.

The SiO_2 vs. Fe_2O_3 relationship is similar for MM Granodiorite and MM Porphyry samples whereas the MM Quartz Porphyry has a much higher content of Fe.

5.3 Trace Element Geochemistry

Trace element geochemistry is more effective in evaluating the primary igneous traits of MMH than major element geochemistry. This is because the trace elements are less likely to be affected by hydrothermal alteration processes. Samples from MM Porphyry and MM Quartz Porphyry were plotted on a Winchester & Floyd (1977) volcanic/sub-volcanic classification diagram (Fig. 5.4) to determine which rock types they are classified as. This classification diagram uses relatively immobile trace elements (Zr, TiO_2 , Nb and Y) which are useful for classification of rocks which have undergone hydrothermal alteration. From Fig. 5.4 we can see that the MM Porphyry (PMM) and the MM Quartz Porphyry (POQ MM) fall into the andesite and rhyodacite fields. The MM Granodiorite on the Winchester & Floyd (1977) diagram plots in the andesite field. It is not clear why two samples of Grd MM plot in the alkaline field.

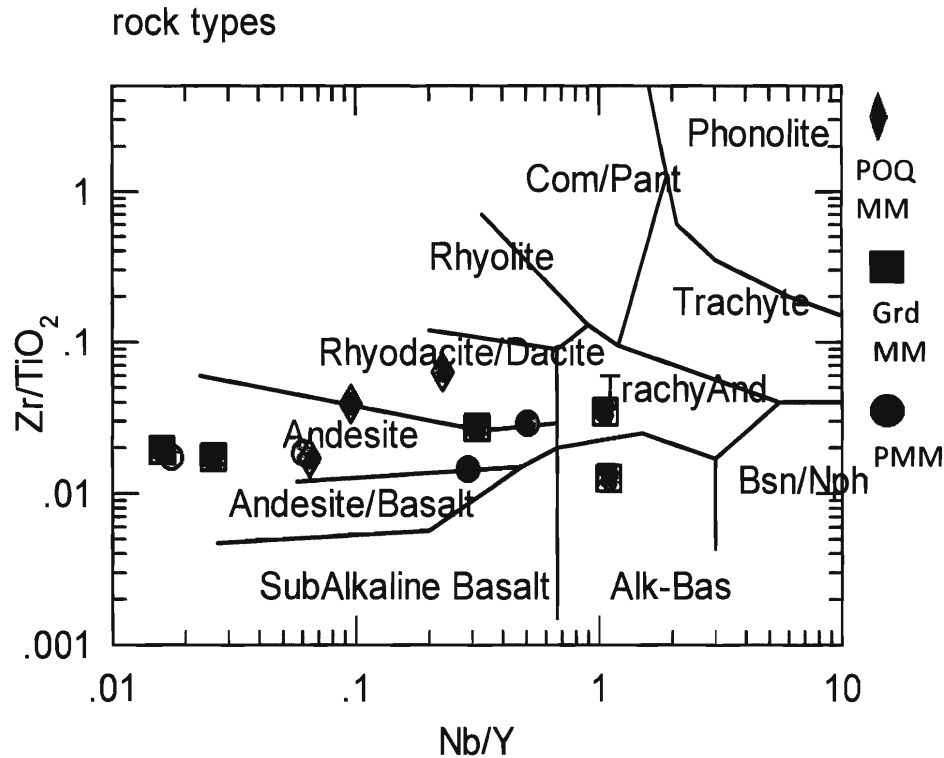


Fig 5.4 Discrimination diagram for volcanic rocks after Winchester & Floyd (1977) was used to classify the sub-volcanic MM Porphyry (PMM) and MM Quartz Porphyry (POQ MM) units of the MMH deposit. The MM granodiorite was also included in this classification diagram (Grd MM)

The rock/chondrite diagram in Figure 5.5 shows concentrations of trace element relative to the primitive mantle source. Figure 5.5 shows that the MM Porphyry, the MM Granodiorite and the MM Quartz Porphyry have almost identical enrichment patterns. This indicates that the units formed under almost identical geochemical conditions with respect to trace elements. This is indicative that the units have originated from the same parental magma source.

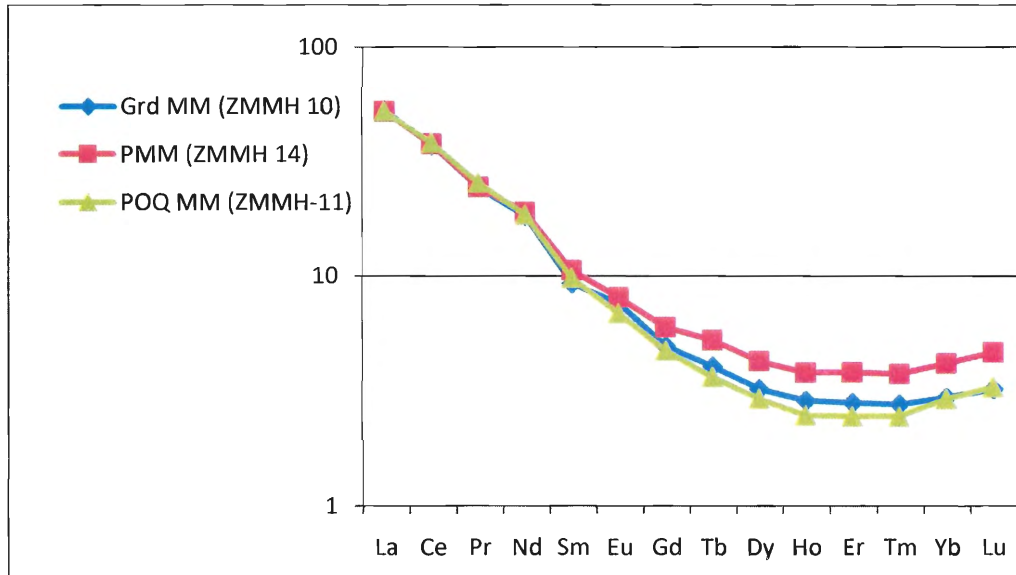


Figure 5.5 Chondrite-normalized REE plot shows enrichment in incompatible REE. Sample ZMMH-10 was used to represent the MM Granodiorite, ZMMH-14 was used to represent the MM Porphyry and ZMMH-11 was used to represent the MM Quartz Porphyry. The curves suggest the both the Triassic and Eocene magmas experienced hornblende fractionation.

Adakites were discussed in Chapter 2. As was mentioned, adakitic geochemical signatures, especially high Sr/Y and La/Yb ratios (≥ 20) are important in exploration as many of the world's major porphyry Cu deposits have adakitic composition. As shown in table 5.1, many of the Sr/Y and La/Yb ratios from the various lithologies in MMH do not fully meet the adakitic classification levels. Within each lithology, less than half of the samples have both the Sr/Y ratio and La/Yb ratio ≥ 20

Sample	Lithology	Sr/Y	La/Yb
ZMMH 05Q	Grd MM	15	38
ZMMH 23Q	Grd MM	1	17
ZMMH 17Q	Grd MM	4	16
ZMMH 18Q	Grd MM	31	20
ZMMH 20Q	Grd MM	19	21
ZMMH 10Q	Grd MM	41	27
ZMMH 07Q	PMM	41	33
ZMMH 09Q	PMM	15	19
ZMMH 15Q	PMM	29	23
ZMMH 21Q	PMM	7	23
ZMMH 14Q	PMM	46	19
ZMMH 11Q	POQ MM	26	27
ZMMH 12Q	POQ MM	8	19
ZMMH 13Q	POQ MM	8	23

Table 5.1 The values of Sr/Y and La/Yb for each individual sample from MMH. The shaded boxes indicate ratio values that do not fall into the adakitic geochemical signature

Figure 5.6 is a classification diagram by Baldwin & Pearce (1982) for productive vs. non-productive porphyries. Through an empirical study they found that porphyries with low Mn and Y were more likely to be productive than those with higher values. The application of this diagram to data from MMH indicates that the data from the MM Quartz Porphyry falls into the productive field, most of the data from the MM Porphyry falls into the sub-productive field and the data from the MM Granodiorite is split between the productive field and the non-productive field.

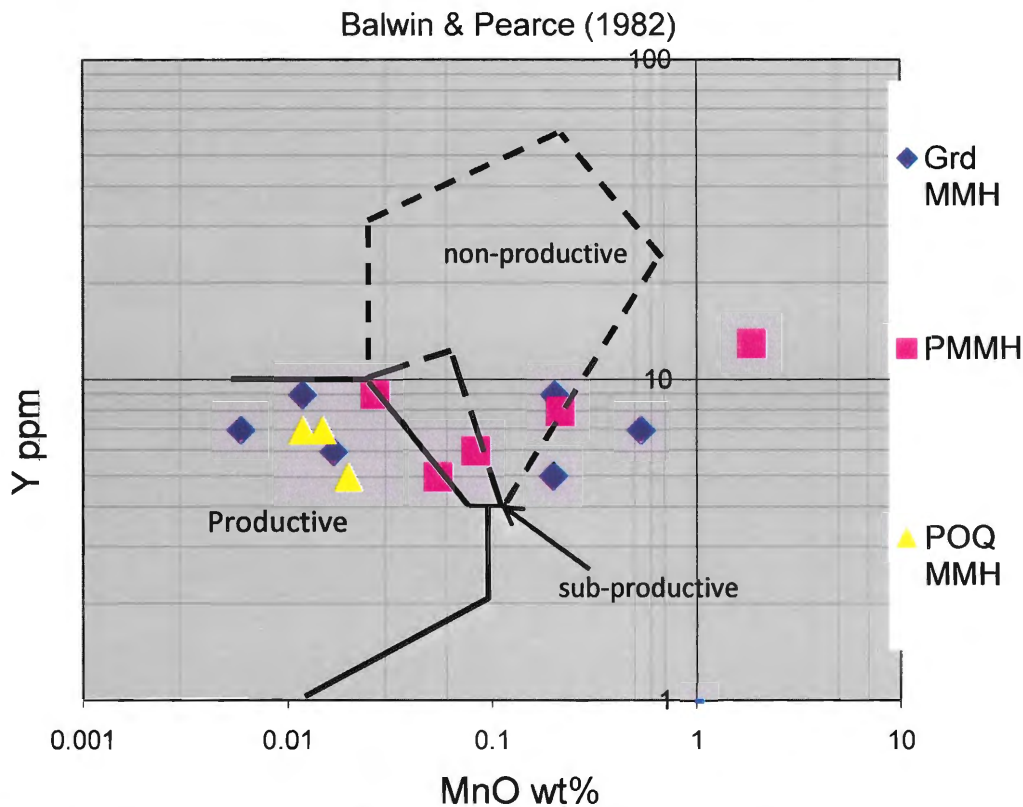


Figure 5.6 Baldwin and Pearce (1982) classification diagram of productive vs. non-productive copper porphyries.

X-Y plots of trace elements vs. SiO_2 were used for comparison of the trace element values within the three lithologic units at MMH (Fig. 5.7). Figure 5.8 is a Zr vs. TiO_2 plot for MM Granodiorite, MM Porphyry and MM Quartz Porphyry. Zr and TiO_2 were used in this plot as they are both incompatible elements which are highly resistant to the type of highly acidic hydrothermal alteration that these samples have been subjected to. The findings from this comparison show that the values from MM Granodiorite and MM Porphyry fall into the same geochemical fields. In the trace element vs. SiO_2 plots, the MM Quartz Porphyry has similar trace element values as the MM Granodiorite and the MM Porphyry however, two of the samples have a higher proportion of SiO_2 . This is expected as it was observed earlier that the MM Quartz Porphyry is much more felsic than the other two units and thus, has a higher proportion of quartz and lesser values of

most other minerals that would contain trace elements. In the Zr vs. TiO_2 plot, the MM Quartz Porphyry plots in a separate field that MM Granodiorite and MM Porphyry (Fig. 5.8). Again, this is likely due to the more felsic nature of the unit.

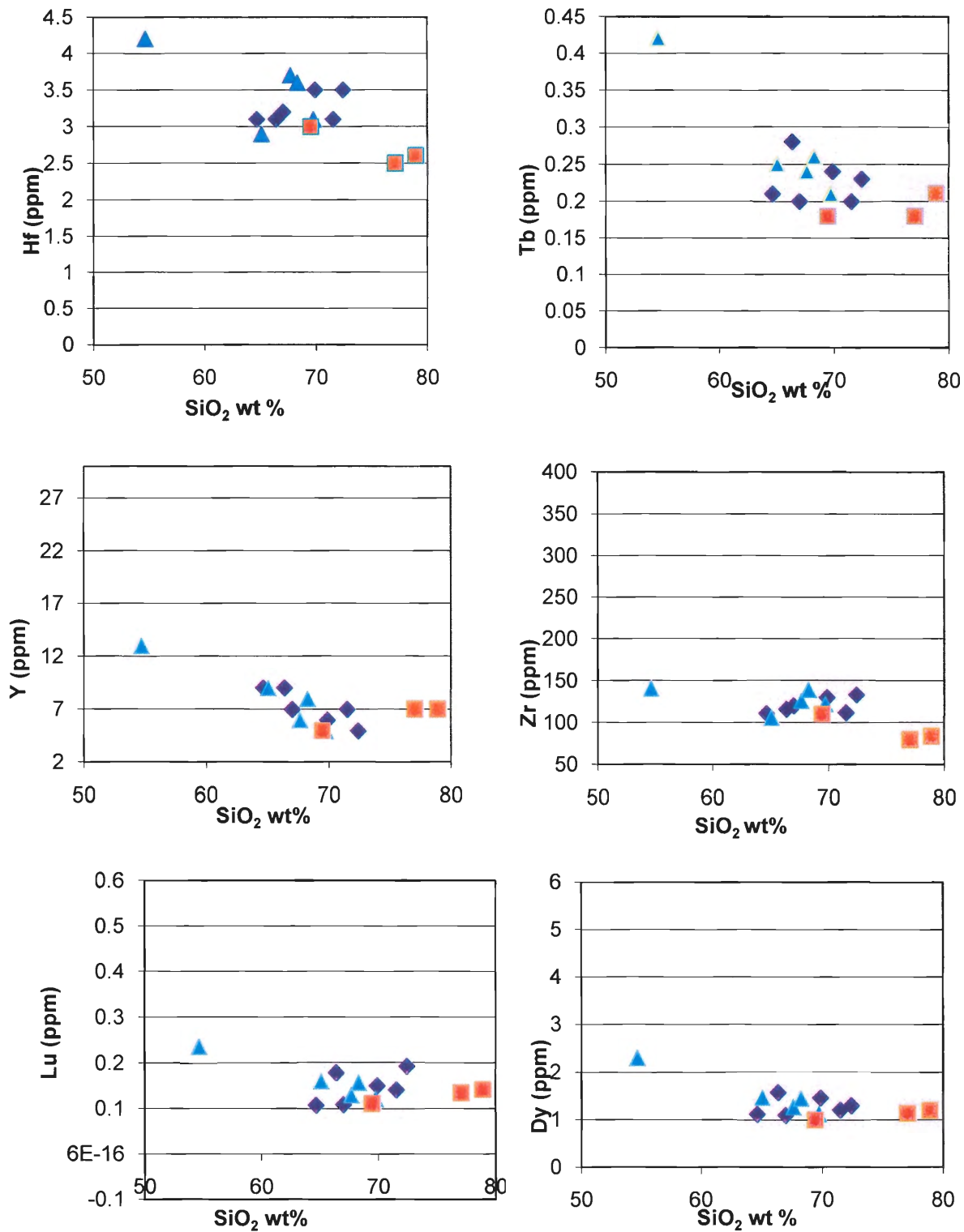


Figure 5.7 SiO₂ wt% vs. trace elements plot containing values from MM Granodiorite, MM Porphyry and MM Quartz Porphyry. The legend for these diagrams are the same as Fig. 5.3 and 5.8

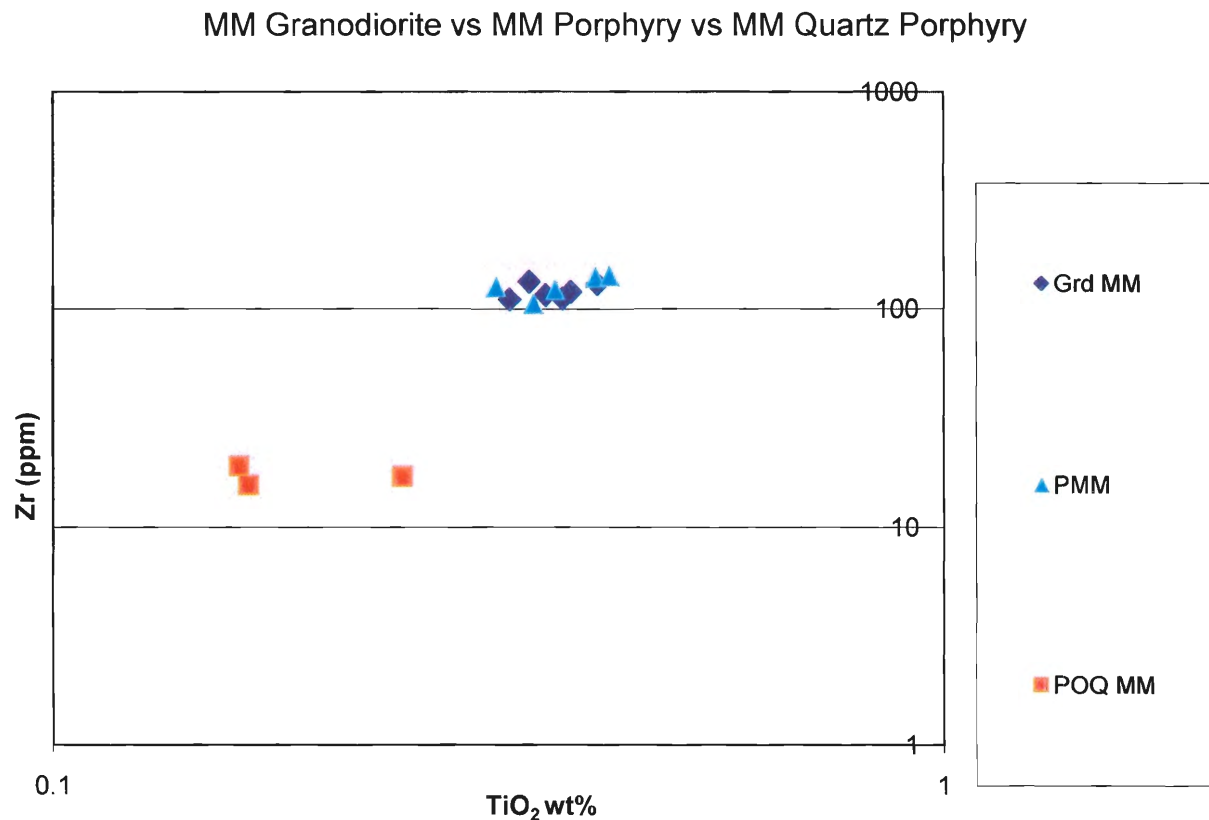


Figure 5.8 Comparison of the Zr and TiO₂ values between the MM Granodiorite (Grd MM), MM Porphyry (PMM) and the MM Quartz Porphyry (POQ MM). To emphasize the spatial arrangement of the data values, the X and Y axis of this plot are log-scale.

5.4 Comparisons with other Intrusive Complexes in the Region

Geochemical comparisons between MMH, Chuquicamata, El Abra and Fortuna were done by comparing the data from various units within the deposits in trace element geochemical plots.

Figure 5.9 compares the geochemistry of intrusive units within the Chuquicamata Intrusive Complex (CIC) with the units of MMH. The points denoted EsteK, EsteSer and Este 2 were all taken from the Este Porphyry (see Chapter 3). For the samples Banco 2 and Este 2 the whole rock geochemistry was done together with the MMH samples in this study. The rest of the data from Chuquicamata is from from Arnott (2003). Ester (a

representative sample of Este Porphyry), Este 2 (ditto.), Este K (potassically altered Este porphyry) and Banco2 plot within the same compositional field as data from MMH. EsteSer (Este Porphyry in the Q-Ser alteration zone), Banco and Fiesta plot outside of the MMH compositional field.

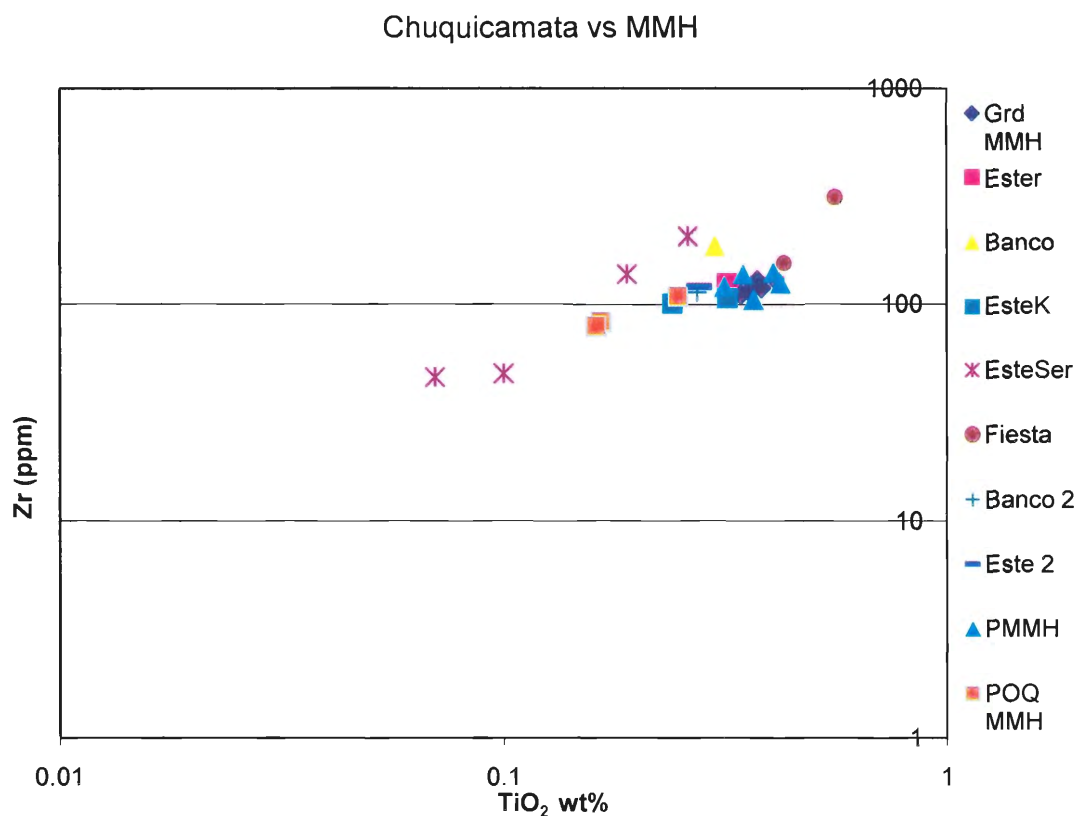


Figure 5.9 Comparison of intrusive rocks from the Chuquicamata mine with the MMH deposit using a TiO_2 vs. Zr diagram. Ester, Banco, EsteK, EsteSer, Fiesta, Banco 2 and Este 2 are all part of the Chuquicamata Intrusive Complex (CIC). Data from the CIC comes from Arnett (2003) except for Banco 2 and Este 2 which were analyzed along with the MMH samples.

Figure 5.10 compares the geochemistry of units from within the El Abra Intrusive Complex to those from MMH. The Zr and TiO_2 values from the El Abra Apolo Leucogranite and the El Abra Mine Porphyry show a strong correlation with those from the MM Granodiorite and the MM Porphyry. Values from the El Abra Aplite have similar TiO_2 values but show Zr values higher than the rest of the units.

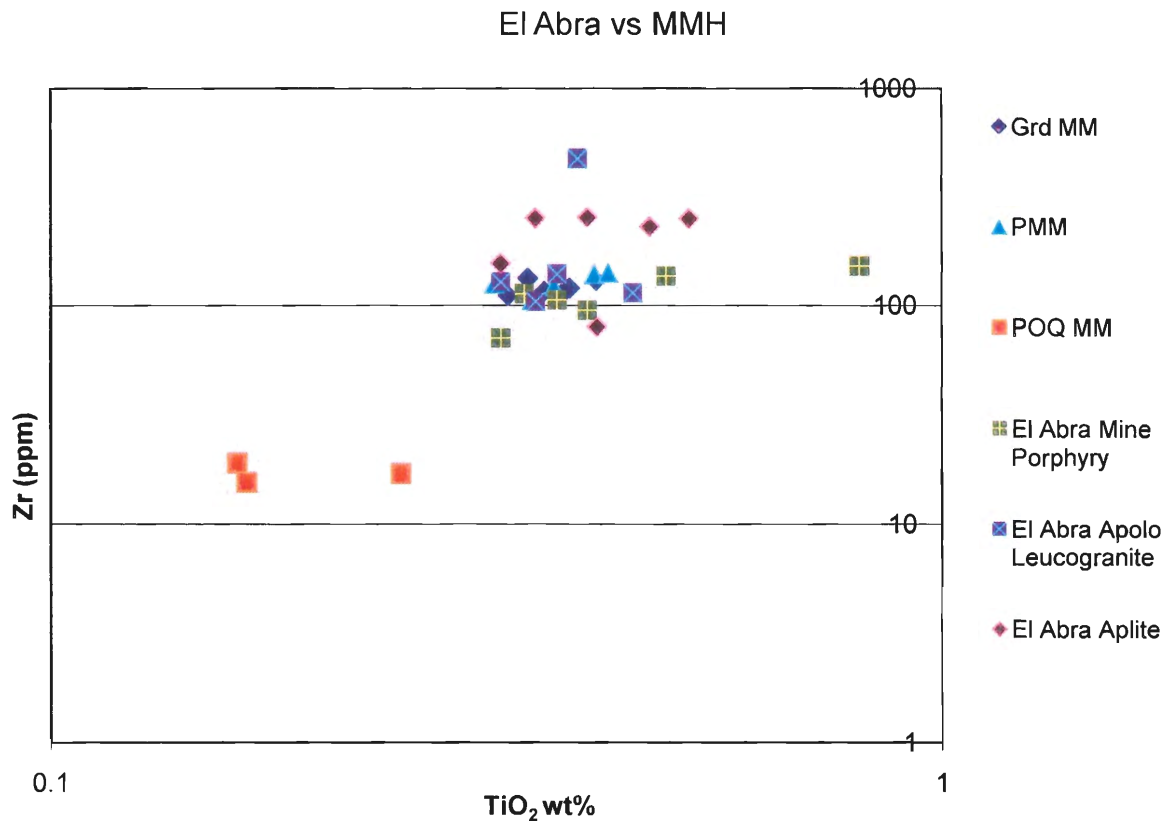


Figure 5.10 Comparison of intrusive rocks from the El Abra Intrusive Complex and MMH using a Zr vs. TiO₂ plot. Data for El Abra from Ballard (2001).

Fig 5.11 compares the geochemistry of San Lorenzo Porphyry and the Fortuna Leucocratic Intrusion to that of the lithologic units within MMH. The Zr and TiO₂ values from the San Lorenzo Porphyry show a strong correlation with those from the MM Granodiorite and MM Porphyry. The geochemical field from the Fortuna Leucocratic Intrusion encompasses the MM Granodiorite and MM Porphyry but has much more variability in terms of TiO₂ values.

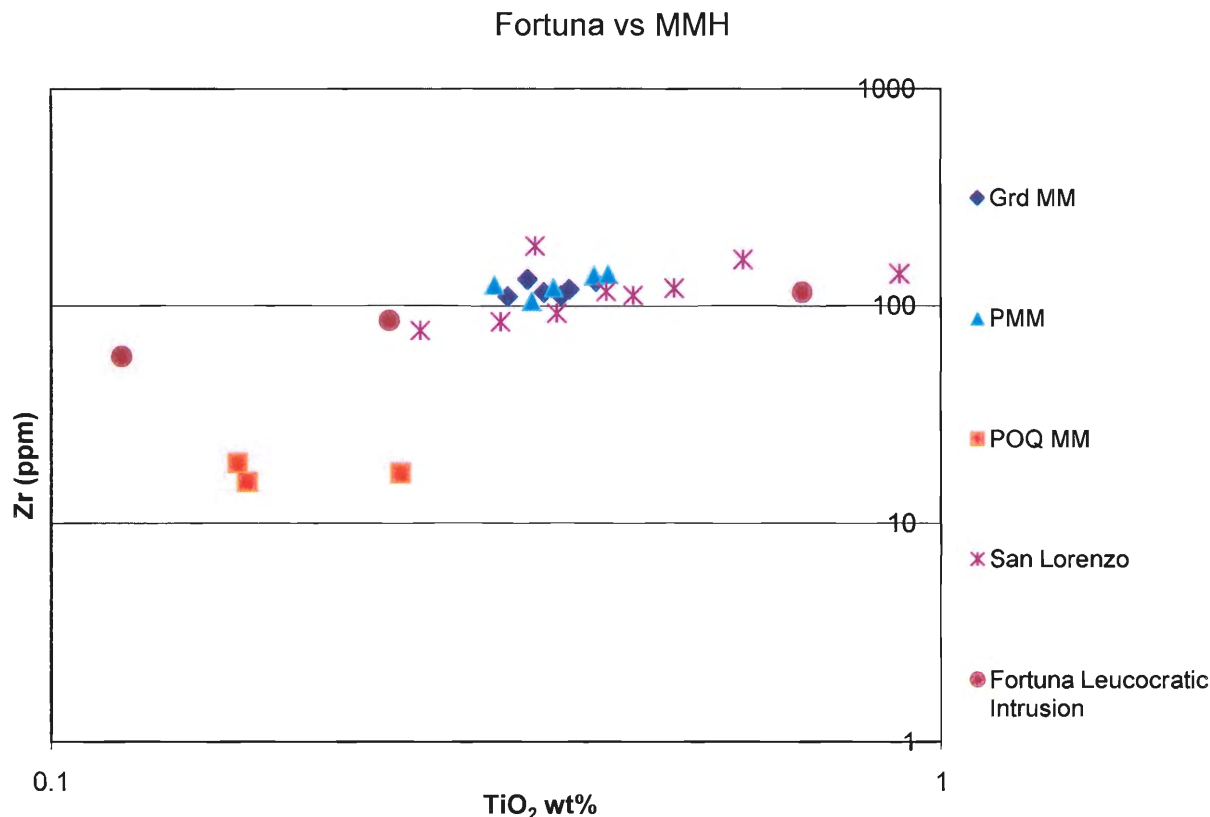


Figure 5.11 Comparison between the Fortuna Igneous Complex and MMH on a Zr vs. TiO₂ plot. Data from Fortuna is from Ballard (2001).

5.5 Discussion

The major element geochemical data from MMH is less a reflection of igneous geochemical character and more a reflection of hydrothermal alteration processes. Taking this into consideration, the Le Maitre (1989) and Rickwood (1989) classification diagram shown in Fig. 5.2 may not indicate that many of the samples are shoshonite/high K calc-alkaline rocks but that these samples underwent potassic alteration. While the SiO₂ wt% vs. major element vaguely follow the fractional crystallization model, deviations from that trend are indicative of hydrothermal alteration.

The classification of the samples from MMH as andesite and rhyodacite is typical of sub-volcanic rocks within this region of the Andes. On the topic of adakite classification, the conclusion is not as clear. The results show that within each lithologic

unit, less than half of samples have both Sr/Y and La/Yb ratios that meet the adakitic classification level (≥ 20). Considering, however, that some of the ratios do meet the adakite standard, it is fair to say that the composition of Triassic MM Granodiorite, and the Eocene MM Porphyry and MM Quartz Porphyry are similar to adakites or sub-adakitic.

The SiO₂ wt % vs. trace element diagrams all indicate that the MM Granodiorite and the MM Porphyry have very similar trace element geochemistry. This is further supported with similar results in the Zr vs. TiO₂ diagram (Fig. 5.8), the TiO₂ vs. SiO₂ and Fe₂O₃ vs. SiO₂ (Fig. 5.3) and the chondrite-normalized diagram (Fig. 5.5). Assuming that the immobile elements used in these comparisons are reflective of the geochemical conditions at crystallization, these findings indicate that there were geochemical and possibly tectonic similarities in the genesis of both the MM Granodiorite and the MM Porphyry. Strengthening this argument is the findings that half of the MM Granodiorite measurements plot in the productive field of the Baldwin & Pearce (1982) diagram (Fig. 5.6). This finding implies a connection between productive copper porphyry deposits and MM Granodiorite.

The result from the geochemical comparison between MMH and the Chuquicamata Intrusive Complex (CIC) is uncertain. Although the two units seem to be similar in their geochemistry, the results from the data obtained from Arnott (2003) as well as this study are not conclusive. For example, although the data from Este 2 and Este K plot within close range of the MMH units, EsteSer, which belongs to the same lithologic unit as the aforementioned, plots in its own distinguishable geochemical field; this result probably reflects the intense high sulphidation alteration (and hydration)

produced by the quartz-sericite phase in Chuquicamata (Ossandón et al. 2001). Because the Este Porphyry is established as the largest contributor of ore within the CIC, this unit is the most essential in terms of finding a geochemical match. The correlation between MMH and the Banco porphyry is inconclusive as one of the data points fits in the MMH geochemical field (Banco) whereas the other does not (Banco 2). Overall, the findings from this comparison indicate that the CIC and MMH lithological units are geochemically similar but it does not confirm a relationship.

The comparison between intrusive units from El Abra and MMH show that the El Abra Mine Porphyry and the El Abra Leucogranite plot in similar geochemical fields in the Zr vs. TiO_2 diagram as the MM Porphyry and the MM Granodiorite. The El Abra Aplite porphyry has similar TiO_2 values as the MM Porphyry and MM Granodiorite but higher Zr values. As discussed earlier in section 5.4, geochemical comparisons using Zr and TiO_2 are reflective of conditions during crystallization as Zr and Ti are both immobile elements. The correlation of the El Abra Mine Porphyry and the MM Porphyry is significant because both these units are considered to be the major contributors of ore in their respective deposits. This similarity indicates that El Abra and MMH formed under similar geochemical and possibly tectonic conditions.

As is the case with El Abra, the most prominent porphyry unit within the Fortuna Igneous Complex, the San Lorenzo Porphyry, falls into the same geochemical field as the MM Porphyry and MM Granodiorite. The geochemical field of the Fortuna Leucocratic Intrusion encompasses the data from the MM Porphyry and MM Granodiorite but is much more variable in its proportion of TiO_2 . This comparison between the Fortuna

Igneous Complex and MMH indicates that the two units formed under similar geochemical and possibly tectonic conditions.

CHAPTER 6: DISCUSSION AND CONCLUSIONS

Whole rock geochemical analysis and petrographic analysis have proven effective methods of evaluating the petrology of the MMH deposit. The trace element geochemical analysis has given insight into the igneous geochemical conditions that existed before the hydrothermal alteration overprint. This analysis allows for comparison of lithologic units within the MMH deposit as well as comparisons between MMH and other deposits in the region. Due to the tectonic implications of the West Fault (ie truncation of major ore deposits), these geochemical comparisons can be used as a tool to correlate deposits that may have been cogenetic. The major element geochemical analysis, the EMP analysis as well as the optical microscopy have been instructive on both the primary igneous conditions of the deposit as well as the subsequent hydrothermal alteration processes.

A trace element geochemical comparison was done between the lithologic units within the MMH deposit to see if the units had distinct geochemical signatures. The results showed that while the MM Quartz Porphyry has a different compositional field from the MM Porphyry and the MM Granodiorite. These two units overlapped and thus they are not distinguishable using trace element geochemistry. This finding implies a petrologic and tectonic connection between the two units which is unusual since the MM Granodiorite has been dated as Triassic (237-222 Ma) (Boric et al. 2009) and the MM Porphyry has been dated as 39 Ma (Eocene; Boric et al. 2009). There are two possible explanations for this coincidence: 1) there was an error in the logging of the core and the samples obtained all belong to the MM Granodiorite, and 2) the MM Granodiorite formed under extremely similar petrologic/tectonic conditions as the MM Porphyry .

Logging nomenclature at MMH has evolved over more than 2 decades by very competent geologists, and consists of two stages: a first fast logging preceding splitting and sampling for grade and physical properties, and a second, more careful logging with analytical data on hand. It is however possible that where the largely barren (unmineralized) host MM Granodiorite starts to show extensive veining, alteration and mineralization, the geologists needed to describe a new rock type. Locally, the alteration can develop large feldspar crystals, silicified rock can look like quartz phenocrysts, and the crosscutting veins characteristic of porphyry deposits led to the designation of the rock as porphyry. Thus it is possible that an alteration overprint caused what was primarily MM Granodiorite to be called MM Porphyry. Although this scenario is plausible, the petrographic evaluation conducted on the MMH samples (Chapter 4) indicate that the MM Porphyry and the MM Granodiorite are distinctive rock types with independent primary igneous features. Definite evidence is the drastic difference between the preserved calcic plagioclase (oligoclase-andesine) in MM Porphyry, compared to pure albite in MM Granodiorite. With this consideration, it is implausible that an error occurred in the logging.

In the very likely case that the samples obtained are true to the lithologic units they have been ascribed, the findings of this study imply that the MM Granodiorite formed under almost identical geochemical and tectonic conditions ~200 million years before MM Porphyry in the Triassic. The geochemical similarities between the MM Porphyry and the MM Granodiorite indicate that the MM Granodiorite could have shared a parental source at depth. As is suggested by the Baldwin & Pearce (1982) diagram, the common parental magma source may have been enriched in copper, or the petrogenetic

conditions to form both magma suites were very similar. This implies that productive copper porphyries may have been produced in the Triassic. In this respect it is interesting to note that the Permo-Triassic basement contains metallic mineralization of Triassic age elsewhere in the region (e.g. Makshev 1990; Camus 2003).

As stated in section 5.5, the results from this study show a relationship between MMH and Chuquicamata but do not indicate a cogenetic relationship. In following with the hypothesis of 35 km of sinistral displacement along the west fault, the location of MMH and Chuquicamata relative to each other indicate that MMH is not a “missing portion” of Chuquicamata. Furthermore, the earliest date for copper porphyry intrusion at Chuquicamata is the Este Porphyry at 34 Ma (see table 3.1), 5 My later than the emplacement of the MM Porphyry. The date of 34 Ma does however compare to the age of the MM Quartz Porphyry (Boric et al. 2009). This date relationship in combination with the geochemical correlation between Chuquicamata and the MM Quartz Porphyry indicates that these two units could have both been related to the same magmatic event and both may share a parental source at depth. This analogy has been used to explain the Late Oligocene-Early Eocene Cu-porphyry mineralization phenomena in the northern Chilean Andes (e.g. Sillitoe 2010).

Figures 5.10 and 5.11 show that geochemical fields of the porphyry units from MMH, Fortuna and El Abra overlap. Dilles et al. (1997) argue that El Abra and Fortuna are one cogenetic igneous complex that intruded 39-37 Ma and was subsequently divided and displaced during the 35 km of post mineralization sinistral displacement along the West Fault. The evidence from this geochemical study strengthens this argument and also indicates that MMH was part of this igneous complex. As discussed in chapter 3.2,

the MM Porphyry has been dated as 39 Ma (Boric et al. 2009). As seen in Fig. 1.1, the MMH deposit is very elongate. Although the main concentration of ore is in the southern region of the deposit, away from Fortuna, the MMH igneous complex extends north within close proximity to Fortuna. The age and location of the deposit together with the geochemical evidence from this study indicate the MMH could be closely linked to the Fortuna-El Abra cogenetic igneous complex, possibly even cogenetic.

The discovery that the MM Granodiorite of Triassic age (e.g. Boric et al. 2009) has characteristics indistinguishable from the Fortuna and El Abra igneous suite of Eocene age, leads to the conclusion that there might be something more than a simple spatial coincidence, and that the MM Granodiorite is not merely a passive host to the Tertiary porphyries. The question that arises is: what tectonic/petrologic conditions could have led to the generation of identical magmas in the Triassic and in the Eocene? The answer could have implications to possibly explain that both magma suites have a common source, either in the crust or mantle. A full discussion of this issue is out of the scope of this thesis, yet some analysis follows.

The existence of giant porphyry copper deposits in the Central Andes in northern Chile has traditionally been associated with the “Andean” orogenic cycle (e.g. Maksaeu 1990; Sillitoe 2010), and “Andean” has generally implied Jurassic to Holocene, ca. 200 My, of which only the Eocene-Oligocene generated giant ore deposits of the likes of Chuquicamata and Escondida. However, copper and molybdenum mineralization, mostly in non-economic concentrations, is known to occur in the pre-Jurassic basement of the Domeyko Cordillera, and even CODELCO, the company that owns Chuquicamata and MMH has developed prospects of Triassic and Carboniferous age (e.g. Marquardt et al.

1997). Several workers, among them Cornejo et al. (1997), have pointed out that directly south of Chuquicamata District and the town of Calama in the Sierra Limón Verde (Lat. $22^{\circ}30'$ - $24^{\circ}S$) the metamorphic basement is intruded by a volcano-plutonic complex of Permo-Carboniferous age, in turn intruded by Triassic, Cretaceous and Eocene plutons. At the Lila prospect, Cu-Mo mineralization is associated to hydrothermal alteration with K-Ar ages of 213 to 195 Ma (Marquardt et al., 1997; Camus, 2003), thus is similar to the Triassic-Jurassic like MM Granodiorite. North of Chuquicamata, Tomlinson et al., (2001) gives K-Ar ages of hydrothermal alteration in Cu-Mo mineralization in the Loa prospect as 223 ± 5 Ma for sericite and 239 ± 7 Ma for biotite (Triassic). More recently, Munizaga et al., (2006) have published (SHRIMP U-Pb) geochronological data that recognize Permian to Triassic magmatic activity, with the Triassic being associated with minor porphyry type Cu-Mo mineralization in the region of Collahuasi (lat. 20° to $22^{\circ}S$; Fig. 1.2) . This Permo-Triassic magmatism (Collahuasi Group) was generated before the “Andean” cycle, along the western margin of the Gondwana supercontinent, and is correlative with an extensive belt of volcanic and subvolcanic rocks throughout the main Andes of Chile, the Frontal Cordillera of Argentina (Choiyoi Group or Choiyoi Granite-Rhyolite Province), and the Eastern Cordillera of Peru. In fact all large porphyry coppers of Chile, including the Eocene-Oligocene in northern Chile, and the Neogene in central Chile, may have a basement that includes this Choiyoi complex. Zentilli et al. (1988) pointed out the remarkable similarity in lead isotopes of all porphyry copper deposits in Chile that suggested a common source for the magmas. If the same source region for the magmas was tapped in the Triassic and

the Eocene, the remarkable similarity between Triassic and Eocene igneous rocks can be explained.

CHAPTER 7: FUTURE WORK

Although this study indicates the tectonic and compositional similarities between the MM Porphyry and MM Granodiorite as well as the cogenetic relationship between MMH, El Abra and Fortuna, there is still work that could be done to confirm these conclusions. For instance, a more thorough age evaluation of the MM Granodiorite and MM Porphyry would solve the question of whether or not MM Porphyry and MM Granodiorite are in fact distinct lithologic units that formed 250 My apart. Comparisons between MMH, El Abra and Fortuna using other evaluative methods such as isotope geochemistry could add to the hypothesis of cogenetic correlation between the three units. Finally, a more thorough investigation into the characteristics of Triassic plutonic rocks, specifically their potential as parental magma sources to Cu porphyries, could give insights into the Cu anomaly that exists in the northern Chilean Eocene-Oligocene magmatic arc.

As a first step, samples of representative lithologies should be collected and analyzed chemically, and U-Pb dating of zircons should be done from exactly the same rocks. This way, any differences and similarities between Triassic and Eocene rocks can be corroborated.

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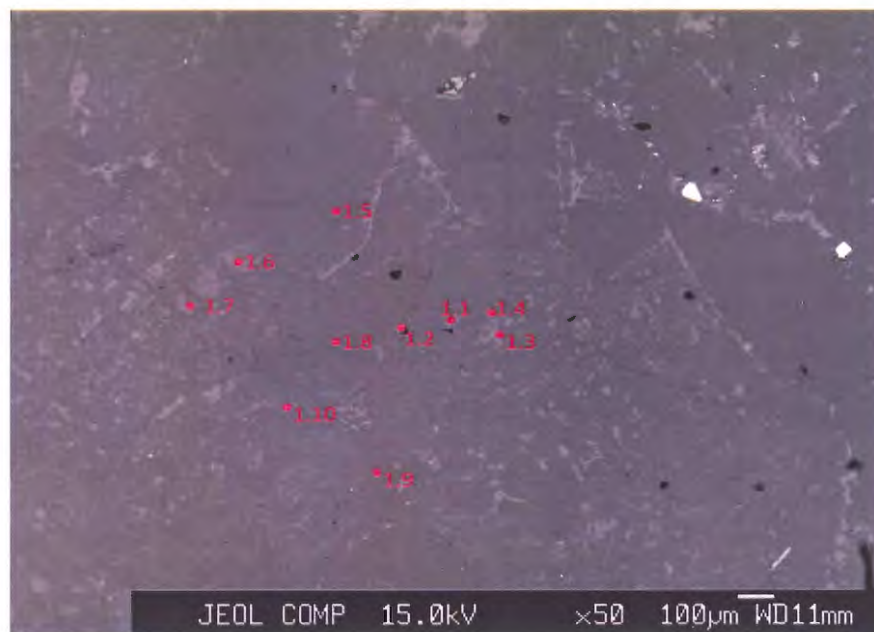
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Appendix 1 EMP Images and Data

This appendix includes EMP data tables from samples ZMMH-10 (MM Granodiorite), ZMMH-11 (MM Quartz Porphyry) and ZMMH-14 (MM Porphyry) (table 4.1) for analysis on both silicates and sulphides. Included with the data are backscatter images of the locations where the points were taken. These images are brighter in locations with high electron density and darker in locations of low electron density. The exact location of each point is marked in red.

ZMMH-10 Silicates (MM Granodiorite)

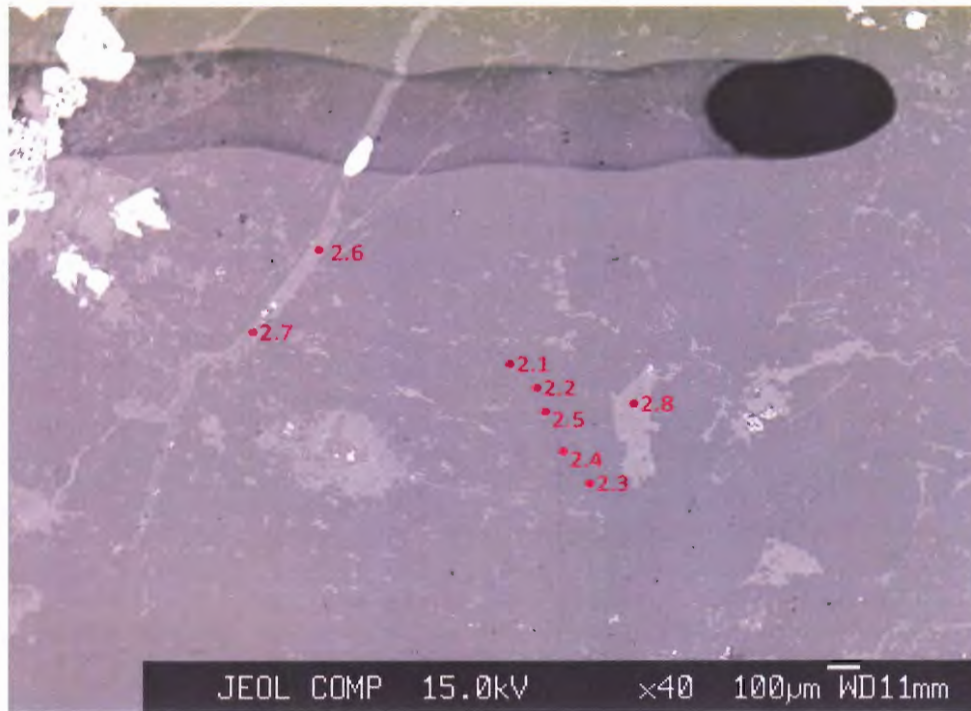


Mass %

No.	K ₂ O	Cr ₂ O ₃	Na ₂ O	Al ₂ O ₃	MnO	CaO	TiO ₂	MgO	SiO ₂	FeO	P ₂ O ₅	Cl	F	NiO	SO ₃	Total	Mineral type
1.1	10.36	0.00	0.27	34.42	0.003	0.01	0.06	0.65	48.11	2.13	0.01	0.00	0.00	0.03	0.03	96.08	K-feldspar
1.2	0.14	0.00	7.75	19.12	0	0.39	0.00	0.00	69.16	0.03	0.00	0.06	0.00	0.01	0.00	96.65	Albite
1.3	10.10	0.03	0.19	33.16	0.012	0.03	0.06	0.95	46.31	2.68	0.00	0.00	0.00	0.04	0.00	93.57	K-feldspar
1.4	0.15	0.00	6.54	19.71	0	0.19	0.00	0.00	71.33	0.03	0.00	0.03	0.00	0.00	0.00	97.97	Albite
1.5	0.37	0.00	8.05	19.13	0	0.18	0.00	0.00	69.80	0.05	0.00	0.02	0.00	0.02	0.00	97.63	Albite
1.6	4.79	0.01	0.09	28.63	0.355	0.06	0.02	9.22	39.00	9.34	0.00	0.01	0.01	0.06	0.02	91.61	K-feldspar
1.7	4.20	0.00	0.08	33.68	0.007	0.09	0.01	1.66	52.07	2.49	0.01	0.00	0.01	0.01	0.01	94.32	K-feldspar
1.8	0.11	0.00	12.17	19.12	0	0.13	0.00	0.00	68.50	0.04	0.00	0.00	0.00	0.02	0.01	100.10	Albite
1.9	0.23	0.00	11.71	19.41	0	0.26	0.00	0.00	68.14	0.10	0.00	0.03	0.00	0.03	0.01	99.91	Albite
1.10	0.16	0.00	10.34	19.38	0	0.41	0.00	0.00	69.61	0.05	0.01	0.00	0.00	0.00	0.00	99.96	Albite

Cation Total

No.	K	Cr	Na	Al	Mn	Ca	Ti	Mg	Si	Fe	P	Cl	F	Ni	S	Total	Mineral type
1.1	0.0792	0	0.0031	0.2431	0	0	0.0003	0.0058	0.2882	0.0107	0.0001	0	0	0.0001	0.0001	0.6308	K-feldspar
1.2	0.001	0	0.0834	0.1251	0	0.0023	0	0	0.3839	0.0001	0	0.0006	0	0.0001	0	0.5966	Albite
1.3	0.0797	0.0001	0.0023	0.2417	0.0001	0.0002	0.0003	0.0087	0.2863	0.0139	0	0	0	0.0002	0	0.6335	K-feldspar
1.4	0.001	0	0.0689	0.1262	0	0.0011	0	0	0.3873	0.0001	0	0.0003	0	0	0	0.585	Albite
1.5	0.0026	0	0.0859	0.1241	0	0.0011	0	0	0.3841	0.0002	0	0.0002	0	0.0001	0	0.5983	Albite
1.6	0.0397	0	0.0011	0.2194	0.002	0.0004	0.0001	0.0894	0.2535	0.0508	0	0.0001	0.0002	0.0003	0.0001	0.6572	K-feldspar
1.7	0.0313	0	0.0009	0.2319	0	0.0006	0	0.0144	0.3042	0.0121	0.0001	0	0.0002	0	0	0.5957	K-feldspar
1.8	0.0008	0	0.129	0.1232	0	0.0008	0	0	0.3746	0.0002	0	0	0	0.0001	0	0.6287	Albite
1.9	0.0016	0	0.1244	0.1254	0	0.0016	0	0	0.3734	0.0005	0	0.0003	0	0.0001	0	0.6274	Albite
1.10	0.0011	0	0.1089	0.1241	0	0.0024	0	0	0.3781	0.0002	0	0	0	0	0	0.6148	Albite

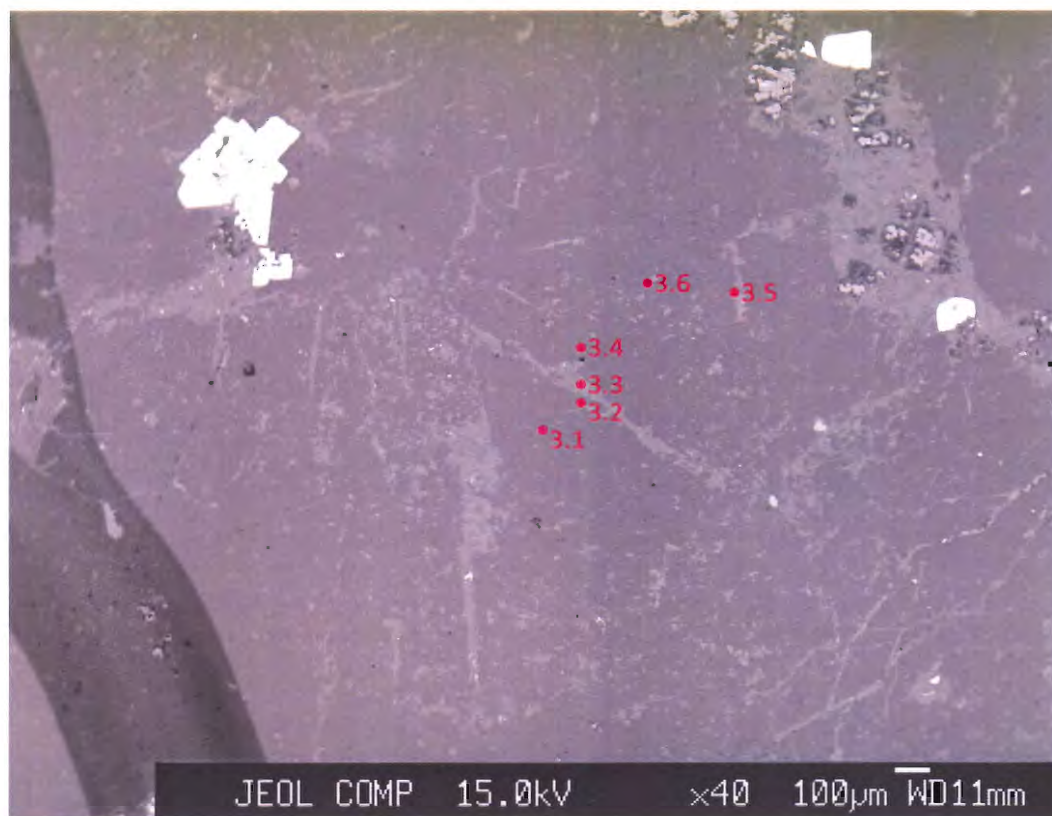


Mass %

	K ₂ O	Cr ₂ O ₃	Na ₂ O	Al ₂ O ₃	MnO	CaO	TiO ₂	MgO	SiO ₂	FeO	P ₂ O ₅	Cl	F	NiO	SO ₃	Total	Mineral type
2.1	5.20	0.00	4.93	24.27	0.007	0.46	0.02	0.97	58.00	1.57	0.03	0.00	0.00	0.01	0.00	95.47	unstable
2.2	0.48	0.00	10.24	19.59	0.005	1.15	0.00	0.00	65.21	0.08	0.00	0.15	0.00	0.01	0.00	96.88	Albite
2.3	0.10	0.00	6.30	19.65	0	0.25	0.00	0.00	70.79	0.08	0.00	0.00	0.00	0.04	0.01	97.20	Albite
2.4	0.64	0.00	10.43	19.33	0	0.54	0.00	0.00	68.18	0.06	0.00	0.00	0.00	0.01	0.01	99.20	Albite
2.5	0.14	0.00	11.99	19.39	0	0.35	0.00	0.00	68.22	0.03	0.00	0.00	0.00	0.02	0.01	100.14	Albite
2.6	16.35	0.01	0.42	17.80	0	0.01	0.01	0.00	64.35	0.03	0.00	0.00	0.00	0.07	0.02	99.07	K-feldspar
2.7	9.06	0.00	0.09	30.20	0.008	0.20	0.06	1.24	50.51	2.46	0.00	0.01	0.06	0.06	0.03	93.97	K-feldspar
2.8	0.02	0.02	0.01	19.24	0.729	0.04	0.08	21.14	27.91	17.11	0.00	0.01	0.02	0.07	0.01	86.40	unstable

Cation Total

No.	K	Cr	Na	Al	Mn	Ca	Ti	Mg	Si	Fe	P	Cl	F	Ni	S	Total	Mineral type
2.1	0.0389	0	0.0561	0.1679	0	0.0029	0.0001	0.0085	0.3404	0.0077	0.0001	0	0	0.0001	0	0.6228	Andesine
2.2	0.0034	0	0.1124	0.1307	0	0.007	0	0	0.3692	0.0004	0	0.0014	0	0.0001	0	0.6247	Albite
2.3	0.0007	0	0.0668	0.1267	0	0.0015	0	0	0.3871	0.0004	0	0	0	0.0002	0	0.5835	Albite
2.4	0.0045	0	0.1113	0.1254	0	0.0032	0	0	0.3752	0.0003	0	0	0	0	0	0.6199	Albite
2.5	0.0009	0	0.1271	0.125	0	0.0021	0	0	0.3731	0.0002	0	0	0	0.0001	0	0.6286	Albite
2.6	0.1219	0	0.0047	0.1226	0	0.0001	0	0	0.376	0.0002	0	0	0	0.0003	0.0001	0.6259	K-feldspar
2.7	0.0702	0	0.0011	0.2162	0	0.0013	0.0003	0.0112	0.3068	0.0125	0	0.0001	0.0011	0.0003	0.0001	0.6213	K-feldspar
2.8	0.0002	0.0001	0.0001	0.1661	0.0045	0.0003	0.0005	0.2308	0.2044	0.1048	0	0.0001	0.0006	0.0004	0	0.7129	unstable

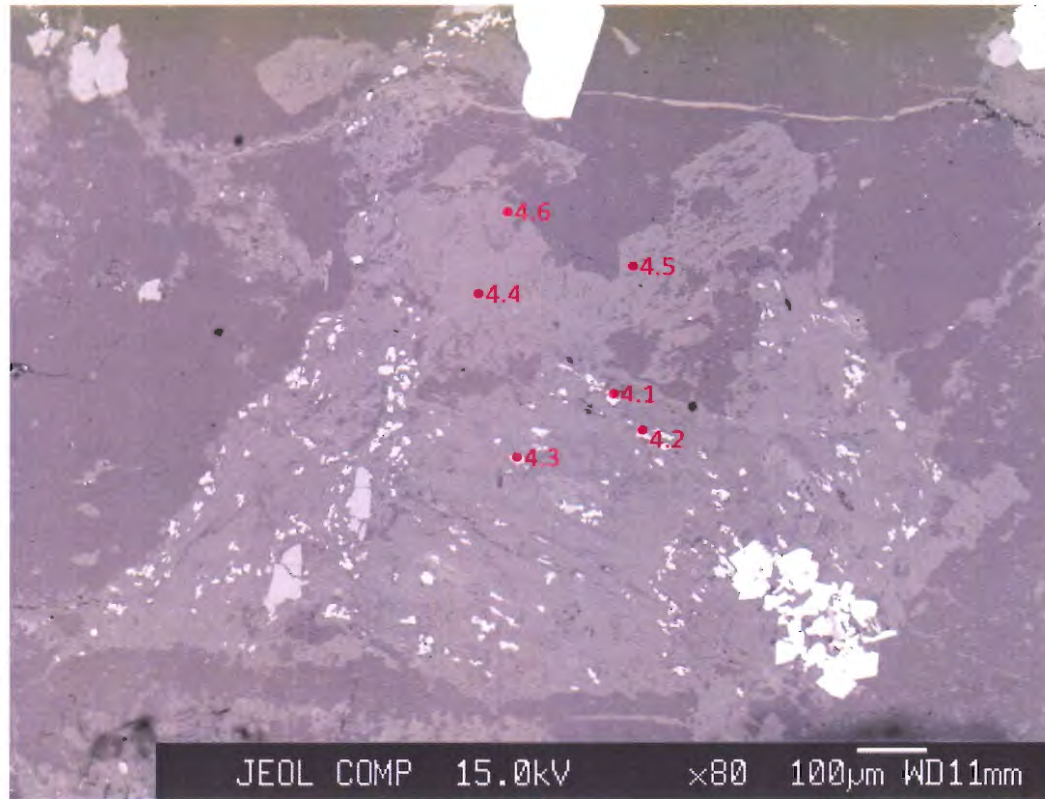


Mass %

No.	K ₂ O	Cr ₂ O ₃	Na ₂ O	Al ₂ O ₃	MnO	CaO	TiO ₂	MgO	SiO ₂	FeO	P ₂ O ₅	Cl	F	NiO	SO ₃	Total	Mineral type
3.1	0.16	0.00	10.97	19.28	0	0.18	0.00	0.00	68.42	0.05	0.01	0.03	0.00	0.01	0.00	99.10	Albite
3.2	9.30	0.00	3.85	18.40	0.004	0.22	0.04	0.00	65.18	0.15	0.01	0.01	0.00	0.04	0.00	97.20	K-feldspar
3.3	0.03	0.02	0.00	19.37	0.684	0.02	0.08	21.69	27.84	15.72	0.05	0.00	0.00	0.00	0.02	85.53	?
3.4	0.17	0.00	11.80	19.01	0	0.44	0.00	0.00	67.95	0.03	0.01	0.01	0.00	0.00	0.01	99.42	Albite
3.5	1.43	0.03	0.01	3.61	1.786	50.54	0.00	0.33	7.07	0.77	0.06	0.00	0.00	0.05	0.05	65.72	?
3.6	0.45	0.00	0.49	8.08	0.375	37.45	0.00	0.09	22.86	0.18	0.06	0.01	0.00	0.01	0.10	70.14	?

Cation Total

No.	K	Cr	Na	Al	Mn	Ca	Ti	Mg	Si	Fe	P	Cl	F	Ni	S	Total	Mineral type
3.1	0.0012	0	0.1169	0.1249	0	0.0011	0	0	0.3761	0.0002	0	0.0003	0	0	0	0.6208	Albite
3.2	0.0685	0	0.0431	0.1254	0	0.0014	0.0002	0	0.3767	0.0007	0	0.0001	0	0.0002	0	0.6163	K-feldspar
3.3	0.0003	0.0001	0	0.1674	0.0042	0.0002	0.0005	0.2371	0.2042	0.0964	0.0003	0	0.0001	0	0.0001	0.711	?
3.4	0.0012	0	0.126	0.1234	0	0.0026	0	0	0.3742	0.0001	0.0001	0.0001	0	0	0	0.6278	Albite
3.5	0.0232	0.0003	0.0002	0.0542	0.0193	0.6893	0	0.0062	0.09	0.0082	0.0007	0	0	0.0005	0.0005	0.8927	?
3.6	0.0056	0	0.0093	0.0935	0.0031	0.394	0	0.0014	0.2245	0.0015	0.0005	0.0002	0	0.0001	0.0007	0.7345	?

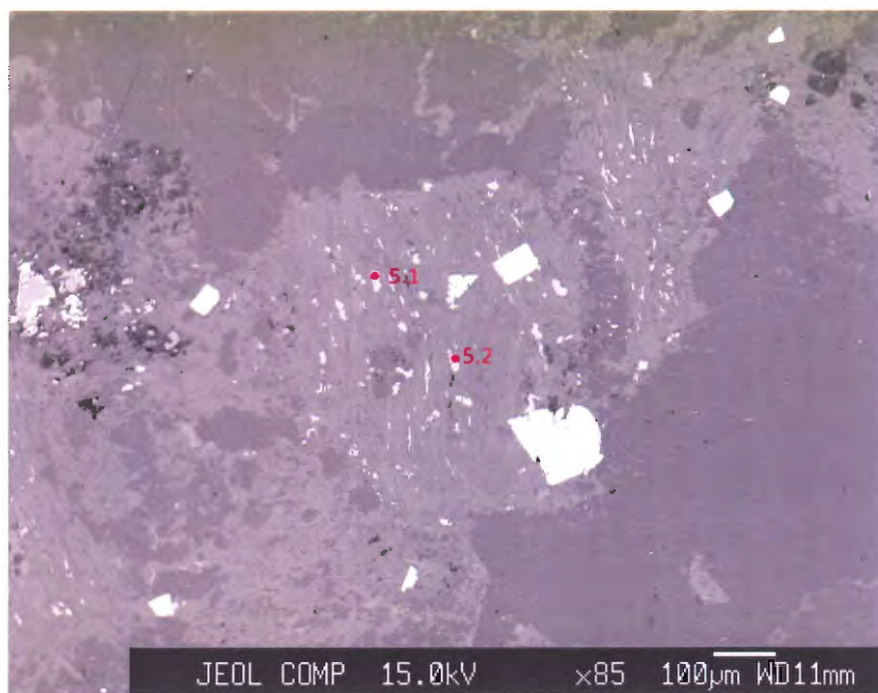


Mass %

No.	K ₂ O	Cr ₂ O ₃	Na ₂ O	Al ₂ O ₃	MnO	CaO	TiO ₂	MgO	SiO ₂	FeO	P ₂ O ₅	Cl	F	NiO	SO ₃	Total	Comment
4.1	0.13	0.15	0.01	0.07	0.078	0.07	94.46	0.00	0.23	0.62	0.05	0.02	0.00	0.11	0.04	96.04	rutile
4.2	0.10	0.12	0.03	0.43	0.099	1.71	88.87	0.00	2.52	0.94	0.07	0.02	0.00	0.13	0.05	95.08	rutile
4.3	0.23	0.17	0.02	0.41	0.137	0.48	90.79	0.00	1.31	1.47	0.03	0.03	0.00	0.11	0.07	95.25	rutile
4.4	15.676	0	0.4241	17.476	0.012	0.017	0.05	0	62.9	0.076	0.003	0.043	0	0.023	0.018	96.704	K-feldspar
4.5	15.788	0	0.3349	17.58	0.004	0	0.014	0	62.67	0.012	0.049	0.068	0	0	0.011	96.516	K-feldspar
4.6	13.776	0.0043	0.4482	17.76	0	0.026	0	0	65.07	0.066	0.016	0.027	0	0	0.002	97.186	K-feldspar

Cation Total

No.	K	Cr	Na	Al	Mn	Ca	Ti	Mg	Si	Fe	P	Cl	F	Ni	S	Total	Mineral type
4.1	0.0011	0.0008	0.0001	0.0006	0.0005	0.0005	0.4937	0	0.0016	0.0036	0.0003	0.0002	0	0.0006	0.0002	0.5038	rutile
4.2	0.0009	0.0007	0.0004	0.0036	0.0006	0.0128	0.4681	0	0.0177	0.0055	0.0004	0.0002	0	0.0007	0.0003	0.512	rutile
4.3	0.0021	0.0009	0.0003	0.0034	0.0008	0.0036	0.4794	0	0.0092	0.0086	0.0002	0.0003	0	0.0006	0.0003	0.5097	rutile
4.4	0.1195	0	0.0049	0.1231	0.0001	0.0001	0.0002	0	0.3759	0.0004	0	0.0004	0	0.0001	0.0001	0.6249	K-feldspar
4.5	0.1206	0	0.0039	0.1241	0	0	0.0001	0	0.3753	0.0001	0.0002	0.0007	0	0	0.0001	0.6252	K-feldspar
4.6	0.1029	0	0.0051	0.1225	0	0.0002	0	0	0.3808	0.0003	0.0001	0.0003	0	0	0	0.6123	K-feldspar



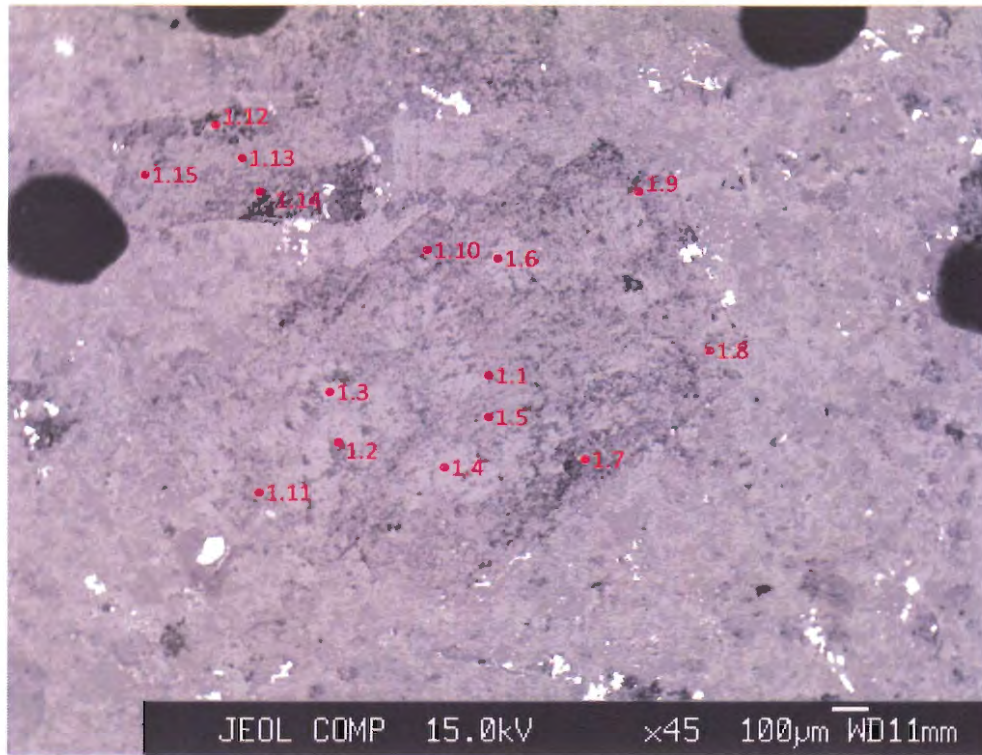
Mass %

No.	K ₂ O	Cr ₂ O ₃	Na ₂ O	Al ₂ O ₃	MnO	CaO	TiO ₂	MgO	SiO ₂	FeO	P ₂ O ₅	Cl	F	NiO	SO ₃	Total	Mineral type
5.1	0.56	0.12	0.02	1.68	0.11	0.12	89.93	0.46	3.01	1.12	0.05	0.02	0.00	0.10	0.05	97.35	rutile
5.2	0.07	0.15	0.00	0.55	0.114	0.32	94.50	0.48	1.10	1.01	0.05	0.02	0.00	0.14	0.07	98.58	rutile

Cation Total

No.	K	Cr	Na	Al	Mn	Ca	Ti	Mg	Si	Fe	P	Cl	F	Ni	S	Total	Mineral type
5.1	0.0048	0.0006	0.0002	0.0135	0.0006	0.0009	0.4603	0.0047	0.0205	0.0064	0.0003	0.0002	0	0.0006	0.0002	0.5138	rutile
5.2	0.0006	0.0008	0	0.0044	0.0007	0.0023	0.4805	0.0048	0.0075	0.0057	0.0003	0.0003	0	0.0008	0.0003	0.5091	rutile

ZMMH 11 Silicates (MM Quartz Porphyry)

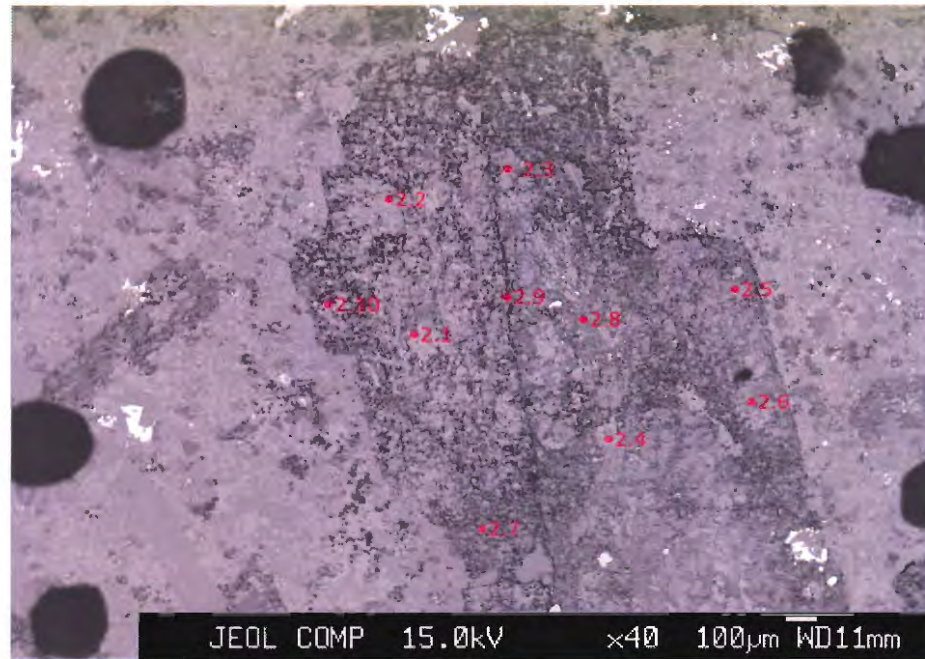


Mass %

No.	K ₂ O	Cr ₂ O ₃	Na ₂ O	SiO ₂	MnO	CaO	TiO ₂	MgO	Al ₂ O ₃	FeO	BaO	Total	Mineral type
1.1	9.29	0.00	0.20	48.71	0.011	0.00	0.06	2.59	28.88	1.38	0.00	91.11	K-feldspar
1.2	0.26	0.00	0.05	44.95	0	0.00	0.00	0.14	35.56	0.05	0.00	81.01	K-feldspar
1.3	15.18	0.00	0.53	63.53	0	0.00	0.13	0.00	18.04	0.04	1.19	98.64	K-feldspar
1.4	15.37	0.00	0.53	63.78	0	0.00	0.05	0.00	17.84	0.03	0.66	98.26	K-feldspar
1.5	15.39	0.00	0.28	63.50	0	0.00	0.12	0.02	18.24	0.10	1.55	99.20	K-feldspar
1.6	10.28	0.00	0.10	47.33	0.095	0.00	0.04	3.04	29.52	2.17	0.00	92.58	K-feldspar
1.7	0.50	0.00	0.03	47.87	0	0.00	0.00	0.19	37.65	0.04	0.00	86.27	K-feldspar
1.8	10.35	0.00	0.25	48.98	0.036	0.16	0.14	2.46	26.76	2.69	0.03	91.86	K-feldspar
1.9	3.55	0.00	0.30	46.61	0	0.59	0.00	1.22	30.30	0.53	0.00	83.10	K-feldspar
1.10	9.08	0.00	0.24	48.89	0.03	0.05	0.08	2.51	27.25	2.27	0.00	90.40	K-feldspar
1.11	14.15	0.00	0.33	61.45	0	1.12	0.00	0.06	16.75	0.12	0.00	93.99	K-feldspar
1.12	7.92	0.00	0.11	35.80	0	0.00	0.00	0.86	14.37	0.40	0.00	59.46	K-feldspar
1.13	9.32	0.00	0.12	47.92	0.019	0.00	0.08	2.49	29.11	1.61	0.00	90.67	K-feldspar
1.14	0.07	0.00	0.08	45.21	0	0.29	0.00	0.10	36.55	0.02	0.00	82.32	unstable
1.15	3.09	0.00	0.07	46.32	0	0.00	0.01	0.68	34.14	0.58	0.00	84.88	K-feldspar

Cation Total

No.	K	Cr	Na	Si	Mn	Ca	Ti	Mg	Al	Fe	Ba	Total	Mineral type
1.1	0.0742	0	0.0024	0.305	0.0001	0	0.0003	0.0242	0.2131	0.0072	0	0.6266	K-feldspar
1.2	0.0022	0	0.0006	0.2934	0	0	0	0.0013	0.2735	0.0003	0	0.5713	K-feldspar
1.3	0.114	0	0.006	0.374	0	0	0.0006	0	0.1252	0.0002	0.0027	0.6227	K-feldspar
1.4	0.1155	0	0.006	0.3756	0	0	0.0002	0	0.1239	0.0002	0.0015	0.6229	K-feldspar
1.5	0.1153	0	0.0032	0.373	0	0	0.0005	0.0002	0.1263	0.0005	0.0036	0.6227	K-feldspar
1.6	0.082	0	0.0013	0.2958	0.0005	0	0.0002	0.0283	0.2175	0.0113	0	0.637	K-feldspar
1.7	0.0039	0	0.0003	0.2938	0	0	0	0.0017	0.2723	0.0002	0	0.5723	K-feldspar
1.8	0.0833	0	0.0031	0.3091	0.0002	0.0011	0.0006	0.0232	0.1991	0.0142	0.0001	0.6341	K-feldspar
1.9	0.0298	0	0.0038	0.3062	0	0.0042	0	0.012	0.2346	0.0029	0	0.5936	K-feldspar
1.10	0.0734	0	0.0029	0.3098	0.0002	0.0004	0.0004	0.0237	0.2035	0.012	0	0.6264	K-feldspar
1.11	0.1106	0	0.004	0.3764	0	0.0074	0	0.0006	0.1209	0.0006	0	0.6205	K-feldspar
1.12	0.0974	0	0.0021	0.345	0	0	0	0.0124	0.1632	0.0032	0	0.6234	K-feldspar
1.13	0.075	0	0.0015	0.3022	0.0001	0	0.0004	0.0234	0.2164	0.0085	0	0.6275	K-feldspar
1.14	0.0005	0	0.001	0.2905	0	0.002	0	0.001	0.2768	0.0001	0	0.572	unstable
1.15	0.0252	0	0.0008	0.2959	0	0	0	0.0065	0.257	0.0031	0	0.5885	K-feldspar



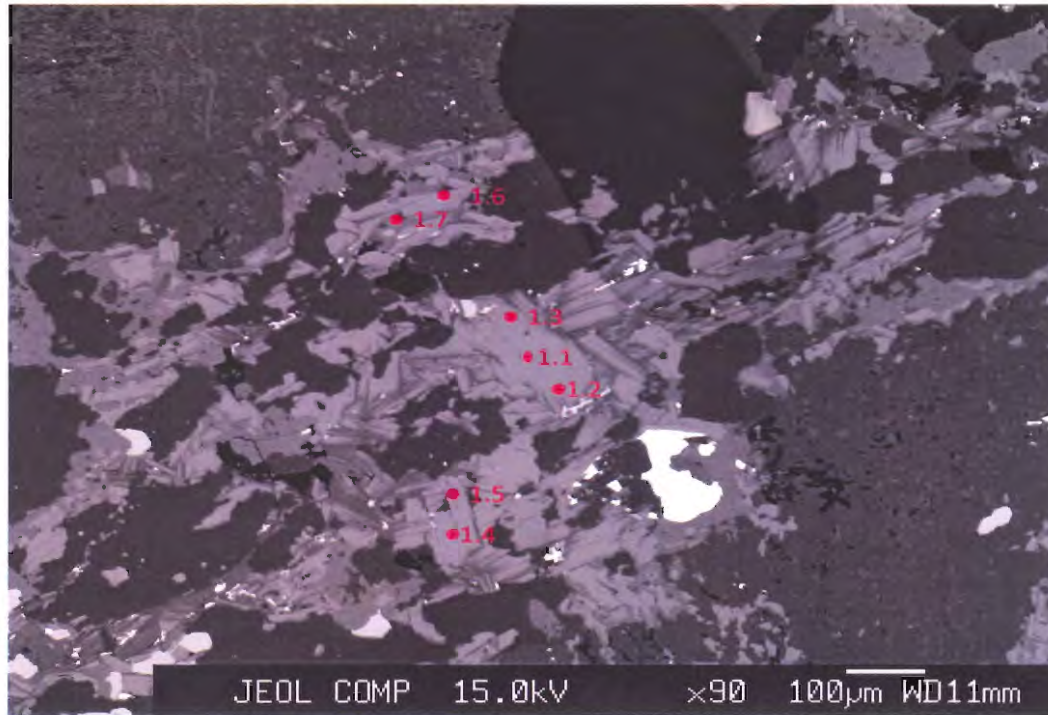
Mass %

No.	K ₂ O	Cr ₂ O ₃	Na ₂ O	SiO ₂	MnO	CaO	TiO ₂	MgO	Al ₂ O ₃	FeO	BaO	Total	Mineral Type
2.1	15.32	0.00	0.33	64.15	0	0.01	0.07	0.03	18.43	0.06	0.77	99.16	K-feldspar
2.2	8.83	0.00	0.13	48.80	0.022	0.00	0.00	2.07	29.39	2.83	0.00	92.07	K-feldspar
2.3	8.55	0.00	0.13	50.78	0.021	0.00	0.00	1.65	31.77	1.29	0.00	94.19	K-feldspar
2.4	8.52	0.00	0.11	48.73	0.008	0.00	0.00	2.05	30.03	1.11	0.00	90.56	K-feldspar
2.5	6.26	0.00	0.20	51.97	0.005	0.01	0.02	0.24	30.48	0.13	0.20	89.52	K-feldspar
2.6	5.45	0.00	0.22	42.12	0.022	0.00	0.01	1.56	26.13	1.53	0.03	77.07	K-feldspar
2.7	7.83	0.00	0.09	37.48	0	0.00	0.00	0.00	10.54	0.00	0.00	55.95	K-feldspar
2.8	15.50	0.05	0.25	63.60	0.033	0.03	0.08	0.01	18.08	0.12	0.87	98.63	K-feldspar
2.9	14.50	0.00	0.29	58.65	0	0.04	0.02	0.03	16.55	0.06	0.32	90.45	K-feldspar
2.10	10.18	0.04	0.23	47.16	0.058	0.20	0.02	1.61	23.08	1.47	0.18	84.23	K-feldspar

Cation Total

No.	K	Cr	Na	Si	Mn	Ca	Ti	Mg	Al	Fe	Ba	Total	Mineral
2.1	0.114	0	0.0037	0.3741	0	0	0.0003	0.0002	0.1267	0.0003	0.0018	0.6212	K-feldspar
2.2	0.0701	0	0.0015	0.3035	0.0001	0	0	0.0192	0.2154	0.0147	0	0.6245	K-feldspar
2.3	0.0653	0	0.0015	0.3043	0.0001	0	0	0.0147	0.2244	0.0065	0	0.6169	K-feldspar
2.4	0.0679	0	0.0014	0.3044	0	0	0	0.0191	0.2211	0.0058	0	0.6198	K-feldspar
2.5	0.0491	0	0.0024	0.3196	0	0.0001	0.0001	0.0022	0.2209	0.0007	0.0005	0.5956	K-feldspar
2.6	0.0504	0	0.0031	0.3057	0.0001	0	0.0001	0.0168	0.2235	0.0093	0.0001	0.6091	K-feldspar
2.7	0.1013	0	0.0017	0.3798	0	0	0	0.0001	0.1259	0	0	0.6089	K-feldspar
2.8	0.1163	0.0002	0.0029	0.3741	0.0002	0.0002	0.0004	0.0001	0.1254	0.0006	0.002	0.6224	K-feldspar
2.9	0.1183	0	0.0035	0.3751	0	0.0003	0.0001	0.0003	0.1247	0.0003	0.0008	0.6234	K-feldspar
2.1	0.089	0.0002	0.0031	0.3233	0.0003	0.0015	0.0001	0.0165	0.1865	0.0084	0.0005	0.6295	K-feldspar

ZMMH-14 Silicates (MM Porphyry)

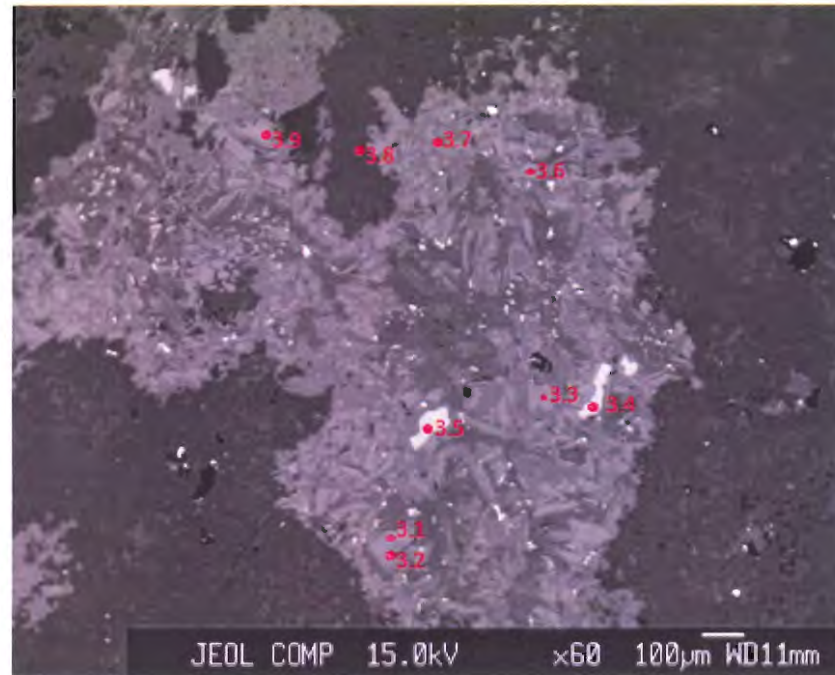


Mass %

No.	K ₂ O	Cr ₂ O ₃	Na ₂ O	Al ₂ O ₃	MnO	CaO	TiO ₂	MgO	SiO ₂	FeO	P ₂ O ₅	Cl	F	NiO	SO ₃	Total	Mineral type
1.1	8.70	0.00	0.17	14.25	0.05	0.00	2.96	16.69	38.01	11.16	0.03	0.13	0.96	0.05	0.04	92.78	biotite
1.2	8.17	0.00	0.18	14.37	0.08	0.03	2.78	16.23	37.97	10.37	0.02	0.14	1.01	0.06	0.03	90.99	biotite
1.3	8.92	0.01	0.08	14.21	0.084	0.02	2.96	17.07	38.26	10.95	0.00	0.13	1.13	0.02	0.00	93.33	biotite
1.4	9.63	0.03	0.18	14.93	0.077	0.00	3.00	17.18	38.58	10.89	0.00	0.14	1.03	0.02	0.00	95.23	biotite
1.7	15.57	0.00	0.72	18.03	0	0.01	0.00	0.00	64.38	0.14	0.02	0.01	0.00	0.00	0.00	98.88	K-feldspar
1.5	9.81	0.05	0.17	15.41	0.085	0.00	2.86	16.44	37.62	11.05	0.00	0.13	0.99	0.02	0.00	94.19	biotite
1.6	9.68	0.01	0.16	14.62	0.104	0.00	2.86	17.03	38.72	10.78	0.01	0.12	1.13	0.02	0.00	94.74	biotite

Cation Total

No.	K2O	Cr2O3	Na2O	Al2O3	MnO	CaO	TiO2	MgO	SiO2	FeO	P2O5	Cl	F	NiO	SO3	Total	Mineral type
1.1	0.0761	0	0.0023	0.1151	0.0003	0	0.0153	0.1706	0.2607	0.064	0.0002	0.0015	0.0204	0.0003	0.0002	0.727	biotite
1.2	0.0723	0	0.0024	0.1176	0.0005	0.0002	0.0145	0.168	0.2636	0.0602	0.0001	0.0016	0.0217	0.0003	0.0001	0.7232	biotite
1.3	0.0776	0.0001	0.0011	0.1142	0.0005	0.0001	0.0152	0.1736	0.261	0.0625	0	0.0014	0.0238	0.0001	0	0.7313	biotite
1.4	0.0823	0.0001	0.0024	0.1179	0.0004	0	0.0151	0.1716	0.2585	0.061	0	0.0015	0.0214	0.0001	0	0.7324	biotite
1.5	0.0832	0.0001	0.002	0.1161	0.0006	0	0.0145	0.171	0.2608	0.0608	0	0.0014	0.0235	0.0001	0	0.7342	K-feldspar
1.6	0.0851	0.0003	0.0022	0.1235	0.0005	0	0.0146	0.1666	0.2557	0.0628	0	0.0015	0.0209	0.0001	0	0.7339	biotite
1.7	0.1158	0	0.0081	0.1239	0	0.0001	0	0	0.3755	0.0007	0.0001	0.0001	0	0	0	0.6244	biotite



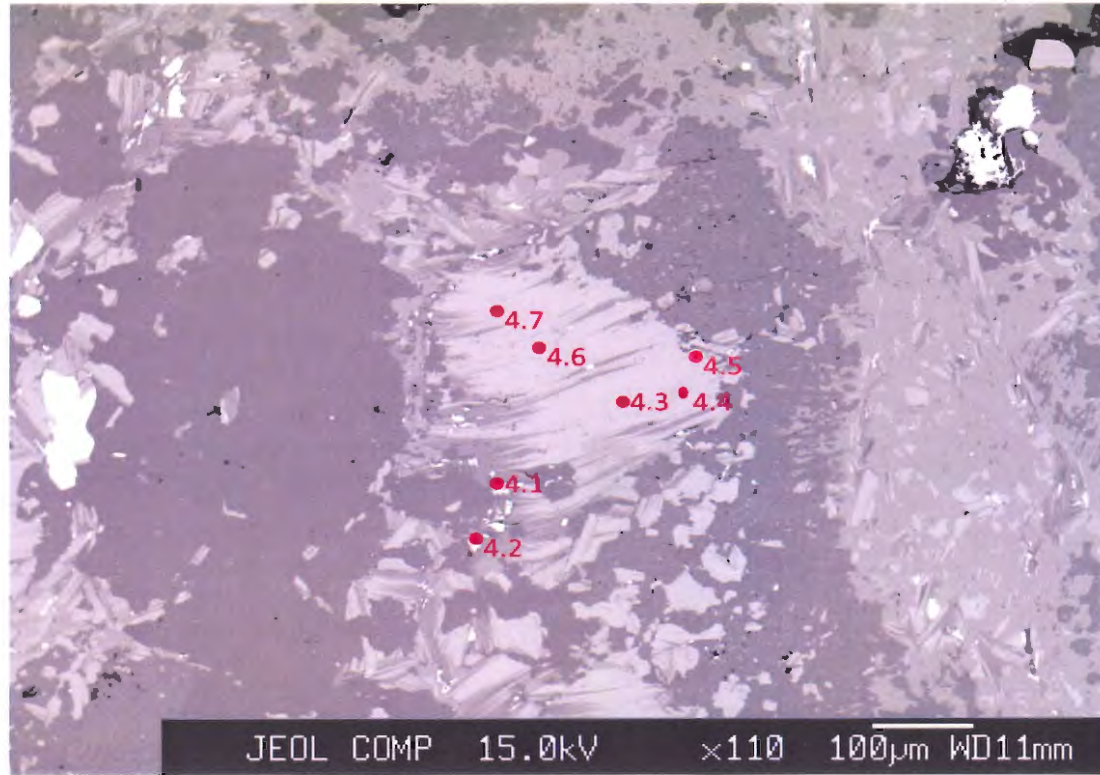
Mass %

No.	K ₂ O	Cr ₂ O ₃	Na ₂ O	Al ₂ O ₃	MnO	CaO	TiO ₂	MgO	SiO ₂	FeO	P ₂ O ₅	Cl	F	NiO	SO ₃	Total	Mineral type
3.1	7.77	0.01	0.17	14.59	0.134	0.28	2.89	15.97	36.80	11.41	0.00	0.16	0.85	0.04	0.04	90.72	biotite
3.2	7.79	0.02	0.21	14.48	0.09	0.28	2.72	15.82	36.90	10.74	0.00	0.14	0.87	0.02	0.00	89.67	biotite
3.3	9.02	0.01	0.21	14.92	0.05	0.01	3.04	16.39	37.52	11.19	0.00	0.13	0.88	0.01	0.00	92.97	biotite
3.4	0.10	0.05	0.19	0.00	0.209	54.15	0.01	0.11	0.10	0.44	41.27	0.66	3.33	0.08	0.03	99.18	apatite
3.5	0.05	0.06	0.19	0.00	0.209	54.13	0.03	0.11	0.07	0.33	41.39	0.75	3.82	0.04	0.05	99.46	apatite
3.6	8.94	0.03	0.08	15.12	0.2	0.01	2.40	17.20	37.40	10.98	0.02	0.12	1.16	0.03	0.04	93.19	biotite
3.7	15.81	0.00	0.73	17.86	0	0.03	0.00	0.00	64.08	0.10	0.00	0.00	0.00	0.00	0.00	98.62	K-feldspar
3.8	1.31	0.58	1.25	7.54	0.215	0.51	0.45	0.59	24.61	1.00	0.32	1.00	0.21	0.58	1.37	41.22	chlorite?
3.9	8.75	0.00	0.23	14.42	0.014	0.07	2.95	16.27	37.59	10.99	0.00	0.14	0.98	0.05	0.00	92.00	biotite

Cation Total

No.	K	Cr	Na	Al	Mn	Ca	Ti	Mg	Si	Fe	P	Cl	F	Ni	S	Total	Mineral type
3.1	0.0694	0.0001	0.0023	0.1204	0.0008	0.0021	0.0152	0.1667	0.2578	0.0668	0	0.0018	0.0185	0.0002	0.0002	0.7224	biotite
3.2	0.0702	0.0001	0.0028	0.1205	0.0005	0.0021	0.0145	0.1665	0.2605	0.0634	0	0.0016	0.019	0.0001	0	0.7219	biotite
3.3	0.0789	0.0001	0.0027	0.1206	0.0003	0.0001	0.0157	0.1675	0.2573	0.0642	0	0.0014	0.0187	0.0001	0	0.7277	biotite
3.4	0.023	0.0064	0.0334	0.1223	0.0025	0.0075	0.0047	0.0121	0.3388	0.0115	0.0037	0.0227	0.0088	0.0064	0.0142	0.6181	apatite
3.5	0.0781	0.0001	0.0011	0.122	0.0012	0	0.0124	0.1755	0.256	0.0628	0.0001	0.0013	0.0244	0.0002	0.0002	0.7354	apatite
3.6	0.0004	0.0003	0.0026	0	0.0012	0.395	0.0002	0.0011	0.0005	0.0019	0.2386	0.0079	0.0754	0.0002	0.0002	0.7255	biotite
3.7	0.1182	0	0.0083	0.1233	0	0.0002	0	0	0.3755	0.0005	0	0	0	0	0	0.626	K-feldspar
3.8	0.0773	0	0.003	0.1176	0.0001	0.0005	0.0154	0.1679	0.2602	0.0636	0	0.0016	0.021	0.0003	0	0.7286	chlorite?
3.9	0.0009	0.0002	0.0025	0	0.0012	0.3954	0.0001	0.0012	0.0007	0.0025	0.2381	0.007	0.0664	0.0004	0.0002	0.7168	biotite

** There is no location 2 for the ZMMH-14 EMP analysis

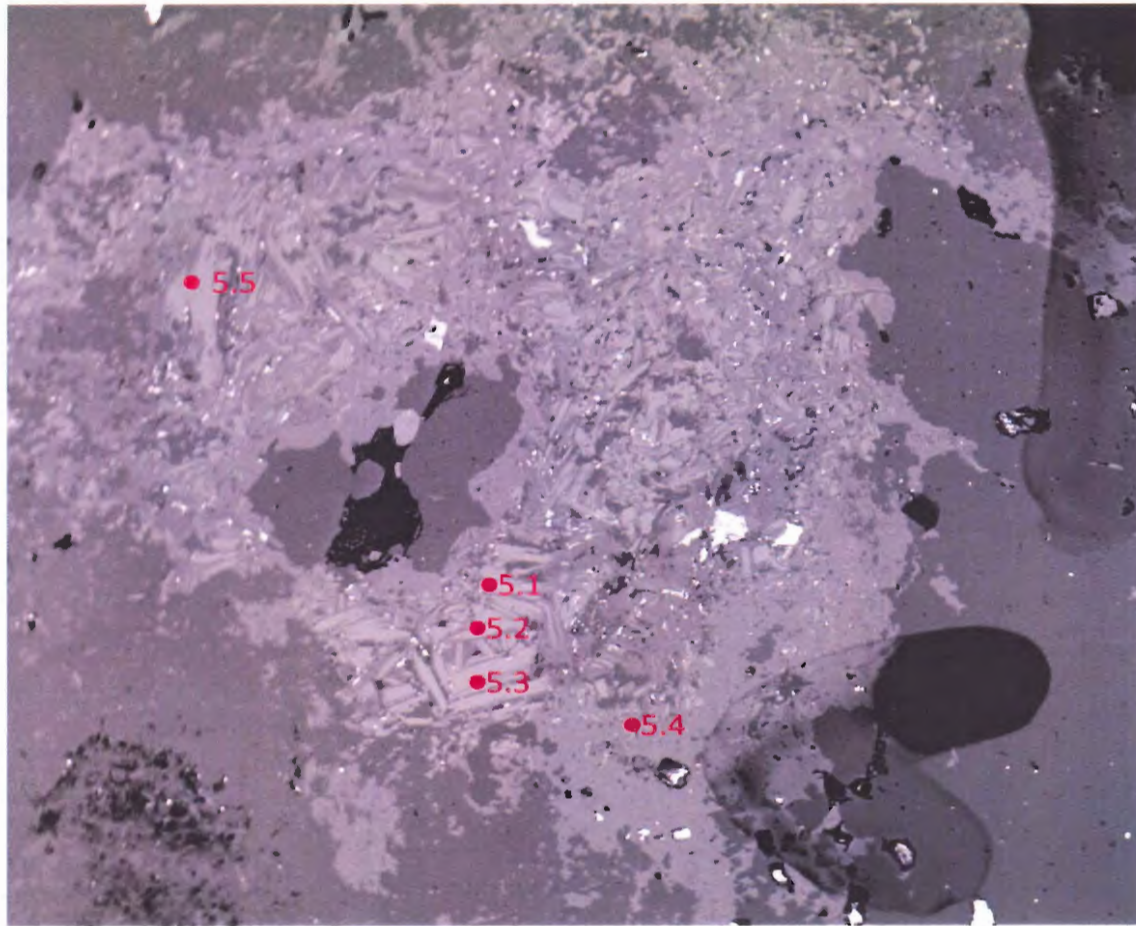


Mass %

No.	K ₂ O	Cr ₂ O ₃	Na ₂ O	Al ₂ O ₃	MnO	CaO	TiO ₂	MgO	SiO ₂	FeO	P ₂ O ₅	Cl	F	NiO	SO ₃	Total	Mineral type
4.1	0.07	0.12	0.01	0.16	0.035	1.11	93.12	0.04	1.48	0.73	0.02	0.01	0.00	0.10	0.01	97.02	rutile
4.2	0.10	0.00	0.00	0.03	0.002	0.01	95.12	0.00	0.09	1.13	0.02	0.00	0.03	0.03	0.00	96.57	rutile
4.3	9.30	0.02	0.25	14.84	0.066	0.02	3.11	17.09	38.94	11.38	0.00	0.14	1.07	0.03	0.02	95.79	biotite
4.4	9.23	0.02	0.21	14.52	0.034	0.01	3.09	16.80	38.56	11.59	0.00	0.14	1.06	0.02	0.02	94.81	biotite
4.5	8.97	0.00	0.23	14.32	0.026	0.06	3.08	16.59	38.35	11.07	0.00	0.11	0.92	0.04	0.01	93.38	biotite
4.6	9.42	0.02	0.14	14.59	0.083	0.02	3.03	16.78	38.52	11.30	0.00	0.12	1.04	0.03	0.01	94.63	biotite
4.7	9.42	0.02	0.20	14.40	0.056	0.01	3.20	16.92	38.28	11.52	0.00	0.15	0.96	0.04	0.00	94.75	biotite

Cation Total

No.	K	Cr	Na	Al	Mn	Ca	Ti	Mg	Si	Fe	P	Cl	F	Ni	S	Total	Mineral type
4.1	0.0006	0.0007	0.0001	0.0013	0.0002	0.0081	0.4812	0.0004	0.0102	0.0042	0.0001	0.0002	0	0.0006	0.0001	0.5081	rutile
4.2	0.0009	0	0	0.0002	0	0.0001	0.4954	0	0.0007	0.0066	0.0001	0	0.0007	0.0002	0	0.5049	rutile
4.3	0.079	0.0001	0.0032	0.1165	0.0004	0.0002	0.0156	0.1697	0.2594	0.0634	0	0.0015	0.0221	0.0002	0.0001	0.7314	biotite
4.4	0.0794	0.0001	0.0028	0.1153	0.0002	0.0001	0.0157	0.1688	0.2599	0.0653	0	0.0016	0.022	0.0001	0.0001	0.7315	biotite
4.5	0.078	0	0.0031	0.115	0.0002	0.0004	0.0158	0.1685	0.2613	0.0631	0	0.0013	0.0195	0.0002	0.0001	0.7265	biotite
4.6	0.0811	0.0001	0.0018	0.1161	0.0005	0.0001	0.0154	0.1688	0.26	0.0638	0	0.0013	0.0216	0.0002	0	0.7308	biotite
4.7	0.0812	0.0001	0.0026	0.1147	0.0003	0.0001	0.0163	0.1704	0.2586	0.0651	0	0.0017	0.02	0.0002	0	0.7313	biotite

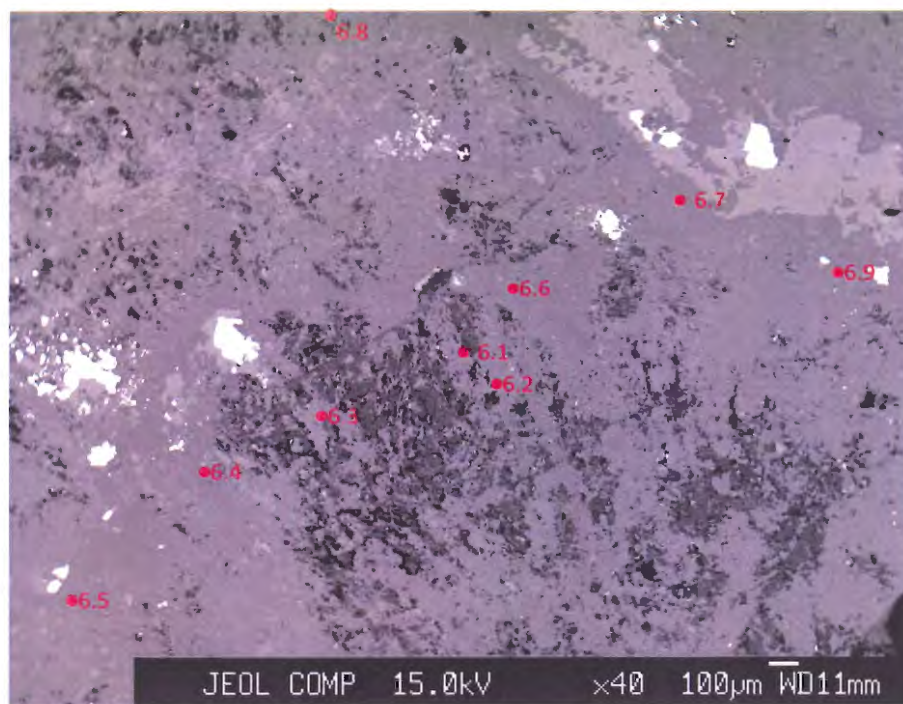


Mass %

No.	K ₂ O	Cr ₂ O ₃	Na ₂ O	Al ₂ O ₃	MnO	CaO	TiO ₂	MgO	SiO ₂	FeO	P ₂ O ₅	Cl	F	NiO	SO ₃	Total	Mineral type
5.1	9.58	0.05	0.23	14.92	0.036	0.00	3.28	16.52	37.93	11.59	0.01	0.12	0.92	0.02	0.02	94.80	K-feldspar
5.2	9.78	0.00	0.15	14.49	0.116	0.01	2.99	16.88	38.41	11.18	0.02	0.14	1.07	0.04	0.03	94.82	K-feldspar
5.3	9.59	0.05	0.25	14.76	0.048	0.03	3.28	16.57	38.62	11.54	0.00	0.13	0.99	0.02	0.00	95.43	K-feldspar
5.4	16.22	0.00	0.47	18.01	0	0.02	0.00	0.00	64.38	0.14	0.07	0.00	0.00	0.02	0.00	99.32	K-feldspar
5.5	9.35	0.01	0.29	14.58	0.043	0.03	3.30	16.55	38.31	11.46	0.00	0.12	0.99	0.02	0.02	94.61	K-feldspar

Cation Total

No.	K	Cr	Na	Al	Mn	Ca	Ti	Mg	Si	Fe	P	Cl	F	Ni	S	Total	Mineral type
5.1	0.0826	0.0003	0.003	0.1189	0.0002	0	0.0167	0.1664	0.2563	0.0655	0.0001	0.0014	0.0192	0.0001	0.0001	0.7309	K-feldspar
5.2	0.0842	0	0.002	0.1154	0.0007	0.0001	0.0152	0.1699	0.2594	0.0631	0.0001	0.0016	0.0222	0.0002	0.0001	0.7343	K-feldspar
5.3	0.082	0.0003	0.0033	0.1166	0.0003	0.0002	0.0165	0.1657	0.259	0.0647	0	0.0014	0.0205	0.0001	0	0.7307	K-feldspar
5.4	0.1206	0	0.0053	0.1236	0	0.0001	0	0	0.375	0.0007	0.0003	0	0	0.0001	0	0.6258	K-feldspar
5.5	0.0806	0	0.0038	0.1161	0.0002	0.0002	0.0168	0.1667	0.2589	0.0647	0	0.0014	0.0208	0.0001	0.0001	0.7305	K-feldspar

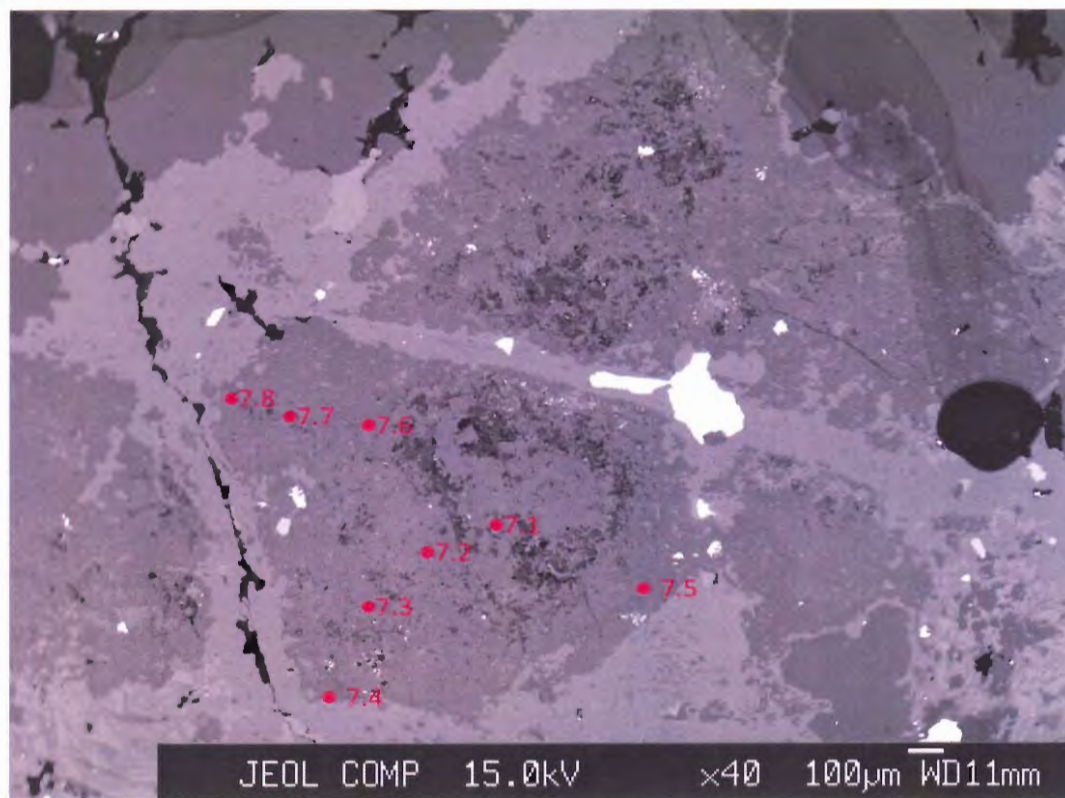


Mass %

No.	K ₂ O	Cr ₂ O ₃	Na ₂ O	Al ₂ O ₃	MnO	CaO	TiO ₂	MgO	SiO ₂	FeO	P ₂ O ₅	Cl	F	NiO	SO ₃	Total	Mineral type
6.1	0.23	0.00	7.92	24.66	0	6.89	0.00	0.00	60.09	0.01	0.00	0.01	0.02	0.00	0.00	99.81	Andesine
6.2	0.24	0.00	7.50	24.68	0	6.75	0.00	0.00	60.37	0.00	0.00	0.00	0.00	0.00	0.00	99.55	Andesine
6.3	0.20	0.00	9.09	21.81	0	3.43	0.00	0.00	65.74	0.00	0.01	0.07	0.00	0.00	0.00	100.34	Oligoclase
6.4	0.27	0.00	9.51	21.56	0	3.24	0.00	0.00	65.23	0.00	0.03	0.04	0.00	0.00	0.00	99.88	Oligoclase
6.5	0.42	0.00	4.74	20.37	0	1.60	0.00	0.01	68.03	0.03	0.28	0.00	0.01	0.00	0.02	95.51	Oligoclase
6.6	0.17	0.00	5.30	21.52	0	2.17	0.00	0.00	68.96	0.00	0.02	0.00	0.01	0.00	0.00	98.13	Oligoclase
6.7	0.34	0.01	9.92	21.63	0.006	3.33	0.00	0.00	64.23	0.06	0.00	0.00	0.00	0.01	0.02	99.56	Oligoclase
6.8	0.29	0.00	9.74	21.79	0	3.36	0.00	0.00	64.10	0.03	0.00	0.00	0.00	0.01	0.01	99.33	Oligoclase
6.9	1.75	0.00	1.86	24.27	0.011	2.04	0.01	0.43	64.97	0.47	0.00	0.01	0.04	0.01	0.00	95.83	unstable

Cation Total

No.	K	Cr	Na	Al	Mn	Ca	Ti	Mg	Si	Fe	P	Cl	F	Ni	S	Total	Mineral type
6.1	0.0016	0	0.0857	0.1624	0	0.0412	0	0	0.3357	0.0001	0	0.0001	0.0003	0	0	0.6272	Andesine
6.2	0.0017	0	0.0813	0.1625	0	0.0404	0	0	0.3372	0	0	0	0	0	0	0.6231	Andesine
6.3	0.0014	0	0.0965	0.1407	0	0.0201	0	0	0.3599	0	0	0.0006	0	0	0	0.6192	Oligoclase
6.4	0.0019	0	0.1016	0.14	0	0.0191	0	0	0.3594	0	0.0002	0.0004	0	0	0	0.6226	Oligoclase
6.5	0.003	0	0.0512	0.1339	0	0.0096	0	0.0001	0.3794	0.0001	0.0013	0	0.0001	0	0.0001	0.5789	Oligoclase
6.6	0.0012	0	0.056	0.1382	0	0.0126	0	0	0.3757	0	0.0001	0	0.0001	0	0	0.584	Oligoclase
6.7	0.0024	0.0001	0.1068	0.1414	0	0.0198	0	0	0.3564	0.0003	0	0	0	0	0.0001	0.6274	Oligoclase
6.8	0.002	0	0.1049	0.1427	0	0.02	0	0	0.3561	0.0002	0	0	0	0	0	0.6259	Oligoclase
6.9	0.0125	0	0.0201	0.1598	0.0001	0.0122	0	0.0035	0.363	0.0022	0	0.0001	0.0007	0	0	0.5743	unstable



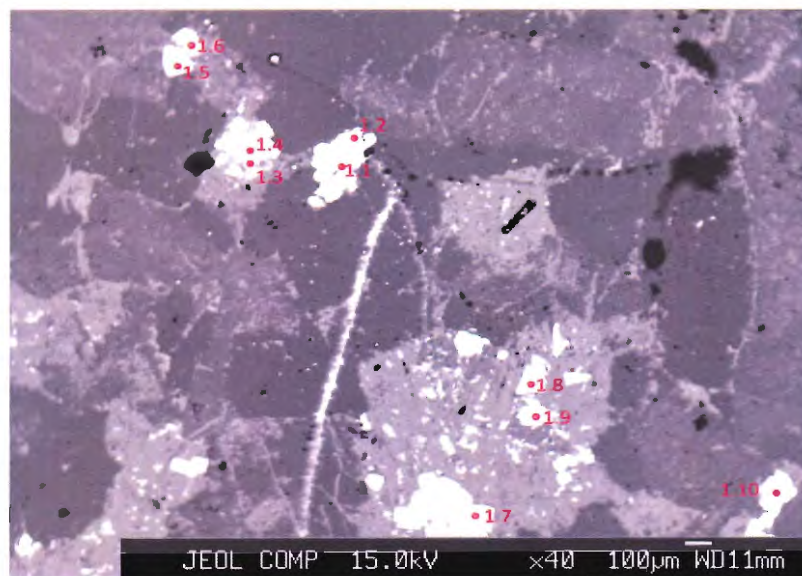
Mass %

No.	K ₂ O	Cr ₂ O ₃	Na ₂ O	Al ₂ O ₃	MnO	CaO	TiO ₂	MgO	SiO ₂	FeO	P ₂ O ₅	Cl	F	NiO	SO ₃	Total	Mineral type
7.1	5.53	0.00	3.98	27.43	0	1.91	0.01	0.63	57.19	0.84	0.75	0.01	0.18	0.00	0.04	98.42	K-feldspar
7.2	4.65	0.00	4.42	24.52	0	4.03	0.00	0.29	58.02	0.24	0.00	0.03	0.03	0.03	0.03	96.27	unstable
7.3	0.47	0.00	9.76	21.56	0	3.13	0.00	0.01	64.65	0.06	0.01	0.00	0.00	0.01	0.02	99.68	Oligoclase
7.4	0.37	0.00	2.69	21.30	0	1.89	0.00	0.03	69.68	0.03	0.00	0.04	0.02	0.00	0.02	96.06	unstable
7.5	0.37	0.00	7.59	21.10	0	2.13	0.00	0.00	68.30	0.09	0.00	0.00	0.00	0.01	0.02	99.62	Anorthoclase
7.6	0.24	0.00	8.62	24.24	0	5.93	0.00	0.01	61.24	0.05	0.00	0.00	0.00	0.00	0.02	100.35	Andesine
7.7	8.21	0.02	1.36	30.97	0	0.67	0.02	1.15	52.50	1.16	0.00	0.01	0.12	0.03	0.05	96.22	K-feldspar
7.8	0.52	0.00	4.98	23.83	0.008	4.41	0.00	0.02	64.98	0.06	0.00	0.01	0.00	0.04	0.03	98.89	Oligoclase

Cation total

No.	K	Cr	Na	Al	Mn	Ca	Ti	Mg	Si	Fe	P	Cl	F	Ni	S	Total	Mineral type
7.1	0.0402	0	0.044	0.1841	0	0.0116	0.0001	0.0053	0.3256	0.004	0.0036	0.0001	0.0032	0	0.0002	0.622	K-feldspar
7.2	0.0345	0	0.0499	0.1683	0	0.0251	0	0.0025	0.3379	0.0012	0	0.0002	0.0005	0.0001	0.0002	0.6205	unstable
7.3	0.0033	0	0.1048	0.1407	0	0.0186	0	0	0.3578	0.0003	0.0001	0	0	0.0001	0.0001	0.6259	Oligoclase
7.4	0.0026	0	0.0287	0.1379	0	0.0111	0	0.0002	0.3828	0.0002	0	0.0004	0.0003	0	0.0001	0.5644	unstable
7.5	0.0025	0	0.08	0.1352	0	0.0124	0	0	0.3713	0.0004	0	0	0	0.0001	0.0001	0.6021	Anorthoclase
7.6	0.0017	0	0.0927	0.1584	0	0.0353	0	0.0001	0.3397	0.0002	0	0	0	0	0.0001	0.6283	Andesine
7.7	0.0616	0.0001	0.0156	0.2148	0	0.0042	0.0001	0.0101	0.309	0.0057	0	0.0001	0.0022	0.0002	0.0002	0.6239	K-feldspar
7.8	0.0037	0	0.053	0.1542	0	0.026	0	0.0002	0.3567	0.0003	0	0.0001	0	0.0002	0.0001	0.5946	Oligoclase

ZMMH-10 Sulphides (MM Granodiorite)

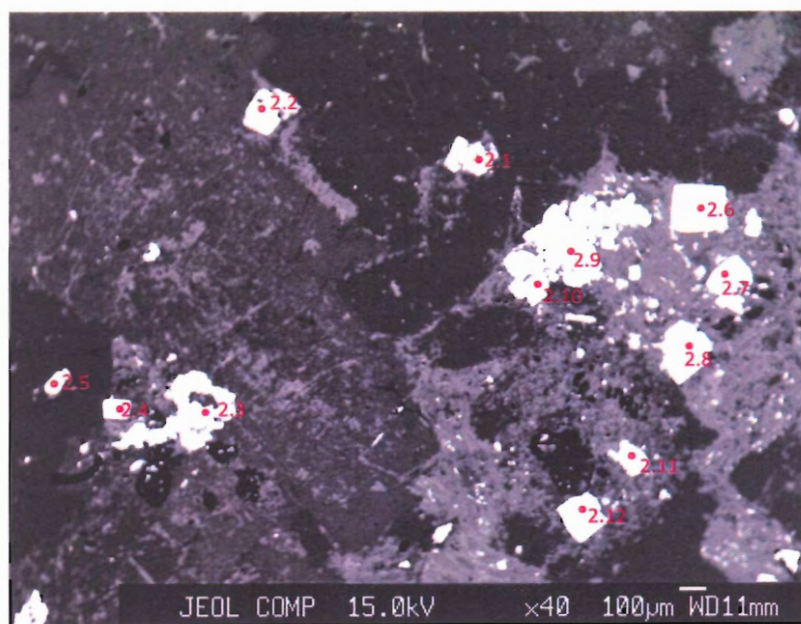


Mass Percent

No.	Mo	S	As	Fe	Pb	Ag	Co	Ti	Ca	Cu	Sn	Sb	Mn	Zn	W	Total	Mineral
1.1	0.0072	50.2275	0.0153	43.9959	0.2043	0.0741	0.3078	0.1203	0.0458	0.269	0.1362	0.1062	0.1487	0.4393	0.9603	97.0579	pyrite
1.2	0.0184	51.3077	0.0498	43.1368	0.2411	0.0437	0.3065	0.099	0.0096	0.2723	0.1494	0.1172	0.1079	0.4171	0.8029	97.0795	pyrite
1.3	0	31.3435	0.0494	30.7517	0.0558	0.0138	0.0126	0.1141	0.057	0.1292	0.1118	0.0487	0.3447	0.2137	0.452	63.6981	pyrite
1.4	0	27.8536	0	24.4639	0	0	0.1074	0.0594	1.633	0.0933	0.0468	0.0015	0.0605	0.1974	0.4246	54.9414	pyrite
1.5	0.0092	51.4537	0.0354	43.4407	0.2028	0.0646	0.2247	0.1068	0.0291	0.255	0.1524	0.1016	0.1069	0.4923	1.0286	97.7038	pyrite
1.6	0.0072	52.1252	0.0283	43.7317	0.2023	0.0454	0.1671	0.1156	0.0243	0.2602	0.1621	0.1211	0.1083	0.4297	0.9519	98.4805	pyrite
1.7	0	52.1629	0.0655	43.8929	0.1745	0.0638	0.0536	0.0978	0.0329	0.2875	0.1354	0.1035	0.1042	0.4066	0.8859	98.467	pyrite
1.8	0	51.391	0.071	43.9334	0.1302	0.0553	0.2331	0.1944	0.0162	0.228	0.1096	0.1023	0.0943	0.4416	0.8927	97.8931	pyrite
1.9	0.0123	51.6535	0.0608	43.6425	0.1648	0.047	0.2571	0.118	0.0084	0.2409	0.1697	0.1078	0.1173	0.4258	0.9795	98.0054	pyrite
1.10	0	6.239	0	4.1785	0	0	0.0186	0.0075	0.9174	0	0.0352	0.0622	0	0	0	11.4584	pyrite
1.11	0.0525	52.3401	0.052	44.0553	0.2127	0.0542	0.3695	0.1229	0.0395	0.3062	0.16	0.1114	0.1226	0.4655	0.9692	99.4336	pyrite

Atomic Ratio

No.	Mo	S	As	Fe	Pb	Ag	Co	Ti	Ca	Cu	Sn	Sb	Mn	Zn	W	Total	Mineral
1.1	0.0031	65.6568	0.0086	33.0134	0.0413	0.0288	0.2189	0.1052	0.0479	0.1774	0.0481	0.0365	0.1134	0.2816	0.2189	100	pyrite
1.2	0.008	66.6295	0.0277	32.1568	0.0485	0.0169	0.2165	0.086	0.01	0.1784	0.0524	0.0401	0.0818	0.2657	0.1818	100	pyrite
1.3	0	63.1265	0.0426	35.5528	0.0174	0.0083	0.0138	0.1537	0.0919	0.1313	0.0608	0.0258	0.4052	0.2111	0.1588	100	pyrite
1.4	0	63.932	0	32.2332	0	0	0.1341	0.0913	2.9982	0.108	0.029	0.0009	0.081	0.2222	0.17	100	pyrite
1.5	0.004	66.5156	0.0196	32.2362	0.0406	0.0248	0.158	0.0924	0.0301	0.1663	0.0532	0.0346	0.0806	0.3121	0.2319	100	pyrite
1.6	0.0031	66.7221	0.0155	32.1336	0.0401	0.0173	0.1164	0.0991	0.0249	0.168	0.0561	0.0408	0.0809	0.2698	0.2125	100	pyrite
1.7	0	66.7235	0.0359	32.2294	0.0345	0.0242	0.0373	0.0838	0.0337	0.1855	0.0468	0.0349	0.0778	0.2551	0.1976	100	pyrite
1.8	0	66.2715	0.0392	32.5218	0.026	0.0212	0.1635	0.1678	0.0167	0.1483	0.0382	0.0347	0.0709	0.2793	0.2008	100	pyrite
1.9	0.0053	66.5252	0.0335	32.2654	0.0328	0.018	0.1802	0.1017	0.0086	0.1565	0.059	0.0365	0.0881	0.269	0.22	99.9998	pyrite
1.1	0	66.2844	0	25.4836	0	0	0.1075	0.0531	7.7963	0	0.1009	0.1741	0	0	0	100	pyrite
1.11	0.0223	66.4812	0.0283	32.1221	0.0418	0.0204	0.2553	0.1045	0.0402	0.1962	0.0549	0.0373	0.0908	0.29	0.2147	100	pyrite



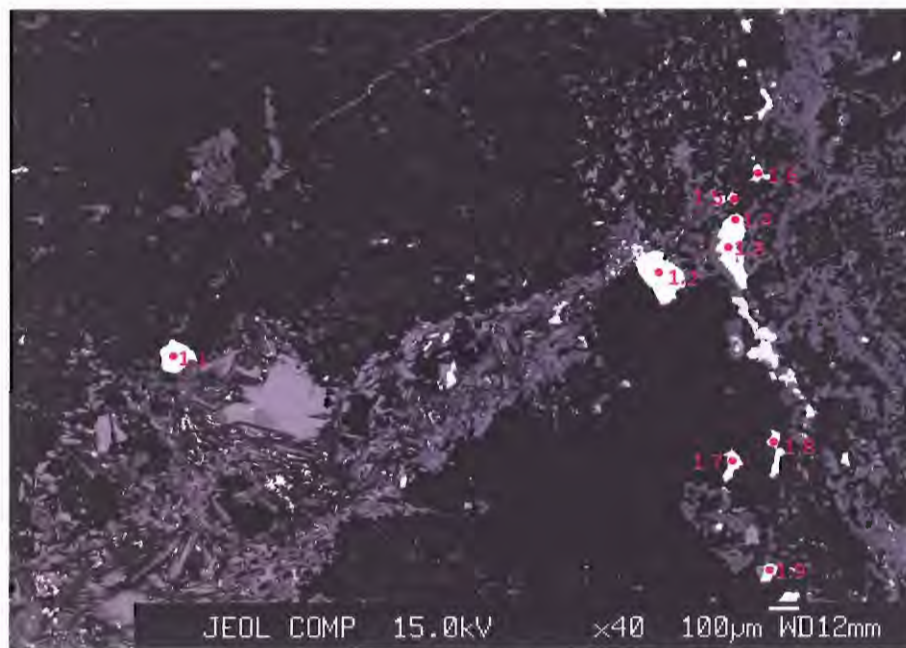
Mass Percent

No.	Mo	S	As	Fe	Pb	Ag	Co	Ti	Ca	Cu	Sn	Sb	Mn	Zn	W	Total	Mineral
2.1	0	52.0586	0.0511	44.0582	0.213	0.0462	0.1187	0.108	0.1285	0.281	0.1589	0.1272	0.107	0.4399	1.0512	98.9476	pyrite
2.2	0.0073	49.4516	0.0425	45.1898	0.1772	0.0521	0.1548	0.1081	0.0453	0.2686	0.156	0.0865	0.1357	0.4379	0.957	97.2705	pyrite
2.3	0	51.0961	0.0269	42.0204	0.1927	0.0614	0.1934	0.0975	0.5997	0.2378	0.1464	0.1091	0.1134	0.3861	0.9376	96.2186	pyrite
2.4	0	50.7116	0.0556	43.3694	0.1273	0.0419	0.0801	0.1106	0.0161	0.2624	0.1758	0.1004	0.1311	0.4106	0.9913	96.5843	pyrite
2.5	0	52.6324	0.0428	44.1574	0.2437	0.0521	0.1421	0.0923	0.0091	0.3587	0.1537	0.1122	0.1079	0.4531	1.014	99.5716	pyrite
2.6	0	52.1416	0.0484	44.0801	0.2162	0.055	0.1428	0.1236	0.0188	0.2972	0.1631	0.1224	0.1127	0.4546	0.9401	98.9167	pyrite
2.7	0	52.2434	0.0454	44.0068	0.1779	0.055	0.1508	0.1041	0.027	0.2736	0.1718	0.097	0.1157	0.4009	0.8298	98.6993	pyrite
2.8	0	52.315	0.0437	44.0905	0.1919	0.0391	0.1022	0.1021	0.0144	0.2362	0.1592	0.0903	0.1293	0.4149	0.8925	98.8214	pyrite
2.9	0	52.4284	0.0669	44.3483	0.2264	0.0575	0.1829	0.1072	0.0199	0.2616	0.1557	0.0977	0.114	0.4654	0.9473	99.4793	pyrite
2.10	0	52.1503	0.0467	43.9622	0.2058	0.0483	0.1914	0.1013	0.0354	0.225	0.1867	0.1081	0.1148	0.4449	1.0207	98.8417	pyrite
2.11	0.0273	52.3065	0.0446	44.0718	0.2195	0.0517	0.2236	0.0971	0.0254	0.2636	0.1333	0.0927	0.1197	0.4041	0.9863	99.0672	pyrite
2.12	0	51.7876	0.0129	44.2807	0.1324	0.0023	0	0.0485	0.0193	0	0	0	0	0.0243	0	96.3081	pyrite

Atomic Ratio

No.	Mo	S	As	Fe	Pb	Ag	Co	Ti	Ca	Cu	Sn	Sb	Mn	Zn	W	Total	Mineral
2.1	0	66.4542	0.0279	32.2848	0.0421	0.0175	0.0824	0.0922	0.1312	0.1809	0.0548	0.0427	0.0797	0.2754	0.234	99.9999	pyrite
2.2	0.0032	64.8037	0.0239	33.9938	0.0359	0.0203	0.1104	0.0948	0.0474	0.1776	0.0552	0.0299	0.1038	0.2814	0.2187	100	pyrite
2.3	0	66.7622	0.015	31.5169	0.039	0.0238	0.1375	0.0852	0.6268	0.1568	0.0517	0.0375	0.0865	0.2474	0.2136	99.9999	pyrite
2.4	0	66.3324	0.0311	32.5643	0.0258	0.0163	0.057	0.0968	0.0169	0.1732	0.0621	0.0346	0.1001	0.2634	0.2261	100	pyrite
2.5	0	66.6974	0.0232	32.1217	0.0478	0.0196	0.098	0.0783	0.0092	0.2293	0.0526	0.0374	0.0798	0.2816	0.2241	100	pyrite
2.6	0	66.5325	0.0265	32.2874	0.0427	0.0208	0.0991	0.1055	0.0192	0.1913	0.0562	0.0411	0.0839	0.2845	0.2092	99.9999	pyrite
2.7	0	66.6707	0.0248	32.2377	0.0351	0.0209	0.1047	0.0889	0.0276	0.1762	0.0592	0.0326	0.0861	0.2509	0.1847	100	pyrite
2.8	0	66.6938	0.0239	32.2659	0.0379	0.0148	0.0709	0.0871	0.0147	0.1519	0.0548	0.0303	0.0962	0.2594	0.1984	100	pyrite
2.9	0	66.522	0.0364	32.301	0.0444	0.0217	0.1262	0.091	0.0202	0.1674	0.0534	0.0326	0.0844	0.2896	0.2096	100	pyrite
2.10	0	66.5961	0.0255	32.2264	0.0407	0.0183	0.133	0.0866	0.0362	0.1449	0.0644	0.0364	0.0855	0.2786	0.2273	100	pyrite
2.11	0.0116	66.6127	0.0243	32.2183	0.0433	0.0196	0.1549	0.0828	0.0259	0.1693	0.0459	0.0311	0.089	0.2524	0.219	100	pyrite
2.12	0	67.0017	0.0072	32.8863	0.0265	0.0009	0	0.042	0.02	0	0	0	0	0.0154	0	100	pyrite

ZMMH-11 Sulphides (MM Quartz Porphyry)

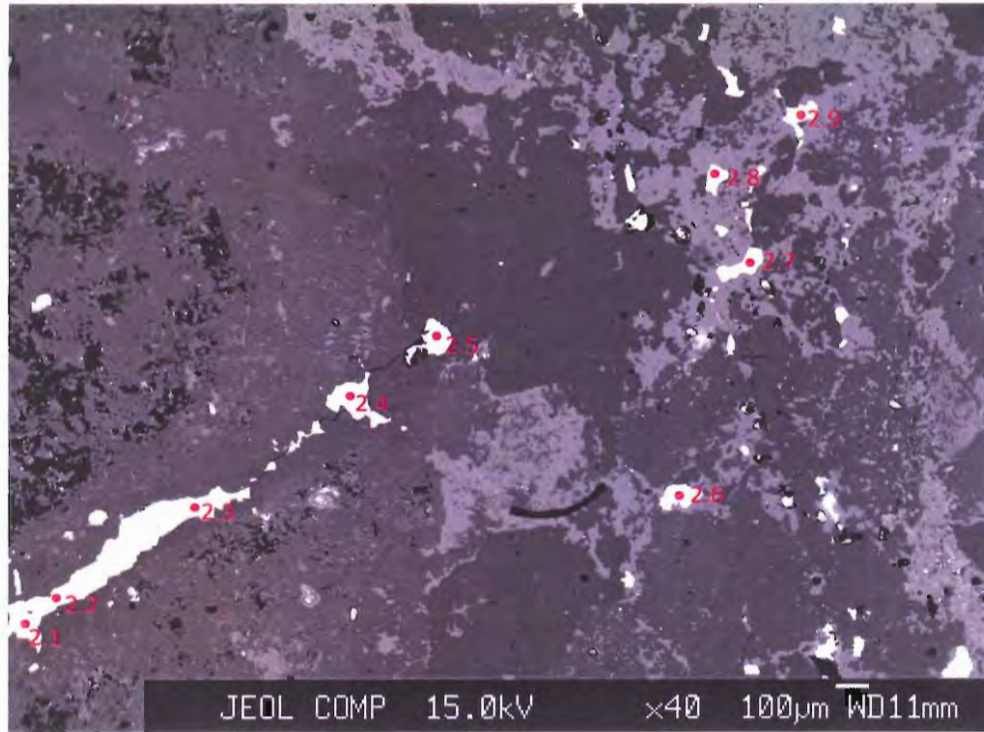


Mass Percent

No.	Mo	S	As	Fe	Pb	Ag	Co	Ti	Ca	Cu	Sn	Sb	Mn	Zn	W	Total	Mineral
1.1	0	33.7854	0	28.9219	0	0.038	0.0873	0.0029	0.0149	34.035	0.0145	0.0114	0	0.0835	0.1233	97.1181	chalcopyrite
1.2	0	33.9183	0	28.6683	0	0.0121	0.0723	0.0072	0.0074	34.1029	0.0392	0.0107	0	0.1273	0.0693	97.035	chalcopyrite
1.3	0	25.7118	0	11.0056	0	0.152	0.097	0.0432	0.0275	60.4177	0.0142	0.0319	0.0245	0.2496	0.405	98.1801	bornite
1.4	0	34.0818	0	28.7392	0	0.0436	0.0202	0.013	0.0034	33.8964	0	0	0.0078	0.1047	0.2588	97.1689	chalcopyrite
1.5	0	25.0454	0	10.3951	0	0.0292	0.151	0.0524	0.0211	61.9548	0.0202	0.0525	0.0241	0.247	0.4244	98.4173	bornite
1.6	0	25.4532	0	10.2796	0	0.0992	0.1506	0.0168	0.0205	62.0918	0.0605	0.056	0	0.2031	0.3668	98.7982	bornite
1.7	0	25.074	0	10.6574	0	0.1069	0.0234	0.0218	0.0162	62.7147	0.0292	0.0546	0.0352	0.2369	0.4634	99.4338	bornite
1.8	0	24.9874	0	10.2942	0.0861	0.1326	0.0637	0.0401	0.0281	62.2888	0.0473	0.0772	0.0214	0.2732	0.499	98.8392	bornite
1.9	0	25.1131	0	10.6564	0	0.129	0.0837	0.0321	0.0414	62.9525	0.0152	0.0291	0.0055	0.2262	0.5141	99.7983	bornite

Atomic Ratio

No.	Mo	S	As	Fe	Pb	Ag	Co	Ti	Ca	Cu	Sn	Sb	Mn	Zn	W	Total	Mineral
1.1	0	49.9047	0	24.5233	0	0.0167	0.0702	0.0028	0.0176	25.3622	0.0058	0.0044	0	0.0605	0.0318	100	chalcopyrite
1.2	0	50.0854	0	24.3007	0	0.0053	0.0581	0.0072	0.0088	25.4048	0.0156	0.0042	0	0.0922	0.0178	100	chalcopyrite
1.3	0	40.8918	0	10.0475	0	0.0718	0.0839	0.0459	0.035	48.4748	0.0061	0.0134	0.0227	0.1947	0.1123	99.9999	bornite
1.4	0	50.2564	0	24.3268	0	0.0191	0.0162	0.0128	0.004	25.2157	0	0	0.0067	0.0757	0.0666	100	chalcopyrite
1.5	0	39.9837	0	9.5263	0	0.0138	0.1311	0.056	0.027	49.8974	0.0087	0.0221	0.0224	0.1934	0.1182	100	bornite
1.6	0	40.3941	0	9.3647	0	0.0468	0.13	0.0179	0.026	49.7117	0.0259	0.0234	0	0.1581	0.1015	100	bornite
1.7	0	39.7104	0	9.6888	0	0.0503	0.0202	0.0231	0.0205	50.1069	0.0125	0.0228	0.0325	0.184	0.128	100	bornite
1.8	0	39.8373	0	9.4211	0.0212	0.0628	0.0552	0.0428	0.0358	50.0987	0.0204	0.0324	0.02	0.2136	0.1387	100	bornite
1.9	0	39.6463	0	9.6572	0	0.0605	0.0719	0.0339	0.0522	50.1376	0.0065	0.0121	0.0051	0.1751	0.1415	100	bornite

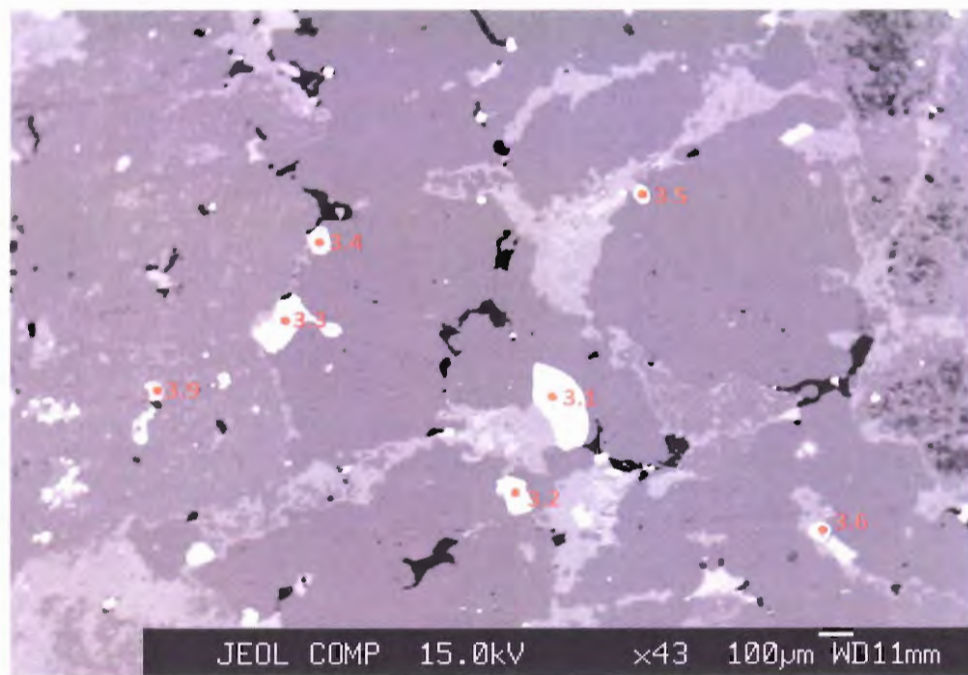


Mass Percent

No.	Mo	S	As	Fe	Pb	Ag	Co	Ti	Ca	Cu	Sn	Sb	Mn	Zn	W	Total	Mineral
2.1	0	25.4042	0	10.7624	0	0.1652	0.0701	0.0118	0.025	62.4988	0.0232	0.06	0.0226	0.2291	0.43	99.7025	bornite
2.2	0	8.557	0	10.3364	0	0	0.11	0	0	62.202	0	0	0.0405	0.2451	0.4299	81.921	bornite
2.3	0	25.3515	0	10.7277	0	0.1047	0.0364	0.0324	0.0305	62.6739	0.0281	0.066	0.0085	0.2449	0.4517	99.7563	bornite
2.4	0	25.2397	0	10.8631	0	0.0978	0.1441	0.0411	0.0284	62.7318	0.0856	0.0488	0.0059	0.2451	0.5742	100.106	bornite
2.5	0	18.9846	0	7.8408	0.6628	0.2747	0	0.0135	0.0653	45.4342	0.0284	0.0071	0.0141	0.1424	0.1127	73.5807	bornite
2.6	0	33.9644	0	28.8563	0.0322	0.0108	0	0.0035	0.0157	34.2971	0	0.0124	0.0039	0.09	0.0443	97.3307	chalcopyrite
2.7	0	31.9853	0	27.5755	0	0.0014	0.0082	0.0123	0.0014	33.116	0.0103	0.0424	0	0.0566	0.0719	92.8814	chalcopyrite
2.8	0	24.9419	0	10.7461	0	0.0649	0.1241	0.038	0.0371	62.8049	0.0603	0.0836	0.0115	0.2599	0.5137	99.6861	bornite
2.9	0	25.297	0	10.7407	0	0.1454	0	0.0338	0.0275	62.9608	0.0856	0.0593	0.0244	0.2072	0.3142	99.896	bornite

Atomic Ratio

No.	Mo	S	As	Fe	Pb	Ag	Co	Ti	Ca	Cu	Sn	Sb	Mn	Zn	W	Total	Mineral
2.1	0	40.0384	0	9.7369	0	0.0774	0.0601	0.0125	0.0315	49.6924	0.0099	0.0249	0.0208	0.1771	0.1182	100	bornite
2.2	0	18.5421	0	12.8572	0	0	0.1297	0	0	67.9968	0	0	0.0513	0.2605	0.1624	100	bornite
2.3	0	39.953	0	9.7049	0	0.049	0.0312	0.0341	0.0385	49.8287	0.012	0.0274	0.0078	0.1893	0.1241	100	bornite
2.4	0	39.7224	0	9.814	0	0.0457	0.1233	0.0432	0.0357	49.8067	0.0364	0.0202	0.0054	0.1892	0.1576	99.9999	bornite
2.5	0	40.6009	0	9.6257	0.2193	0.1746	0	0.0193	0.1116	49.019	0.0164	0.004	0.0176	0.1494	0.042	99.9999	bornite
2.6	0	50.0125	0	24.3913	0.0073	0.0047	0	0.0035	0.0184	25.4777	0	0.0048	0.0033	0.065	0.0114	100	chalcopyrite
2.7	0	49.5208	0	24.5076	0	0.0006	0.0069	0.0127	0.0017	25.8657	0.0043	0.0173	0	0.043	0.0194	100	chalcopyrite
2.8	0	39.4659	0	9.7607	0	0.0305	0.1068	0.0403	0.047	50.1341	0.0258	0.0348	0.0106	0.2017	0.1417	99.9999	bornite
2.9	0	39.8237	0	9.7061	0	0.068	0	0.0356	0.0346	50.0023	0.0364	0.0246	0.0224	0.16	0.0863	100	bornite



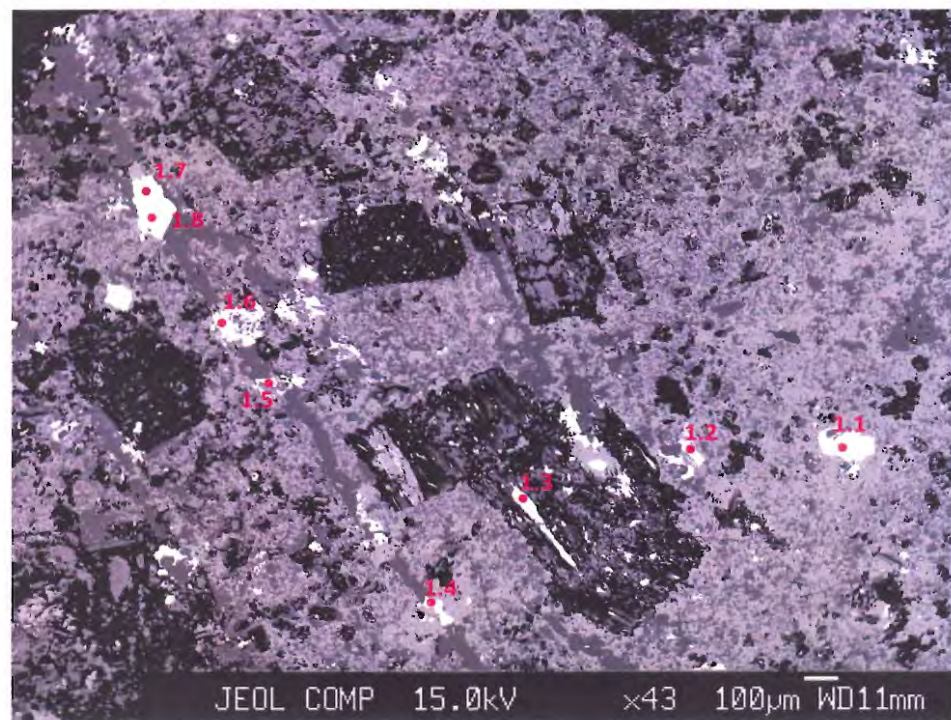
Mass Percent

No.	Mo	S	As	Fe	Pb	Ag	Co	Ti	Ca	Cu	Sn	Sb	Mn	Zn	W	Total	Mineral
3	0.0448	34.1344	0	28.7132	0.0139	0.0024	0	0.011	0.0145	34.3811	0	0	0.0097	0.1288	0.2284	97.6823	chalcopyrite
3.1	0	34.1069	0	29.2585	0	0.0008	0	0.0012	0.0027	34.442	0.0166	0	0	0.1414	0.1468	98.117	chalcopyrite
3.2	0	34.0181	0	28.9445	0	0.0108	0.1091	0.0045	0.0102	34.2642	0	0.0262	0.0144	0.0943	0.2394	97.7358	chalcopyrite
3.3	0	25.3308	0	10.8286	0	0.0726	0.0432	0.0155	0.0295	63.7064	0.0262	0.0426	0.0444	0.2357	0.3114	100.687	bornite
3.4	0	25.4149	0	10.6887	0.0031	0.0212	0	0.0411	0.0193	63.0256	0.0564	0.0644	0.0355	0.273	0.4525	100.096	bornite
3.5	0	34.2284	0	28.7625	0.0719	0.0092	0.0644	0.0214	0.0173	34.0917	0.036	0.0248	0	0.1249	0.2019	97.6545	chalcopyrite
3.6	0.2834	0.349	0.0733	0.4101	84.381	0.0875	0.6414	0.2866	0.4566	1.0202	0.392	1.1207	0.3282	1.3839	3.261	94.475	???
3.9	0.0749	23.4471	0.0546	4.4176	0.0886	0.1605	0.1973	0.1774	0.0604	73.0209	0.2181	0.1955	0.1518	0.8043	1.8528	104.922	bornite

Atomic Ratio

No.	Mo	S	As	Fe	Pb	Ag	Co	Ti	Ca	Cu	Sn	Sb	Mn	Zn	W	Total	Mineral
3	0.022	50.1188	0	24.2008	0.0032	0.0011	0	0.0108	0.017	25.4669	0	0	0.0083	0.0927	0.0585	100	chalcopyrite
3.1	0	49.8782	0	24.5618	0	0.0004	0	0.0011	0.0032	25.4099	0.0066	0	0	0.1014	0.0374	100	chalcopyrite
3.2	0	49.956	0	24.3997	0	0.0047	0.0872	0.0045	0.012	25.3844	0	0.0101	0.0123	0.0679	0.0613	100	chalcopyrite
3.3	0	39.5925	0	9.7157	0	0.0337	0.0367	0.0162	0.0369	50.2336	0.0111	0.0176	0.0405	0.1807	0.0849	100	bornite
3.4	0	39.9219	0	9.6381	0.0008	0.0099	0	0.0432	0.0243	49.9445	0.0239	0.0266	0.0326	0.2103	0.124	100	bornite
3.5	0	50.2415	0	24.235	0.0163	0.004	0.0514	0.021	0.0204	25.2448	0.0143	0.0096	0	0.0899	0.0517	99.9999	chalcopyrite
3.6	0.5554	2.0463	0.1839	1.3802	76.562	0.1525	2.0462	1.125	2.1415	3.0178	0.6208	1.7304	1.1232	3.9799	3.3345	100	???
3.9	0.039	36.5666	0.0364	3.9548	0.0214	0.0744	0.1674	0.1852	0.0753	57.4501	0.0919	0.0803	0.1381	0.6152	0.5039	100	bornite

ZMMH-14 Sulphides (MM Porphyry)

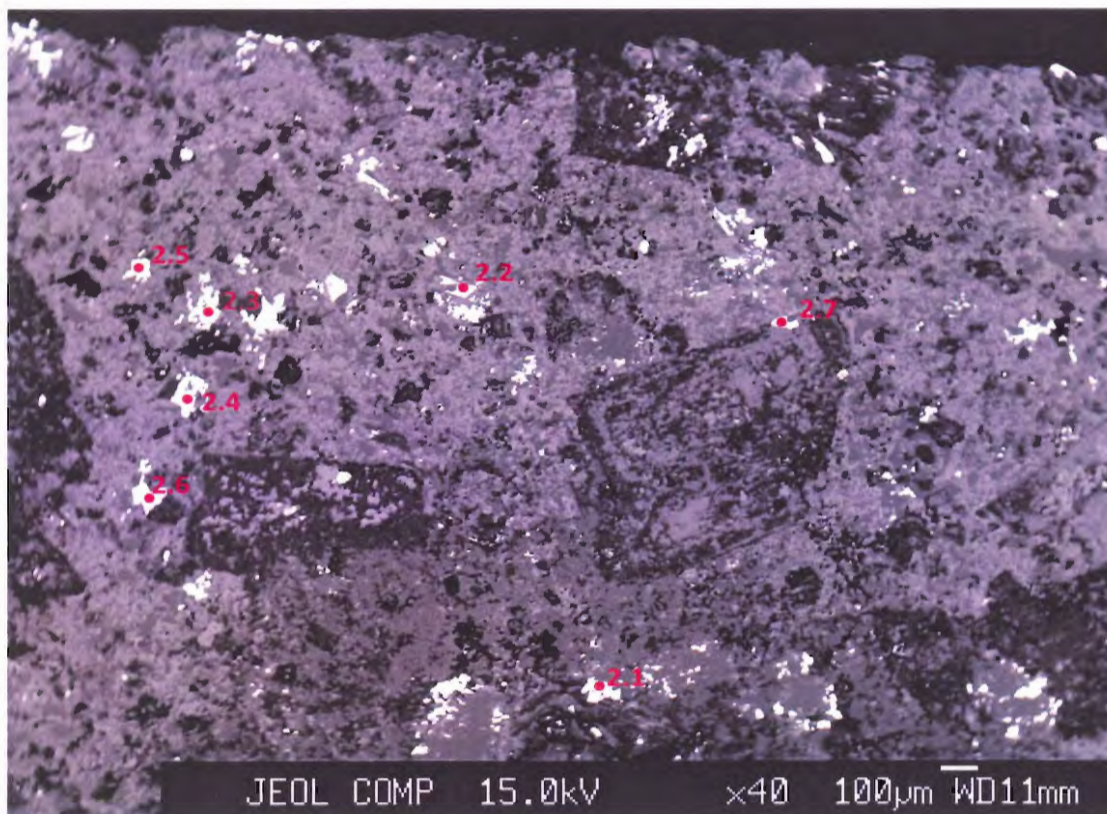


Mass Percent

No.	Mo	S	As	Fe	Pb	Ag	Co	Ti	Ca	Cu	Sn	Sb	Mn	Zn	W	Total	Mineral
1.1	0	35.1452	0.0108	28.8937	0	0.0659	0.0039	0.0412	0.0052	34.2039	0.0468	0.0823	0.0214	0.1353	0.3833	99.0389	chalcopyrite
1.2	0.0183	25.1428	0	10.5293	0	0.1344	0	0.0605	0.0497	62.1406	0.0538	0.1031	0.0414	0.3282	0.5793	99.1815	bornite
1.3	0	34.1697	0	28.8313	0	0.0483	0.0497	0.0523	0.0181	34.0941	0.0314	0.0626	0.0113	0.2485	0.2589	97.8763	chalcopyrite
1.4	0	34.2483	0	28.5306	0	0.0686	0.02	0.0257	0.024	34.5086	0.0262	0.0778	0.0295	0.2673	0.4148	98.2415	chalcopyrite
1.5	0.0019	25.4927	0.0271	10.2518	0	0.0706	0.0162	0.0643	0.0396	61.5839	0.065	0.0815	0.0278	0.375	0.6401	98.7375	bornite
1.6	0	25.52	3.4071	8.4763	0.0259	0.1866	0.0299	0.0615	0.0609	57.5389	0.051	0.1298	0.059	0.7099	0.521	96.7779	bornite
1.7	0.0061	25.9352	0	10.7544	0	0.1173	0.0022	0.0506	0.0379	62.6309	0.0891	0.0983	0.0504	0.2976	0.5607	100.631	bornite
1.8	0	34.524	0	29.0048	0	0.057	0	0.024	0.0102	34.6069	0.0387	0.0329	0.028	0.1473	0.3472	98.821	chalcopyrite

Atomic Ratio

No.	Mo	S	As	Fe	Pb	Ag	Co	Ti	Ca	Cu	Sn	Sb	Mn	Zn	W	Total	Mineral
1.1	0	50.7698	0.0067	23.9598	0	0.0283	0.0031	0.0399	0.006	24.9266	0.0182	0.0313	0.018	0.0958	0.0966	100	chalcopyrite
1.2	0.0097	39.9154	0	9.5955	0	0.0634	0	0.0643	0.0632	49.768	0.0231	0.0431	0.0384	0.2555	0.1604	100	bornite
1.3	0	50.0953	0	24.2639	0	0.021	0.0397	0.0514	0.0212	25.2164	0.0124	0.0242	0.0096	0.1787	0.0662	100	chalcopyrite
1.4	0	50.1066	0	23.9611	0	0.0298	0.0159	0.0251	0.0281	25.4701	0.0104	0.03	0.0252	0.1918	0.1058	100	chalcopyrite
1.5	0.001	40.5204	0.0184	9.354	0	0.0334	0.014	0.0684	0.0504	49.3826	0.0279	0.0341	0.0258	0.2924	0.1774	100	bornite
1.6	0	41.4567	2.3685	7.9042	0.0065	0.0901	0.0264	0.0669	0.0792	47.1546	0.0224	0.0555	0.056	0.5656	0.1476	100	bornite
1.7	0.0032	40.4414	0	9.6264	0	0.0544	0.0019	0.0528	0.0472	49.269	0.0375	0.0404	0.0458	0.2276	0.1525	100	bornite
1.8	0	50.1498	0	24.1856	0	0.0246	0	0.0233	0.0118	25.3605	0.0152	0.0126	0.0237	0.1049	0.088	100	chalcopyrite

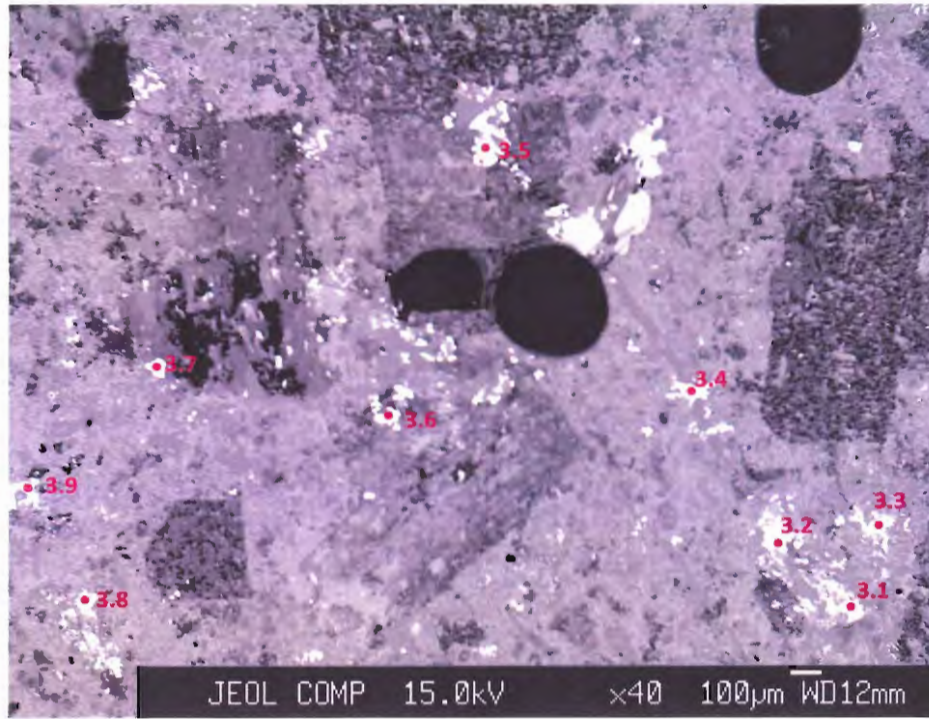


Mass Percent

No.	Mo	S	As	Fe	Pb	Ag	Co	Ti	Ca	Cu	Sn	Sb	Mn	Zn	W	Total	Mineral
2.1	0	34.1049	0.0064	28.0119	0	0.0628	0	0.0915	0.0405	33.429	0.0293	0.0909	0.0093	0.2095	0.4523	96.5384	chalcopyrite
2.2	0	34.2452	0.0011	28.4041	0	0.0295	0	0.0314	0.0429	33.8787	0.018	0.0525	0.019	0.2098	0.4007	97.3329	chalcopyrite
2.3	0	34.492	0.006	28.3696	0	0.0579	0	0.0122	0.0266	34.1822	0.0037	0.0595	0.0291	0.1963	0.3747	97.8099	chalcopyrite
2.4	0	34.2232	0.0164	28.7472	0	0.034	0	0.0422	0.0123	33.6642	0.0365	0.0723	0.0303	0.2342	0.3501	97.463	chalcopyrite
2.5	0	34.2973	0	28.1732	0.0464	0.0212	0.0421	0.0085	0.0287	33.4263	0.0642	0.0827	0.0105	0.1877	0.3639	96.7527	chalcopyrite
2.6	0	33.9901	0	28.0206	0	0.0556	0	0.0348	0.0238	33.1062	0	0.0768	0.0051	0.1873	0.3254	95.8258	chalcopyrite
2.7	0	34.2066	0	28.1391	0	0.0516	0.0274	0.0267	0.0305	33.3935	0.0109	0.046	0.0194	0.1909	0.3762	96.5189	chalcopyrite

Atomic Ratio

No.	Mo	S	As	Fe	Pb	Ag	Co	Ti	Ca	Cu	Sn	Sb	Mn	Zn	W	Total	Mineral
2.1	0	50.6134	0.0041	23.8634	0	0.0277	0	0.0909	0.048	25.0277	0.0117	0.0355	0.0081	0.1525	0.117	100	chalcopyrite
2.2	0	50.4283	0.0007	24.0102	0	0.0129	0	0.0309	0.0506	25.168	0.0072	0.0204	0.0163	0.1515	0.1029	99.9999	chalcopyrite
2.3	0	50.5258	0.0038	23.8555	0	0.0252	0	0.0119	0.0311	25.2606	0.0014	0.0229	0.0249	0.141	0.0957	99.9999	chalcopyrite
2.4	0	50.3386	0.0103	24.2726	0	0.0149	0	0.0416	0.0145	24.9803	0.0145	0.028	0.026	0.1689	0.0898	100	chalcopyrite
2.5	0	50.7366	0	23.9243	0.0106	0.0093	0.0339	0.0084	0.034	24.9458	0.0257	0.0322	0.0091	0.1362	0.0939	100	chalcopyrite
2.6	0	50.7261	0	24.0047	0	0.0247	0	0.0348	0.0284	24.9251	0	0.0302	0.0044	0.1371	0.0847	100	chalcopyrite
2.7	0	50.7034	0	23.943	0	0.0227	0.0221	0.0265	0.0362	24.9711	0.0044	0.018	0.0168	0.1388	0.0972	100	chalcopyrite



Mass Percent

No.	Mo	S	As	Fe	Pb	Ag	Co	Ti	Ca	Cu	Sn	Sb	Mn	Zn	W	Total	Mineral
3.1	0	34.3127	0.0294	28.1762	0	0.0304	0.042	0.0317	0.0209	33.6142	0.0067	0.0892	0.0256	0.2285	0.3623	96.9698	chalcopyrite
3.2	0	34.3256	0.012	28.5121	0.0101	0.0603	0.1013	0.0135	0.0301	33.9874	0.0437	0.0622	0.0039	0.1931	0.3485	97.7039	chalcopyrite
3.3	0	20.5339	0.1312	7.9407	0	0.1092	0	0.0279	0.0176	53.7881	0.0022	0.1677	0.0107	0.2814	0.309	83.3196	bornite
3.4	0	34.0149	0.0154	28.627	0	0.0467	0	0.0472	0.023	33.5406	0.0304	0.0463	0.0128	0.1679	0.3981	96.9704	chalcopyrite
3.5	0	34.0455	0	28.29	0	0.0431	0	0.0179	0.0273	33.6357	0.0283	0.0688	0	0.2433	0.3097	96.7097	chalcopyrite
3.6	0	26.4013	0.0131	11.0672	0	0.0705	0	0.0567	0.0351	60.5895	0.0368	0.1191	0.0316	0.3793	0.6593	99.4596	bornite
3.7	0	34.4408	0	28.4827	0	0.0491	0.0123	0.0544	0.0195	33.9806	0	0.0588	0.0276	0.2714	0.3626	97.7599	chalcopyrite
3.8	0	26.0869	0	10.7029	0	0.1123	0.1032	0.0533	0.0442	60.9184	0.0065	0.0941	0.0382	0.3542	0.6664	99.1807	bornite
3.9	0	34.273	0.0027	28.3445	0	0.0508	0	0.0226	0.0127	33.6845	0.0088	0.064	0.0288	0.243	0.3336	97.0691	chalcopyrite

Atomic Ratio

No.	Mo	S	As	Fe	Pb	Ag	Co	Ti	Ca	Cu	Sn	Sb	Mn	Zn	W	Total	Mineral
3.1	0	50.6516	0.0186	23.8759	0	0.0133	0.0337	0.0314	0.0246	25.0327	0.0027	0.0347	0.0221	0.1655	0.0933	100	chalcopyrite
3.2	0	50.3769	0.0076	24.0206	0.0023	0.0263	0.0808	0.0133	0.0354	25.1639	0.0173	0.0241	0.0033	0.139	0.0892	100	chalcopyrite
3.3	0	39.0439	0.1068	8.6673	0	0.0617	0	0.0355	0.0268	51.5961	0.0012	0.084	0.0119	0.2625	0.1025	100	bornite
3.4	0	50.304	0.0097	24.3024	0	0.0205	0	0.0467	0.0272	25.0237	0.0121	0.018	0.0111	0.1218	0.1027	100	chalcopyrite
3.5	0	50.4382	0	24.0587	0	0.019	0	0.0178	0.0324	25.1391	0.0113	0.0268	0	0.1767	0.08	100	chalcopyrite
3.6	0	41.3982	0.0088	9.9617	0	0.0329	0	0.0595	0.0441	47.9293	0.0156	0.0492	0.0289	0.2917	0.1803	100	bornite
3.7	0	50.474	0	23.9616	0	0.0214	0.0098	0.0533	0.0228	25.1231	0	0.0227	0.0236	0.195	0.0927	100	chalcopyrite
3.8	0	41.106	0	9.6811	0	0.0526	0.0885	0.0562	0.0557	48.426	0.0028	0.0391	0.0351	0.2738	0.1831	100	bornite
3.9	0	50.5561	0.0017	24.0011	0	0.0223	0	0.0224	0.015	25.0668	0.0035	0.0249	0.0248	0.1758	0.0858	100	chalcopyrite

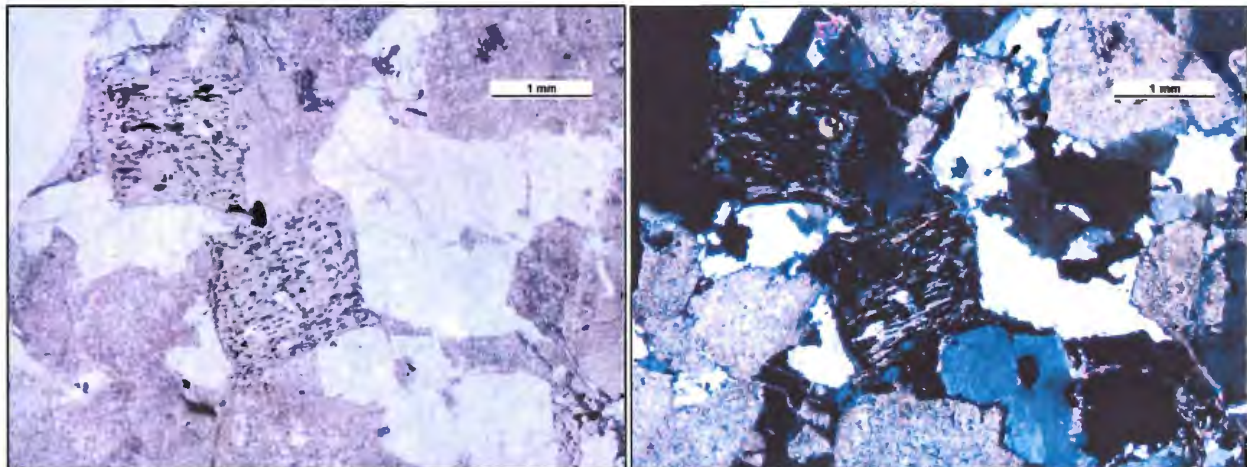
Appendix 2 Petrographic Descriptions

The following section contains petrographic descriptions of each core chip sample obtained from MMH (table 4.1). Photos of the core chip sample and photos taken with a microscopic camera are also included. The petrographic descriptions were done by observations using optical and reflected light microscopy.

ZMMH-4

DDH 3173 888.5-888.63 m

PHb (Hornblende Porphyry)



Observations

Rock Chip

The sample has a light-green to light-grey colour with a pink colouring surrounding the veins. There are also dark green hornblende phenocrysts as well as quartz phenocrysts.

Thin Section/Polished Section

The sample has a porphyritic texture with feldspar phenocrysts that have been entirely altered to sericite. These phenocrysts range in size from 2.5-0.5mm and are euhedral to subhedral prismatic. Hornblende phenocrysts are rectangular and have undergone chloritization. The hornblende is poikilitic and chloritized with inclusions of quartz, chalcopyrite and

molybdenite and occasionally zircon and apatite. Quartz phenocrysts are massive and anhedral with sutured boundaries. The quartz is interstitial between feldspar phenocrysts. Veins are common in the thin section. They are <0.25mm thick and mostly comprised of quartz and anhydrite with inclusions of chalcopyrite and chalcocite. The veins contain halos of chlorite.

Discussion

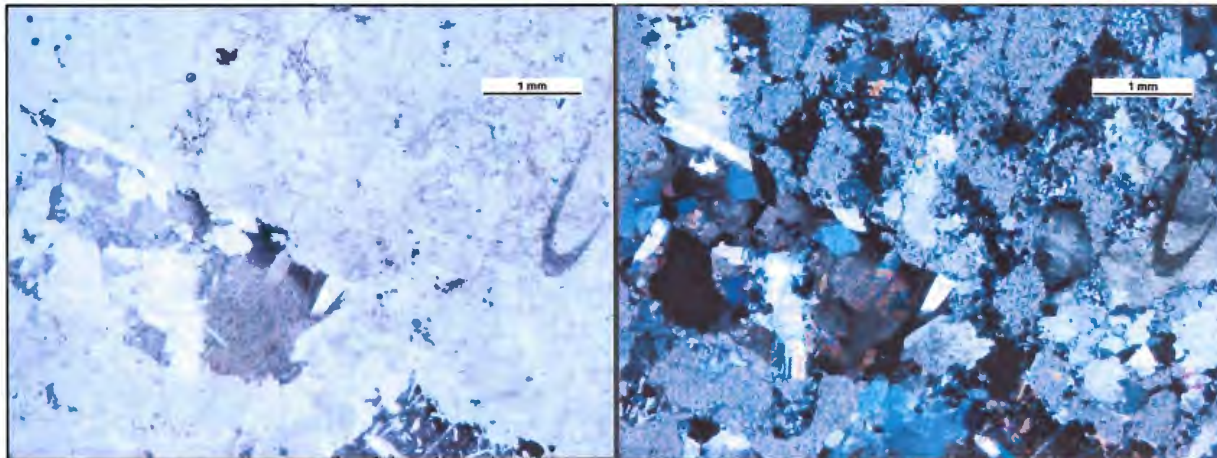
Due to the poikiloblastic nature of the hornblende phenocrysts, it is possible that they replace some other pre-existing mafic mineral. The chalcocite + chalcopyrite + molybdenite sulphide assemblage as well as the pink colour observed in the rock chip indicates that this sample has undergone potassic alteration. The presence of chalcopyrite and anhydrite in the veins indicate that the sample contains A- and/or B-type quartz veins. The latter is the most probably as there was no feldspar (or sericite) observed in the veins. The overprinting of feldspar phenocrysts by sericite and the light green colour of the rock chip indicate a later alteration phase such as chlorite-sericite alteration affected the sample.

ZMMH-5

DDH 3173 853.75-853.86m

Grd MM (MM Granodiorite)





Observations

Rock Chip

The sample has a dark- to medium-green colour as well as portions that are light-grey to light-pink. Quartz phenocrysts are visible as well as disseminated chalcopyrite and pyrite.

Thin Section/Polished Section

The sample has a generally equigranular, hypidiomorphic texture. Feldspars show a prismatic shape and have been almost entirely altered to sericite. Chloritized remnants of biotite and/or amphibole are poikiloblastic with inclusions of quartz, apatite, chalcopyrite and rutile. There is one evident vein that is 1.5 mm thick. It contains euhedral, prismatic quartz, anhydrite, sericite and pyrite. It does not appear to have a chloritic halo. The thin sections contain sulphides such as pyrite and chalcocite which are disseminated as well as a small amount of what appear to be galena (not confirmed).

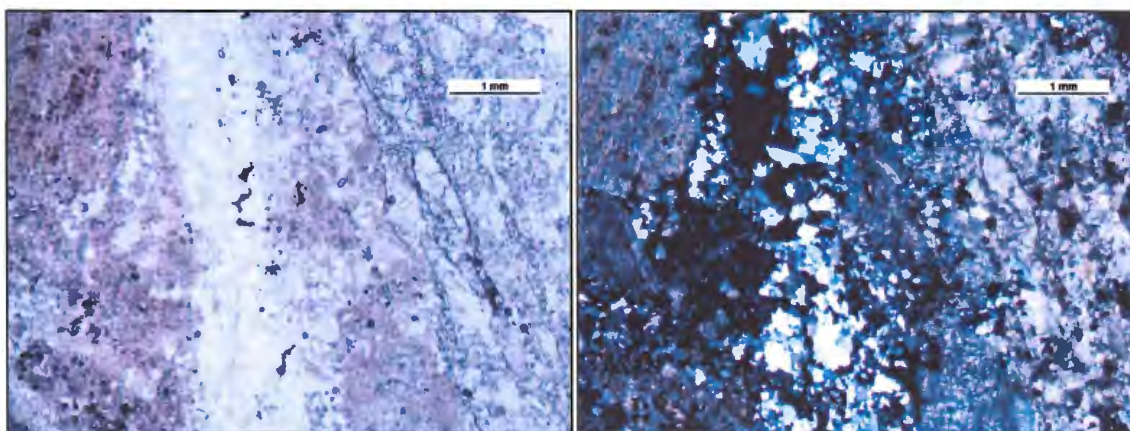
Discussion

The presence of chalcopyrite within the altered biotite and hornblende, the presence of chalcocite as well as the pink colour observed in the rock chip indicates that the sample had initially undergone potassic alteration. The sericitization of the feldspars, the chloritization of the hornblende/biotite and the pyrite indicates that the sample was later affected by a higher sulphidation alteration phase, likely chlorite-sericite. It is difficult to classify this vein as it has characteristics of both A- and B-type veins however; the euhedral quartz and central channel indicates that it is more of a B-type vein. It is likely that this vein was formed during conditions that were transitional from A-type quartz vein formation to B-type.

ZMMH-7

DDH 4903 1012.0-1012.25m

PMM (MM Porphyry)



Observations

Rock Chip

The sample is light- to dark- grey with a carbonate vein on the edge.

Thin Section/Polished Section

The sample is vuggy and porphyritic with feldspar phenocrysts that have been partially sericitized. These feldspars are plagioclase and are up to 4 mm in length. There are also large quartz phenocrysts with embayed grain boundaries. Less commonly, there are smaller phenocrysts (1 mm in length) of biotite that has been thoroughly chloritized. These phenocrysts are poikilitic with inclusions of carbonate, euhedral quartz, chalcopryrite and chalcocite. Apatite and rare zircon crystals typically appear within close proximity of these chloritized biotite

phenocrysts. Occasionally they appear elsewhere in the sample. The groundmass is fine-grained quartz with embayed and sub-grain boundaries, much like the quartz phenocrysts. There is an aplitic quartz vein that is 1.5 mm thick which contains chalcopryite and bornite. This vein does not have a chloritized halo. There is also a carbonate vein at the edge of the slide with an undeterminable thickness. Beyond the sulphides that occur in the vein, a small amount of disseminated chalcopryite, bornite and chalcocite occur in other regions of the thin section.

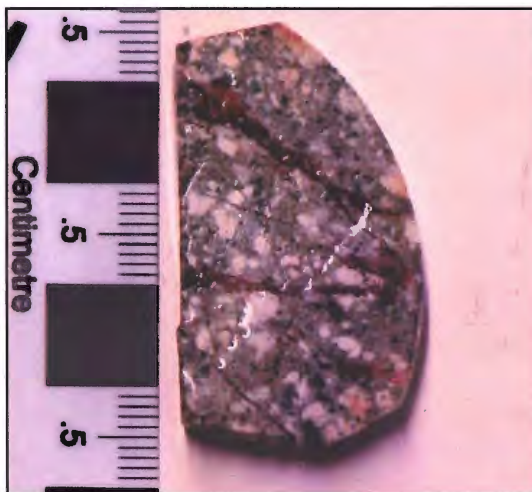
Discussion

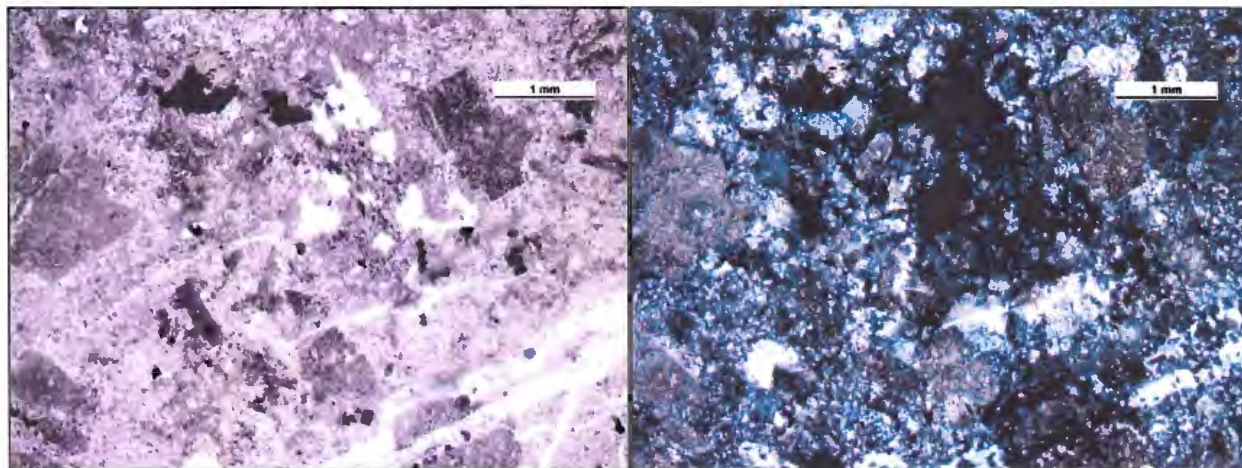
The chalcopryite-bornite assemblage within the aplitic vein indicates that it is likely an A-type quartz vein. The carbonate vein likely occurred much late with respect to the rest of the sample. The presence of plagioclase is indicative of preservation of primary feldspars or albitization of primary feldspars. If the latter is true than this is indication of potassic alteration. The partial sericitization and chloritization indicates that the sample later underwent less pervasive phyllic alteration.

ZMMH-9

DDH 4884 158.9-159.0m

PMM (MM Porphyry)





Observations

Rock Chip

The sample is porphyritic with a green-grey groundmass and white feldspar phenocrysts. Iron oxide is visible in the veins.

Thin Section/Polished Section

The sample is porphyritic and has undergone pervasive veining. The feldspar phenocrysts have been entirely sericitized and in some cases have an alteration rim around the outside. One of the feldspar phenocrysts show blurred microcline twinning suggesting it was originally plagioclase. These phenocrysts range in size from 4.5 mm to 1 mm. The groundmass is aplitic and contains quartz and sericite.

The sample is vuggy and has undergone pervasive veining. The thickness of the veins is ≤ 1 mm. Many open spaces occur within the veins. The veins also contain prismatic quartz, carbonate and iron-oxide.

Sulphides are disseminated throughout the sample. The most common sulphide is chalcopyrite followed by chalcocite, pyrite and rare instances of bornite.

The sample contains no chloritized biotite, apatites or zircons.

Discussion

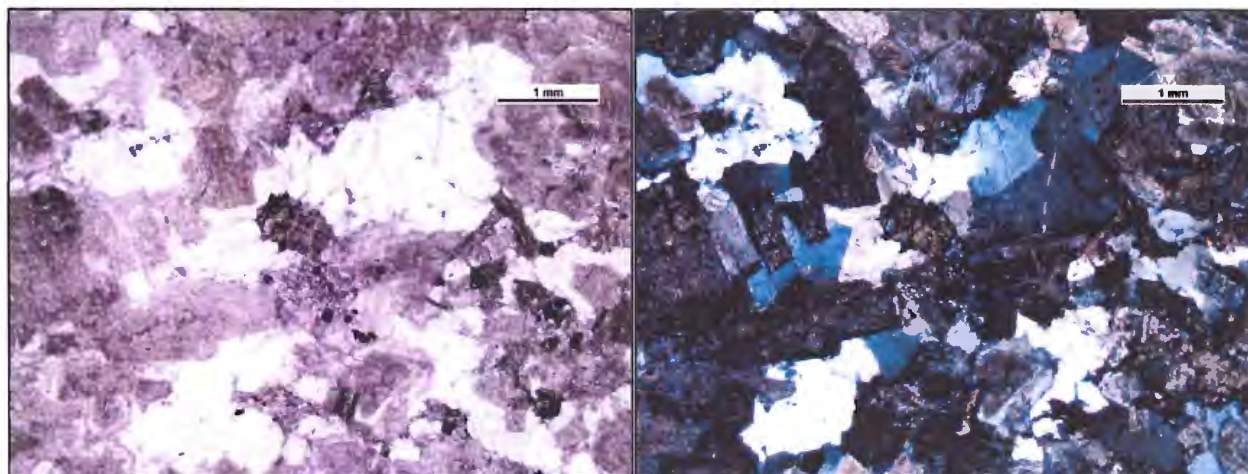
The chalcopyrite-chalcocite-bornite sulphide assemblage indicates that the sample originally underwent potassic alteration. The sericitization of the feldspars and pyrite indicate that the sample later underwent phyllic alteration. The vuggy texture, carbonate, iron oxide within the veins and the shallow depth of the sample indicate advanced weathering has occurred;

this observation is consistent with the shallow provenance of the sample in the drill-hole (159 m). The spaces that occur within the veins as well as the euhedral quartz, carbonate and iron-oxide indicate these veins are D-type quartz veins. The chalcopyrite + bornite sulphide assemblage could have been in part contributed to by the presence of these veins.

ZMMH-10

DDH 5014 161.0-161.2m

Grd MM



Observations

Rock Chip

The rock is coarse-grained equigranular with white to light-green feldspar, dark green chlorite and quartz.

Thin Section

The sample is equigranular and hypidiomorphic. It contains euhedral to subhedral feldspars which have undergone light to moderate sericitization (with respect to other samples). Most of these feldspars are plagioclase, however there are a few K-feldspars. These feldspars are intergrown and a few show zoning patterns with sericitized centers. In some places there are remnants of biotite that have been chloritized. This chloritized biotite is poikilitic with inclusions of sulphides and prismatic quartz. Anhedral quartz occurs interstitially between feldspar and chloritized biotite. Veining is not pervasive and the veins that do exist < 0.2 mm. These veins are comprised of quartz. Carbonate exists in one region of the thin section.

The polished section had been carbon coated for use in the electron microprobe and therefore sulphides cannot be correctly identified under reflected light.

Discussion

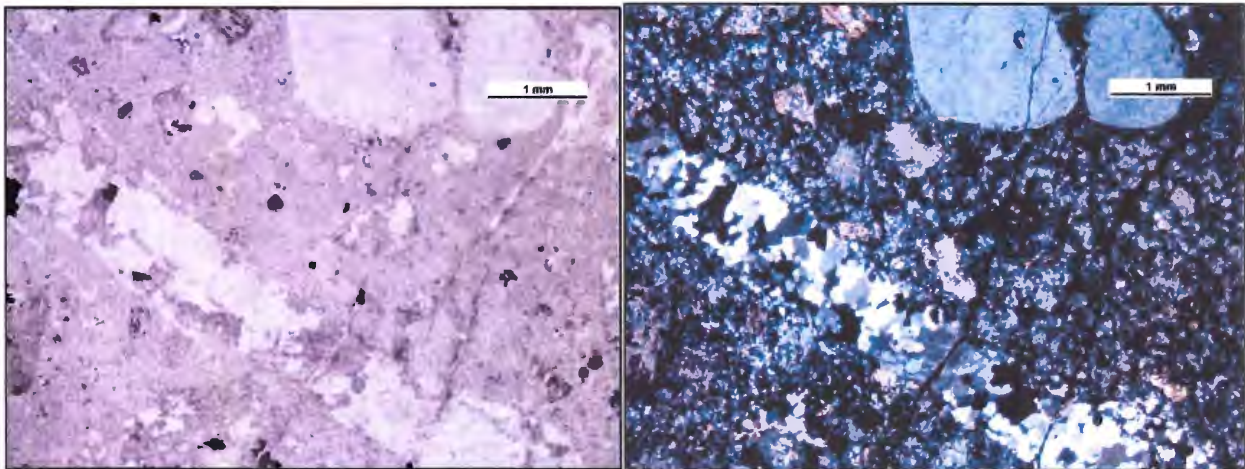
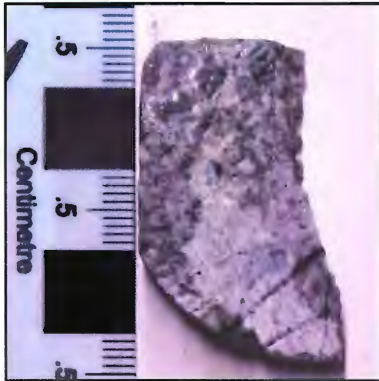
The classification of alteration in this sample is limited by the fact that sulphides are not observable due to use of the polished section in the electron microprobe. EMP results show that the plagioclase is albite and thus, there are two conclusions that could be drawn. The first is that the albite and K-feldspar are primary, the second is that the sample has been affected by potassic alteration whereby the primary feldspars were altered to K-feldspar and albite. Tobey (2004) states that albitization has not thoroughly affected MMH although the latter situation seems more likely than the former. Since the veining is not pervasive and is composed only of quartz than it is difficult to categorize them with respect to Gustafson & Hunt's (1975) quartz vein classification model.

The amount of sericitization and chloritization indicate that the sample has not gone pervasive phyllic alteration. Due to the shallow depth of the sample, the carbonate could have originated from potassic alteration or, more likely from weathering and precipitation processes.

ZMMH-11

DDH 4976 897.87 – 898.00 m

POQ MM



Observation

Rock Chip

Quartz phenocrysts are present in the rock chip as well as a milky white groundmass. Veining is visible with the veins containing vugs.

Thin Section:

The sample is porphyritic with phenocrysts of quartz and feldspar. The feldspar phenocrysts are zoned with the centers having been more heavily altered to quartz with the outsides containing more sericite. Sericitization is more prominent along cleavage planes. These feldspar phenocrysts are 0.5 mm to 4.25 mm. The quartz phenocrysts are spheroid with a diameter of 2.5-6mm. The groundmass aplitic and is mostly comprised of quartz and sericite (the sericite has presumably replaced feldspars). The veins are < 1 mm thick and comprised of

quartz and sulphides. All of the veins in the sample run parallel to one another. The sample is vuggy with most of the open spaces occurring along the veins. The sample does not contain chlorite, apatites or zircons.

The polished section had been carbon coated for use in the electron microprobe and therefore sulphides cannot be properly identified.

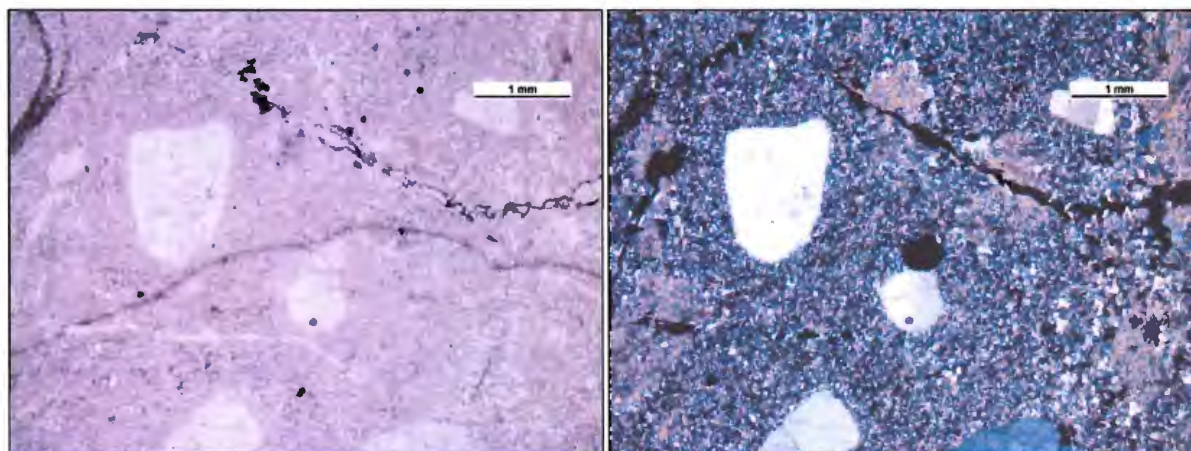
Discussion

The EMP results show that the feldspars of this section are K-feldspar indicating that the section has undergone potassic alteration. The sericitization of these K-feldspars indicate that the sample underwent phyllic alteration subsequent to the potassic alteration. The open spaces within the veins indicate that these veins are D-type quartz veins.

ZMMH-12

1128.0-1128.12 m

POQ MM (Quartz Porphyry)



Observations

Rock Chip

The sample is milky white with visible quartz phenocrysts. The veins contain most of the sulphides.

Thin Section/Polished Section

The sample is porphyritic with quartz and feldspar phenocrysts. The feldspar phenocrysts are < 1mm and have been completely sericitized. The quartz phenocrysts are spheroidal and <2.5 mm in diameter. The groundmass is aplitic and consists of quartz and sericite (presumably the sericite is result of altered feldspar). The veins occur in various directions with some cross-cutting each other. These veins contain empty spaces (vugs) and are comprised of pyrite and chalcopyrite. There is a quartz phenocryst that cuts through a vein (alternatively, the vein did not fracture the phenocryst). The sample contains a small amount of zircon and apatite (<10) as well as rutile.

Discussion

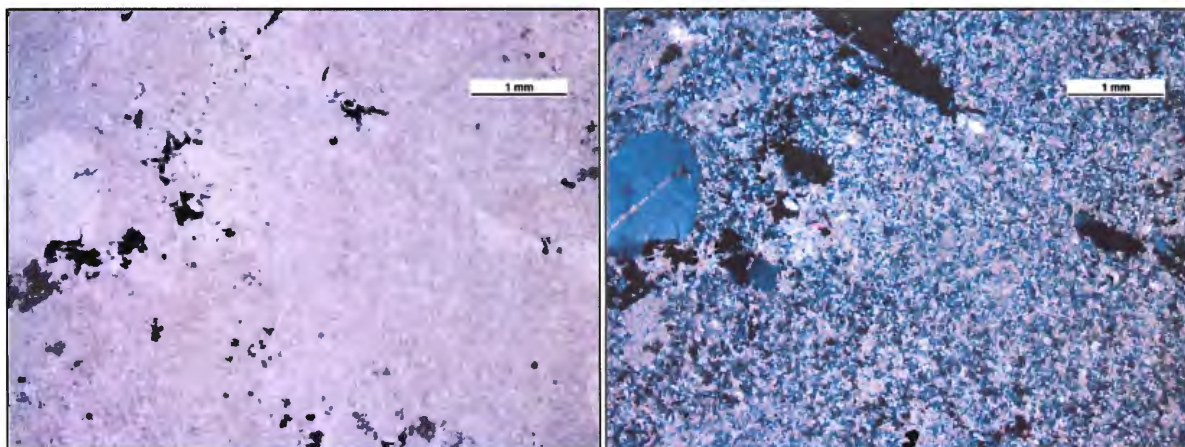
The sericitization of the feldspars indicates that the sample underwent phyllic alteration. It is possible that potassic alteration preceded the phyllic alteration but was completely overprinted. The quartz phenocrysts that seem to truncate the veins indicate that the formation of these phenocrysts occurred after the formation of the veins, or that the fractures where the veins were emplaced did not rupture the quartz crystals. The open spaces within the veins along with the chalcopyrite + pyrite sulphide assemblage indicates they are D-type veins.

ZMMH-13

DDH 4976 1123.78-1123.90 m

POQ MM (Quartz Porphyry)





Rock Chips

The sample is white with quartz phenocrysts and a milky groundmass. There are sulphides visible as well as dark green areas which contain sulphides.

Thin Section/Polished Section

The sample is porphyritic with quartz and feldspar phenocrysts. The feldspar phenocrysts are completely sericitized and <2.5 mm. The quartz phenocrysts are spheroidal and also <2.5 mm. Some of these quartz phenocrysts (“quartz eyes”) have internal sub-grain boundaries. The groundmass is aplitic and consists of quartz and sericite (presumably altered feldspar). The veins occur in many directions. They consist of quartz as well as pyrite and chalcopyrite and all contain empty spaces (vugs). The veins cut through the feldspar phenocrysts but the quartz phenocrysts are not affected by the veins. Pyrite is the most common sulphide with occasional chalcopyrite and bornite. The chalcopyrite that occurs forms a halo around pyrite. The sample also contains rutile. There were no apatites or zircons seen in the sample.

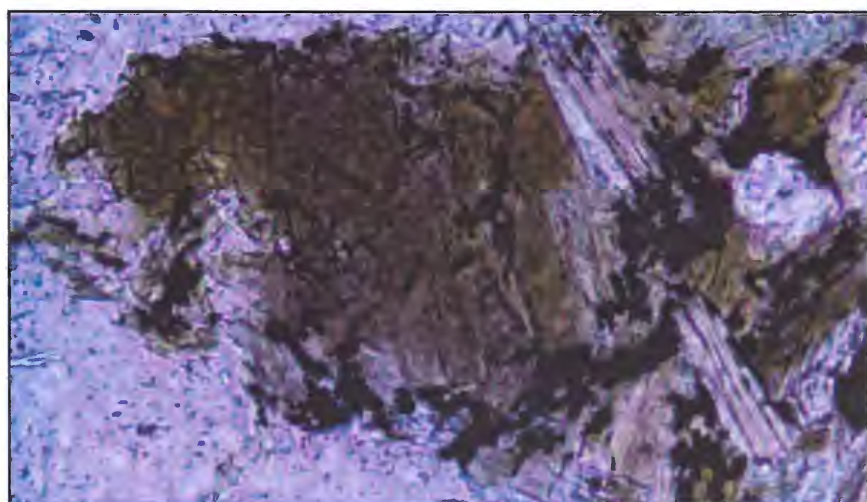
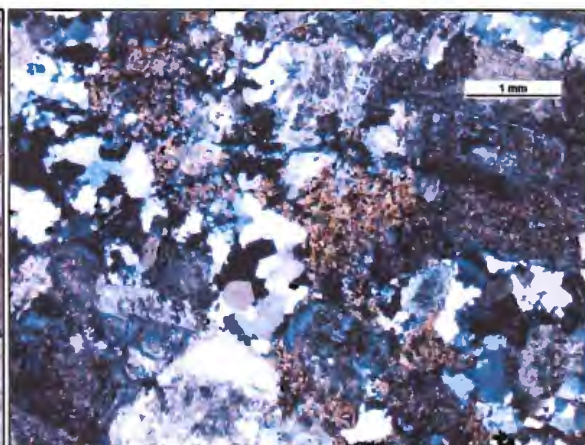
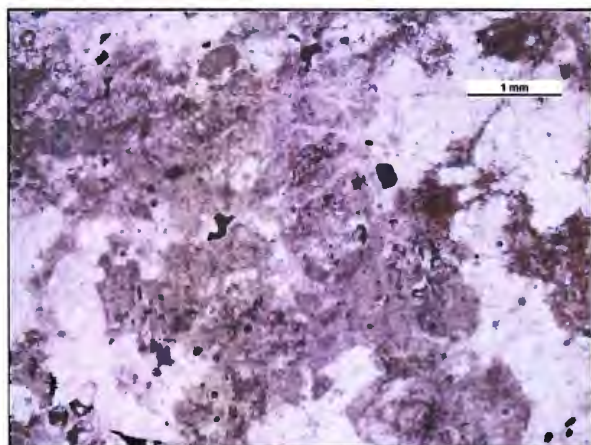
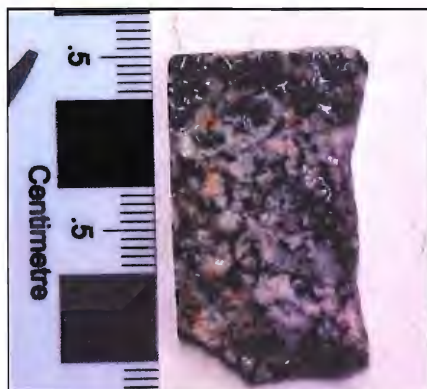
Discussion

The sericitization of the feldspar phenocrysts, the vuggy texture as well as the presence of pyrite indicates that the sample underwent phyllic alteration. It is possible that potassic alteration preceded the phyllic alteration but was entirely overprinted. The apparent truncation of veins by quartz phenocrysts indicates that the phenocrysts post-date the veins, or the fracturing did not crack the quartz crystals. The open spaces within the veins as well as the sulphide assemblage of pyrite + chalcopyrite indicate that they are D-type quartz veins.

ZMMH-14

DDH 4976 1241.00-1241.12m

PMM



Observations

Rock Chips

The rock chip has a milky-white groundmass with quartz phenocrysts, pink feldspar phenocrysts as well as dark phenocrysts visible. There is also iron-oxide staining

Thin Section

The sample is vuggy and porphyritic with plagioclase and K-feldspar phenocrysts. The plagioclase phenocrysts are euhedral to subhedral with alteration rims around the outside. The size of the phenocrysts ranges from 5 mm to 2.5 mm. The K-feldspar phenocrysts are subhedral to euhedral and also have alteration rims as well as penetrative alteration veins. The size of these phenocrysts is approximately 2 mm. These feldspar phenocrysts have undergone moderate sericitization (relative to other samples). There are vugs which occur within these phenocrysts. The interstitial minerals are biotite and quartz. The quartz is anhedral with embayed boundaries. In some places it looks as though quartz has entirely replaced the feldspar phenocryst (see photo) and the quartz has subgrains. The biotite has undergone minimal chloritization and occurs as pockets of accumulated fine-grained elongated grains. Some of these biotites contain microlath inclusions of exsolved rutile ("sagenitic"). These crystals occur at angles of 60°-120° (see photo above). The veins are 0.25 mm-0.5 mm thick and are mostly comprised of quartz with some carbonate and sulphides. These veins show an aplitic texture in some places. The sulphides occur mostly within the veins but also within the interstitial quartz and occasionally within the feldspar phenocrysts.

The polished section had been carbon coated for use in the electron microprobe and therefore sulphides cannot be correctly determined.

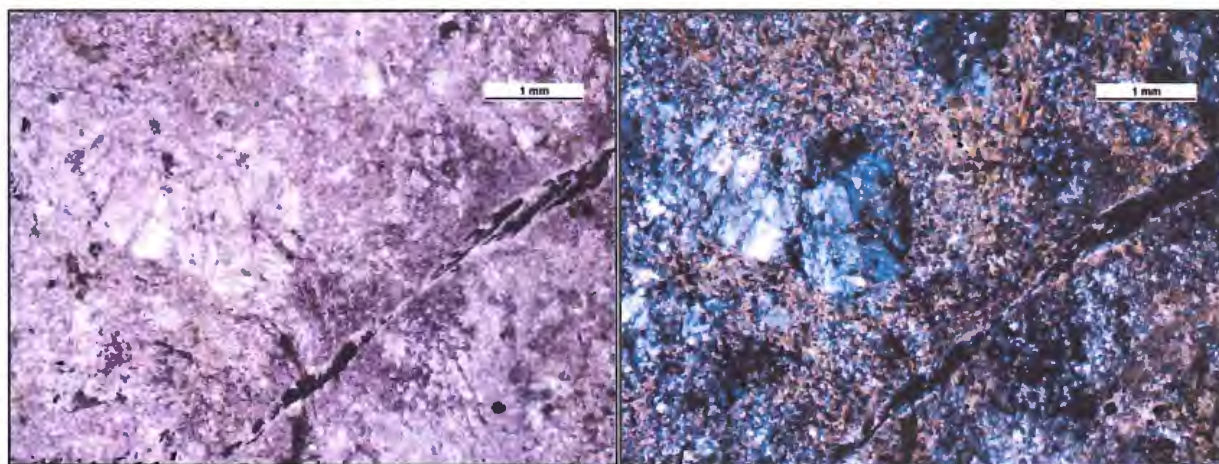
Discussion

The occurrence of K-feldspar and albite (as indicated by EMP data) as well as carbonate indicate that the sample has undergone potassic alteration. The phyllic alteration was likely confined to the veins and halos observed in the feldspar phenocrysts. The sagenitic biotite suggests that the secondary titanium minerals were exsolved during alteration.

ZMMH-15

DDH 4976 1286.47-1286.60 m

PMM



Observations

Rock Chips

The samples has greenish tint with black portions of biotite as well as visible feldspar phenocrysts that are light green. There are visible veins that have a pinkish colour.

Thin Section/Polished Section

The sample is porphyritic and has been heavily sericitized. There are few K-feldspar and plagioclase phenocrysts which still show relic characteristics. These phenocrysts are 2.5 mm and 1.5 mm respectively. These phenocrysts are subhedral. The groundmass is comprised of quartz, biotite and sericite. The biotite is shreddy and occurs in concentrated regions as subparallel and elongate grains. Veins are 0.5 to 1 mm thick and are comprised of carbonate, quartz and chalcopyrite.

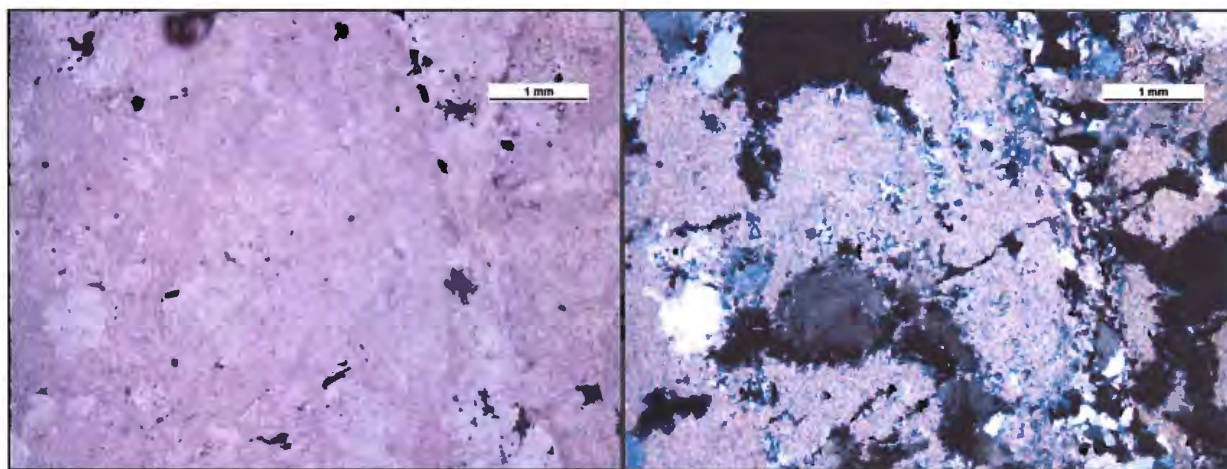
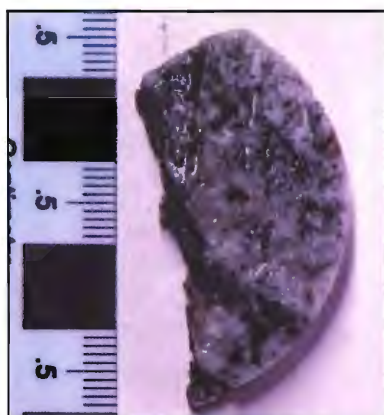
Discussion

The presence of K-feldspar as well as the pink coloured veins indicates that the sample has undergone potassic alteration. The sericitization of the feldspars as well as the green colour of the rock chips indicates that the sample underwent subsequent phyllic alteration. The shredded texture of the biotite may indicate it is secondary. The carbonate in the veins indicates that these are D-type quartz veins

ZMMH-17

DDH 4976 855.40 – 855.50 m

Grd MM



Observations

Rock Chips

The sample has a milk-white groundmass with visible quartz phenocrysts. Molybdenite can be identified within the veins as well as disseminated pyritic sulphides

Thin Section/Polished Section

The sample is vuggy and has been completely sericitized. There are quartz phenocrysts as well as quartz veins that contain molybdenite, chalcopyrite and pyrite. These veins run subparallel to one another. There is also chalcopyrite and pyrite disseminated throughout the sample.

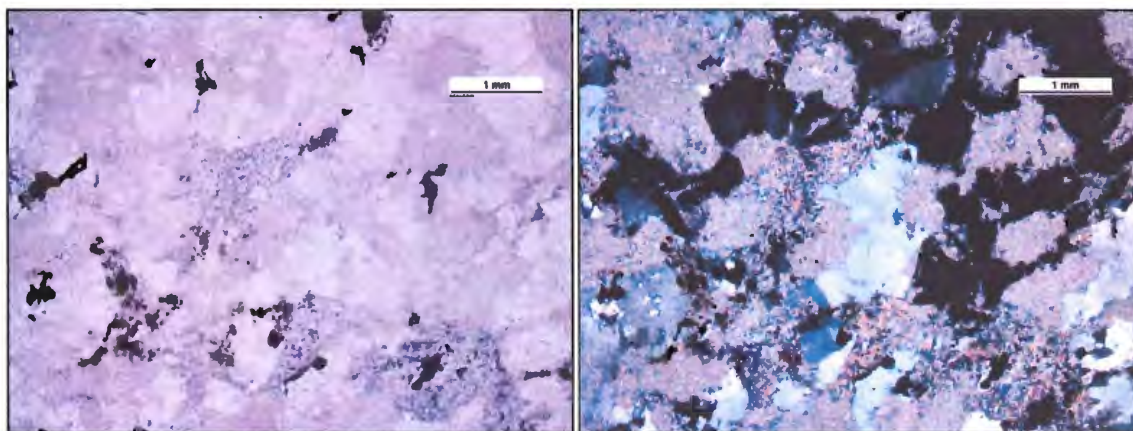
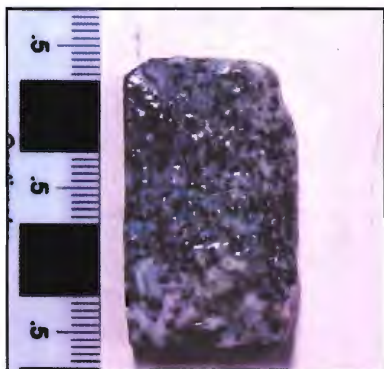
Discussion

The veins that contain molybdenite indicate that the veins are B-type quartz veins and that the sample underwent metasomatism with magmatic brine in the initial phases of alteration. The pervasive sericitization as well as the presence of pyrite indicates that the sample underwent phyllic alteration, likely sericitic. The presence of chalcopyrite indicates that perhaps the sample underwent potassic alteration at one time, but this cannot be proven from the observations.

ZMMH-18

DDH 4976 845.85-845.95 m

Grd MM



Observations

Rock Chips

The sample has a milky-white groundmass with quartz phenocrysts. Pyritic sulphides are visible as well a brownish mineral with staining which is probably rutile.

Thin Section/Polished Section

The sample is vuggy and has been completely sericitized. There are some quartz phenocrysts that are <1.5 mm. There are microlaths of rutile that occur in concentrated areas of the thin section. The sample does not have veining. Pyrite, bornite and chalcocite are disseminated throughout the sample.

Discussion

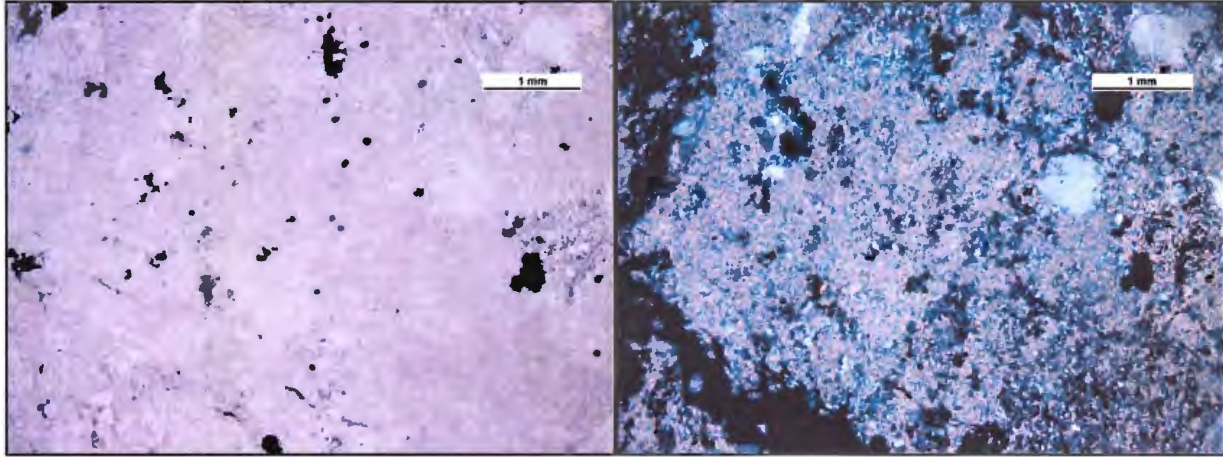
The presence of chalcocite indicates that the sample was strongly mineralized. The microlaths of rutile indicate that biotite was likely originally present in the rock, rutile having exsolved during alteration. The vuggy texture, complete sericitization and missing biotites indicate that the sample has undergone pervasive phyllic alteration, likely sericitic. The chalcopyrite and bornite could have originated in earlier alteration phases but it is difficult to confirm this.

ZMMH-20

DDH 4976 870.90 – 871.05 m

Grd MM





Observations

Rock Chip

The sample is vuggy with many spaces occurring within a vein. The groundmass is milky-white and quartz phenocrysts are visible. Sulphides appear to be disseminated throughout the sample as well as a brown mineral that stains the sample which is likely rutile.

Thin Section/Polished Section

The sample has been completely sericitized with remnant quartz phenocrysts. The sample is vuggy with spaces occurring disseminated throughout the sample as well as along sub parallel veins. The sulphides that occur are pyrite, chalcopyrite and bornite and are dispersed throughout the sample. Rutile occurs within concentrated areas within the sample as well.

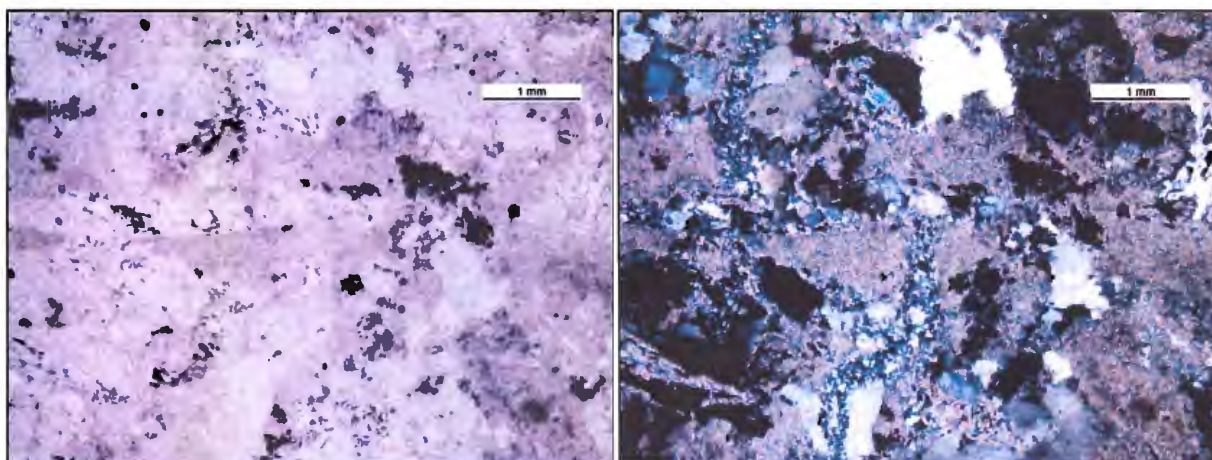
Discussion

The vuggy texture, the complete sericitization as well as the pyrite indicates that the sample underwent phyllic alteration, likely sericitic. The presence of bornite and chalcopyrite could be indicators of previous potassic alteration although these sulphides assemblages do occur in phyllic alteration zones as well. The presence of rutile indicates that biotite was initially present but was likely altered beyond recognition.

ZMMH-21

DDH 4976 883.20 – 883.24 m

PMM



Observations

Rock Chips

The sample has quartz phenocrysts as well as light-green phenocrysts and much smaller dark green areas. Sulphides appear dispersed throughout the sample.

Thin Section/Polished Section

The sample is vuggy and has been completely sericitized. The sample contains quartz phenocrysts. There are two generations of veins with a younger, medium grained quartz vein cutting an aplitic quartz vein (see photo). Chlorite occurs in some regions indicating chloritization of biotite or hornblende. The sulphides that occur are chalcopyrite, pyrite and chalcocite and are disseminated throughout the sample.

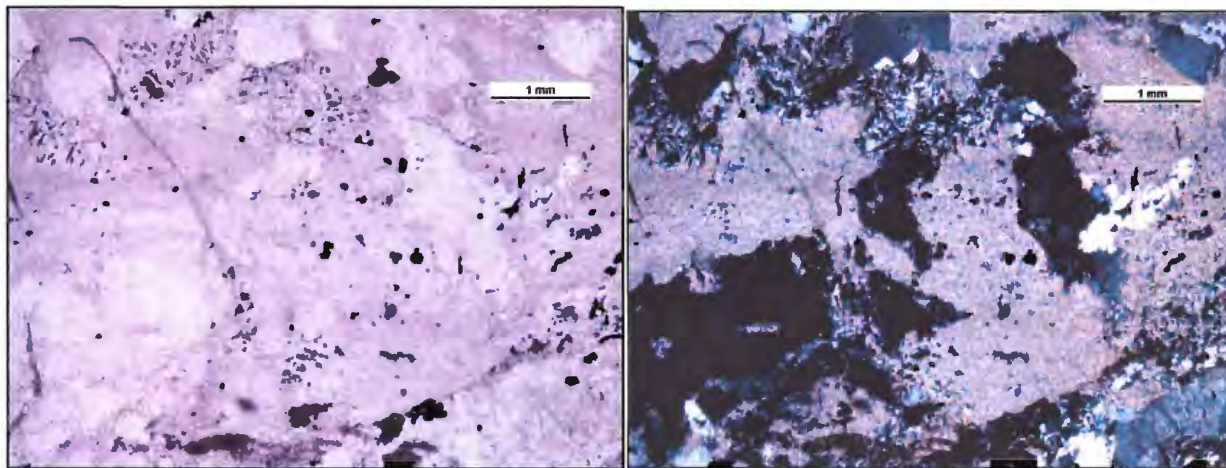
Discussion

The appearance of chalcocite suggests that the sample was heavily mineralized. The complete sericitization and chloritization, vuggy texture and presence of pyrite indicate the sample later underwent phyllic alteration, possibly both chlorite-sericite and sericitic alteration. The aplitic vein that is displaced is likely an A-type quartz vein with the younger, coarser grained vein likely a B-type vein.

ZMMH-23

DDH 4976 1064.00 – 1064.15 m

Grd MM



Observations

Rock Chips

The sample has phenocrysts of quartz, feldspar and a green mineral, likely chlorite. The sample has a vuggy texture, especially close to the veins.

Thin Section/Polished Section

The sample is vuggy and heavily altered with the feldspars completely sericitized and the biotite and/or hornblende completely chloritized. The remnants of the biotite/hornblende are poikiloblastic with inclusions of sulphides. There is one vein in the sample that is 1.25 mm thick and comprised of mainly quartz with some sulphides as well. The sulphides that occur in the sample are chalcopyrite, chalcocite and pyrite. The chalcopyrite has in some places undergone alteration to chalcocite. The sulphides that do not occur in the veins are disseminated throughout the sample.

Discussion

The occurrence of chalcocite indicates that the sample was heavily mineralized. The vuggy texture, sericitization and chloritization as well as the presence of pyrite indicates that the sample subsequently underwent phyllic alteration, possibly both sericitic and chlorite-sericite type. The Chalcopyrite could have origins in potassic and/or phyllic alteration phases. The vugs within the vein as well as the pyrite + chalcopyrite sulphide assemblage is indicative that it is a D-type quartz vein.

Appendix 3 Activation Laboratories Quality Control

The following table outlines the tests for quality control conducted by Activation Laboratories in Ancaster, Ontario where the whole-rock geochemical analysis was performed.

Analyte Symbol	Tl	Tm	Yb	Lu	Mass	Ga	Pb	Sn	Nb	Rb
Unit Symbol	ppm	ppm	ppm	ppm	g	ppm	ppm	ppm	ppm	ppm
Detection Limit	0.05	0.005	0.01	0.002		5	5	5	1	2
Analysis Method	FUS-MS	FUS-MS	FUS-MS	FUS-MS	INAA	PPXRF	PPXRF	PPXRF	PPXRF	PPXRF
GXR-1 Meas										
GXR-1 Cert										
WMG-1 Meas		0.171	1.08	0.152						
WMG-1 Cert		0.2	1.3	0.21						
NIST 694 Meas										
NIST 694 Cert										
DNC-1 Meas	< 0.05	0.313	1.98	0.288						
DNC-1 Cert	0.026	0.38	2.01	0.32						
BIR-1 Meas						17	< 5	< 5	< 1	3
BIR-1 Cert						16	3	0.65	0.6	0.25
GBW 07113 Meas										
GBW 07113 Cert										
MICA-FE Meas						98	25	85	295	2190
MICA-FE Cert						95	13	70	270	2200
GXR-4 Meas										
GXR-4 Cert										
GXR-2 Meas	0.69	0.264	1.74	0.249						
GXR-2 Cert	1.03	0.3	2.04	0.27						
SDC-1 Meas										
SDC-1 Cert										
SCO-1 Meas										
SCO-1 Cert										
GXR-6 Meas										
GXR-6 Cert										
NIST 1633b Meas										
NIST 1633b Cert										
SY-3 Meas						28	150	13	155	214
SY-3 Cert						27	133	6.5	148	206
BE-N Meas						17	6	8	104	60
BE-N Cert						17	4	2	105	47
AC-E Meas						42	41	14	108	157
AC-E Cert						39	39	13	110	152
W-2a Meas	< 0.05	0.329	2.09	0.29						
W-2a Cert	0.2	0.38	2.1	0.33						
OREAS 13P Meas										
OREAS 13P Cert										
SY-4 Meas										
SY-4 Cert										
CTA-AC-1 Meas			10.6	1.07						
CTA-AC-1 Cert			11.4	1.08						
BIR-1a Meas	< 0.05	0.264	1.65	0.239						
BIR-1a Cert	0.01	0.26	1.65	0.26						
NCS DC86312 Meas										
NCS DC86312 Cert										
VS-N Meas						387	> 1000	810	741	825
VS-N Cert						400	1000	800	700	800
NCS DC70014 Meas		0.53	3.41	0.477						
NCS DC70014 Cert		0.57	3.3	0.5						
NCS DC70009 (GBW07241) Meas	2.15	2.43	16.4	2.34						
NCS DC70009 (GBW07241) Cert	1.8	2.2	14.9	2.4						
OREAS 100a (Fusion) Meas		2.34	15	2.1						
OREAS 100a (Fusion) Cert		2.31	14.9	2.26						
OREAS 101a (Fusion) Meas		2.97	18.2	2.51						
OREAS 101a (Fusion) Cert		2.9	17.5	2.66						
JR-1 Meas	1.28	0.696	4.65	0.692						
JR-1 Cert	1.56	0.67	4.55	0.71						
ZMMH 15Q Orig	0.84	0.203	1.41	0.234						
ZMMH 15Q Dup	0.86	0.201	1.38	0.238						
ZMMH 17Q Orig										
ZMMH 17Q Dup										
Method Blank Method Blank					1					
Method Blank Method Blank						< 5	< 5	< 5	< 1	< 2
Method Blank Method Blank										

Appendix 4 Whole-rock Geochemical Data for MMH Samples

The following section contains the data table for the whole-rock geochemical analysis performed on the MMH and Chuquicamata samples by Activation Labs in Ancaster, Ontario.

MMH												
Analyte Symbol	SiO2	Al2O3	Fe2O3(T)	MnO	MgO	CaO	Na2O	K2O	TiO2	P2O5	LOI	Total
Unit Symbol	%	%	%	%	%	%	%	%	%	%	%	%
Detection Limit	0.01	0.01	0.01	0.001	0.01	0.01	0.01	0.01	0.001	0.01		0.01
Analysis Method	FUS-ICP	FUS-ICP	FUS-ICP	FUS-ICP	FUS-ICP	FUS-ICP	FUS-ICP	FUS-ICP	FUS-ICP	FUS-ICP	FUS-ICP	FUS-ICP
ZMMH 04Q	64.18	15.29	2.34	0.293	1.02	2.16	0.36	9.51	0.336	0.16	4.5	100.1
ZMMH 05Q	64.77	15.58	2.8	0.54	1.5	1.57	0.09	5.67	0.326	0.14	4.76	97.74
ZMMH 07Q	67.73	15.42	2.02	0.219	0.89	1.83	2.9	5.54	0.315	0.15	3.61	100.6
ZMMH 09Q	65.17	14.57	3.45	1.89	1.04	0.23	0.14	8.16	0.347	0.12	4.47	99.59
ZMMH 10Q	67.1	15	3.53	0.201	1.4	1.97	5.09	2.19	0.382	0.1	3.39	100.4
ZMMH 11Q	69.55	15.63	1.87	0.02	0.58	0.16	0.16	7.53	0.247	0.1	2.73	98.58
ZMMH 12Q	79.01	12.71	1.02	0.015	0.36	0.05	0.09	3.82	0.166	0.04	1.97	99.27
ZMMH 13Q	77.17	11.79	3.37	0.012	0.22	0.05	0.08	3.22	0.162	0.03	3.12	99.23
ZMMH 14Q	68.36	15.98	1.98	0.055	1.78	2.19	4.47	3.26	0.407	0.12	2.22	100.8
ZMMH 15Q	54.74	21.73	3.06	0.084	3.3	1.95	4.58	4.61	0.422	0.11	4.2	98.78
ZMMH 17Q	71.57	14.8	2.55	0.017	0.75	0.18	0.09	4.59	0.343	0.1	3.04	98.02
ZMMH 18Q	72.46	16.36	3.57	0.006	0.05	0.06	0.09	1.77	0.409	0.06	4.49	99.32
ZMMH 20Q	69.95	15.55	3.35	0.012	0.51	0.08	0.08	4.46	0.374	0.07	3.78	98.23
ZMMH 21Q	69.79	15.54	2.39	0.027	1.1	0.18	0.1	5.56	0.367	0.09	3.1	98.24
ZMMH 23Q	66.48	17.2	3.18	0.205	2.8	0.16	0.09	4.85	0.358	0.09	3.78	99.21
Cu-1333 Banco 2	69.27	17.1	1.65	0.01	0.32	1.91	4.96	2.74	0.273	0.04	0.96	99.22
Cu-2022 Este 2	68.65	16.18	2.49	0.03	0.61	2.87	4.68	3	0.276	0.09	0.34	99.23

MMH

Analyte Symbol	Au	Ag	As	Ba	Be	Bi	Br	Cd	Co	Cr	Cs	Cu
Unit Symbol	ppb	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm
Detection Limit	1	0.5	1	1	1	0.1	0.5	0.5	0.1	0.5	0.1	1
Analysis Method	INAA	INAA / TD-ICP	INAA	FUS-ICP	FUS-ICP	FUS-MS	INAA	TD-ICP	INAA	INAA	FUS-MS	TD-ICP
ZMMH 04Q	14	6.8	< 1	1262	2	1	< 0.5	3.7	2.2	6.7	2.3	5080
ZMMH 05Q	3	5.3	1	790	2	0.6	1.1	30	3.3	9.9	4.6	5220
ZMMH 07Q	5	1.2	4	842	2	< 0.1	< 0.5	0.9	2.7	6.9	7	3310
ZMMH 09Q	< 1	0.8	12	1815	1	< 0.1	< 0.5	6.9	8.3	12.7	6.9	118
ZMMH 10Q	< 1	< 0.5	15	455	1	< 0.1	< 0.5	3.6	12.1	12.3	7.2	33
ZMMH 11Q	21	2.6	118	922	2	0.7	< 0.5	0.5	1.7	5.1	2.6	7510
ZMMH 12Q	4	0.9	12	253	1	0.1	< 0.5	< 0.5	0.8	7.7	0.9	2770
ZMMH 13Q	13	3	8	130	< 1	0.2	< 0.5	< 0.5	3.3	10.5	5.1	6210
ZMMH 14Q	15	1.6	< 1	504	2	0.2	< 0.5	< 0.5	4.2	13.1	3	3610
ZMMH 15Q	24	0.6	2	588	3	0.2	< 0.5	< 0.5	7.3	11.4	11.1	4810
ZMMH 17Q	3	< 0.5	4	706	1	< 0.1	< 0.5	< 0.5	8.1	13.4	2.2	7170
ZMMH 18Q	70	2.6	51	79	< 1	0.5	1.2	< 0.5	7.2	22.6	1.2	7100
ZMMH 20Q	30	5.9	37	307	1	0.7	1.2	< 0.5	4	23.6	1.9	> 10000
ZMMH 21Q	9	1.4	118	404	2	< 0.1	1.3	< 0.5	2.4	16.2	4.2	9470
ZMMH 23Q	5	1.8	2	544	2	0.2	1.2	3.8	6.8	15.2	4	7030
Cu-1333 Banco 2	< 1	< 0.5	9	790	2	< 0.1	0.9	< 0.5	2.8	6	2.2	1560
Cu-2022 Este 2	< 1	< 0.5	2	791	2	< 0.1	< 0.5	< 0.5	3.6	5.1	0.7	306

MMH

Analyte Symbol	Ga	Ge	Hf	Hg	In	Ir	Mo	Nb	Ni	Pb	Rb	S
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppb	ppm	ppm	ppm	ppm	ppm	%
Detection Limit	1	0.5	0.1	1	0.1	1	2	0.2	1	5	1	0.001
Analysis Method	FUS-MS	FUS-MS	FUS-MS	INAA	FUS-MS	INAA	FUS-MS	FUS-MS	TD-ICP	TD-ICP	FUS-MS	TD-ICP
ZMMH 04Q	18	2.4	3	< 1	< 0.1	< 1	16	8	5	76	231	1.57
ZMMH 05Q	18	1.4	3.1	< 1	< 0.1	< 1	67	7.6	11	1970	227	1.48
ZMMH 07Q	17	1.8	3.7	< 1	< 0.1	< 1	21	9.4	4	73	132	0.28
ZMMH 09Q	14	3	2.9	< 1	< 0.1	< 1	< 2	4	14	374	220	1.69
ZMMH 10Q	15	0.9	3.2	< 1	< 0.1	< 1	< 2	4.5	19	369	58	1.06
ZMMH 11Q	17	4.3	3	< 1	< 0.1	< 1	> 100	6.3	6	12	191	0.714
ZMMH 12Q	10	1.6	2.6	< 1	< 0.1	< 1	28	5.2	2	< 5	88	0.355
ZMMH 13Q	8	1.4	2.5	< 1	< 0.1	< 1	27	6.9	5	6	92	2.53
ZMMH 14Q	18	2.3	3.6	< 1	< 0.1	< 1	2	5.6	16	< 5	78	0.219
ZMMH 15Q	29	2.2	4.2	< 1	< 0.1	< 1	7	5.2	23	< 5	163	0.408
ZMMH 17Q	16	1.9	3.1	< 1	< 0.1	< 1	> 100	3.7	23	< 5	133	1.58
ZMMH 18Q	9	1.8	3.5	< 1	< 0.1	< 1	18	4.7	17	14	46	3.15
ZMMH 20Q	21	1.9	3.5	< 1	< 0.1	< 1	20	65.1	11	< 5	114	3.03
ZMMH 21Q	15	4.4	3.1	< 1	0.1	< 1	60	3.7	6	5	220	0.917
ZMMH 23Q	19	1.6	3.1	< 1	0.3	< 1	2	4.9	20	< 5	159	1.46
Cu-1333 Banco 2	22	1.6	3.2	< 1	< 0.1	< 1	< 2	5.7	3	5	69	0.164
Cu-2022 Este 2	20	1.3	3.6	< 1	< 0.1	< 1	< 2	9.8	4	8	66	0.039

MMH

Analyte Symbol	Sb	Sc	Se	Sn	Sr	Ta	Th	U	V	W	Y	Zn
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm
Detection Limit	0.1	0.01	0.5	1	2	0.01	0.05	0.01	5	1	1	1
Analysis Method	INAA	INAA	INAA	FUS-MS	FUS-ICP	FUS-MS	FUS-MS	FUS-MS	FUS-ICP	INAA	FUS-ICP	INAA / TD-ICP
ZMMH 04Q	0.6	3.88	< 0.5	2	307	0.54	5.48	0.95	46	10	6	906
ZMMH 05Q	0.8	3.68	< 0.5	2	106	0.52	5.88	2.17	45	9	7	6210
ZMMH 07Q	0.5	3.73	< 0.5	1	244	0.7	9.68	1.7	41	5	6	227
ZMMH 09Q	2.5	6.66	< 0.5	< 1	137	0.31	4.36	1.34	54	13	9	1660
ZMMH 10Q	0.8	6.72	< 0.5	1	207	0.28	4.17	0.73	63	< 1	5	730
ZMMH 11Q	2.5	2.93	< 0.5	3	132	0.43	4.14	1.97	27	13	5	93
ZMMH 12Q	1.9	2.54	< 0.5	7	58	0.48	5.23	0.91	22	11	7	21
ZMMH 13Q	1.5	2.43	3.8	17	56	0.47	5.11	0.96	36	18	7	22
ZMMH 14Q	0.3	8.12	< 0.5	2	370	0.32	4.46	1.04	75	4	8	55
ZMMH 15Q	0.5	8.65	< 0.5	2	373	0.48	5.79	1.86	126	< 1	13	45
ZMMH 17Q	0.3	7.57	< 0.5	4	26	0.31	3.5	1.12	73	13	6	21
ZMMH 18Q	1.5	7.4	5.5	11	214	0.31	4.91	0.98	85	17	7	16
ZMMH 20Q	0.4	7.87	5.4	6	168	58.4	4.45	3.57	70	15	9	35
ZMMH 21Q	2.2	5.75	< 0.5	3	37	0.23	3.84	0.91	54	17	5	98
ZMMH 23Q	0.4	7.8	< 0.5	2	12	0.39	4.77	1.35	66	15	9	1250
Cu-1333 Banco 2	0.3	2.95	< 0.5	2	640	0.38	4.46	5.06	27	8	11	343
Cu-2022 Este 2	< 0.1	3.19	< 0.5	1	647	0.61	7.22	2.72	33	< 1	5	121

MMH

Analyte Symbol	Zr	La	Ce	Pr	Nd	Sm	Eu	Gd	Tb	Dy	Ho	Er
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm	ppm
Detection Limit	1	0.05	0.05	0.01	0.05	0.01	0.005	0.01	0.01	0.01	0.01	0.01
Analysis Method	FUS-MS	FUS-MS	FUS-MS	FUS-MS	FUS-MS	FUS-MS	FUS-MS	FUS-MS	FUS-MS	FUS-MS	FUS-MS	FUS-MS
ZMMH 04Q	107	21.4	41.4	4.21	15.5	2.59	0.641	1.81	0.24	1.23	0.23	0.64
ZMMH 05Q	111	24.5	45.6	4.34	15.1	2.44	0.804	1.56	0.21	1.12	0.21	0.63
ZMMH 07Q	126	25.7	47.7	4.69	16.5	2.65	0.666	1.76	0.24	1.27	0.24	0.7
ZMMH 09Q	106	18.4	35.4	3.49	12.3	2.17	0.733	1.65	0.25	1.47	0.29	0.87
ZMMH 10Q	120	17.3	32.2	3.15	11.5	1.88	0.581	1.36	0.2	1.1	0.22	0.63
ZMMH 11Q	110	17.1	32.8	3.3	11.6	1.98	0.526	1.3	0.18	1	0.19	0.55
ZMMH 12Q	84	15.6	28.6	2.69	9.37	1.59	0.439	1.34	0.21	1.19	0.24	0.74
ZMMH 13Q	80	19	36.2	3.42	11.4	1.76	0.379	1.14	0.18	1.13	0.24	0.75
ZMMH 14Q	139	17.1	32.4	3.16	11.9	2.14	0.618	1.64	0.26	1.45	0.29	0.85
ZMMH 15Q	141	31.5	58.9	5.66	20.1	3.63	1.01	2.87	0.42	2.3	0.44	1.31
ZMMH 17Q	112	12.6	24.9	2.47	8.94	1.48	0.325	1.24	0.2	1.21	0.25	0.76
ZMMH 18Q	133	20.7	40	3.9	14	2.56	0.627	1.75	0.23	1.3	0.27	0.89
ZMMH 20Q	130	19	35.4	3.35	11.6	2.01	0.499	1.66	0.24	1.46	0.28	0.83
ZMMH 21Q	122	16.4	29.7	2.89	10.4	1.84	0.581	1.42	0.21	1.13	0.23	0.68
ZMMH 23Q	116	17.4	33.3	3.22	11.8	2.09	0.404	1.72	0.28	1.57	0.31	0.91
Cu-1333 Banco 2	115	14.1	26.2	2.43	8.58	1.57	0.537	1.42	0.28	1.8	0.36	1.05
Cu-2022 Este 2	122	19.5	37.5	3.77	13.6	2.32	0.61	1.55	0.2	1.07	0.21	0.61

MMH

Analyte Symbol	Tl	Tm	Yb	Lu	Mass	Ga	Pb	Sn	Nb	Rb
Unit Symbol	ppm	ppm	ppm	ppm	g	ppm	ppm	ppm	ppm	ppm
Detection Limit	0.05	0.005	0.01	0.002		5	5	5	1	2
Analysis Method	FUS-MS	FUS-MS	FUS-MS	FUS-MS	INAA	PPXRF	PPXRF	PPXRF	PPXRF	PPXRF
ZMMH 04Q	1.56	0.096	0.64	0.107	1.362	17	73	9	6	244
ZMMH 05Q	1.94	0.095	0.64	0.108	1.2	23	> 1000	7	6	197
ZMMH 07Q	0.95	0.11	0.79	0.13	1.426	17	73	5	7	140
ZMMH 09Q	3.12	0.136	0.95	0.16	1.418	14	350	6	3	225
ZMMH 10Q	0.58	0.097	0.65	0.109	1.479	15	346	9	4	54
ZMMH 11Q	1.18	0.086	0.64	0.111	1.223	15	14	12	7	207
ZMMH 12Q	0.49	0.118	0.84	0.141	1.344	9	< 5	18	5	96
ZMMH 13Q	0.65	0.118	0.82	0.134	1.409	7	< 5	17	5	99
ZMMH 14Q	0.37	0.131	0.91	0.157	1.519	17	8	7	4	85
ZMMH 15Q	0.85	0.202	1.4	0.236	1.34	27	8	10	4	176
ZMMH 17Q	0.97	0.117	0.81	0.141	1.273	14	< 5	17	4	144
ZMMH 18Q	0.36	0.15	1.06	0.192	1.207	8	13	12	4	52
ZMMH 20Q	0.75	0.128	0.9	0.15	1.185	20	9	8	4	122
ZMMH 21Q	1.19	0.104	0.72	0.122	1.136	13	< 5	9	3	237
ZMMH 23Q	1.22	0.146	1.04	0.178	1.24	20	9	< 5	3	173
Cu-1333 Banco 2	0.21	0.166	1.14	0.188	1.439	21	10	9	5	79
Cu-2022 Este 2	0.12	0.095	0.67	0.116	1.669	20	8	< 5	7	76

Appendix 5 Data from Ballard (2001) Thesis

The following section contains data tables from Ballard (2001). This whole-rock geochemical data was used to compare the MMH data with the Fortuna and El Abra igneous complexes.

Los Picos

Analyte Symbol	SiO2	Al2O3	Fe2O3(T)	MnO	MgO	CaO	Na2O
Unit Symbol	%	%	%	%	%	%	%
98-518	60.6	16.6	5.88	0.09	2.81	5.03	3.75
98-519	62.6	16.4	5.42	0.09	2.42	4.74	3.77
98-520	58.7	17	6.66	0.07	3.11	5.54	3.73
98-522	59.6	16.5	6.54	0.11	3.24	5.8	3.71
98-524	60.9	16.5	6.21	0.1	2.86	5.27	3.74
98-525	68.1	14.7	3.76	0.06	1.33	2.84	3.29
98-526	57.1	17.8	6.74	0.11	3.6	6.04	4.03
98-527	61.7	16.7	6.13	0.09	2.45	4.87	4
98-540	66.7	15.2	4.26	0.08	1.54	3.22	3.36
98-549	61.4	16.9	5.65	0.07	2.52	4.6	3.91
98-564	65.8	15.5	4.25	0.07	1.68	3.66	3.68

Fortuna Gris

Analyte Symbol	SiO2	Al2O3	Fe2O3(T)	MnO	MgO	CaO	Na2O
Unit Symbol	%	%	%	%	%	%	%
98-532	64.4	16.5	4.39	0.05	1.81	4.19	4.13
98-534	65.7	15.2	4.85	0.07	1.72	3.44	3.25
98-544	61.2	17.1	5.89	0.1	2.81	5.18	3.77
98-548	73.1	14.1	1.8	0.02	0.37	1.29	3.4
98-550	67.9	15.1	3.67	0.05	1.23	2.88	3.48
98-551	63.5	16.8	4.93	0.06	2	4.43	4.08
98-552	62.6	16.9	5.4	0.09	2.31	4.9	4
98-553	63.7	16.7	4.77	0.06	2.07	3.63	3.97
98-559	64.4	16.4	4.37	0.07	1.83	4.19	4.04
98-561	66.7	16.1	3.7	0.05	1.39	3.69	4.06
98-566	66.7	16.9	4	0.06	1.44	3.78	4.1
98-576	63.4	16.7	4.83	0.07	2.21	4.65	4.11

Fortuna Clara

Analyte Symbol	SiO2	Al2O3	Fe2O3(T)	MnO	MgO	CaO	Na2O
Unit Symbol	%	%	%	%	%	%	%
98-535	64.8	17.3	3.72	0.06	1.52	3.7	4.56
98-536	65.3	16.8	3.8	0.08	1.68	3.96	4.42
98-538	66.6	16	3.57	0.06	1.43	3.69	4.27
98-541	65.9	16.9	3.08	0.04	1.11	3.23	4.9
98-547	67.1	16.6	3.3	0.07	1.17	3.21	4.78
98-554	65.6	16.9	3.76	0.03	1.5	3.89	4.63
98-556	66.4	16.4	3.95	0.04	1.45	3.11	4.4
98-567	67.2	16.7	0.47	0.06	1.09	3.96	4.65
98-570	68.2	15.7	3.46	0.02	1.26	2.97	4.02
98-572	65.5	17	3.91	0.05	1.47	3.85	4.34
98-575	66.9	16.4	3.11	0.02	1.29	3.54	4.66
98-577	68.4	15.4	2.95	0.04	1.67	3.35	4.1

San Lorenzo

Analyte Symbol	SiO ₂	Al ₂ O ₃	Fe ₂ O ₃ (T)	MnO	MgO	CaO	Na ₂ O
Unit Symbol	%	%	%	%	%	%	%
98-531	66.7	16.2	3.23	0.04	1.36	3.62	4.19
98-542	65.4	16.7	3.17	0.03	1.3	3.04	4.81
98-545	70.8	15	2.09	0.05	0.73	2.1	4.21
98-546	68.4	16	2.83	0.06	0.91	3.11	4.54
98-560	65.4	19.9	3.87	0	0.62	0.51	1.96
98-562	70.9	15.5	2.07	0.02	0.76	1.56	4.25
98-563	69.9	15.6	2.33	0.04	0.85	2.7	4.12
98-568	64.3	15.9	8.43	0.01	0.41	0.58	4.66
98-571	76.4	13.1	0.61	0.01	0.23	1.17	3.35
98-573	68.1	16.1	2.79	0.03	1.29	2.24	3.11

Fortuna Leucocratic Intrusion

Analyte Symbol	SiO ₂	Al ₂ O ₃	Fe ₂ O ₃ (T)	MnO	MgO	CaO	Na ₂ O
Unit Symbol	%	%	%	%	%	%	%
98-557	77.5	12.2	0.79	0.01	0.12	0.61	2.76
98-558	76.4	12.4	0.95	0.01	0.39	0.83	3.05
98-569	78.1	12	0.58	0.01	0.05	0.41	2.83

Pajonal diorite

Analyte Symbol	SiO ₂	Al ₂ O ₃	Fe ₂ O ₃ (T)	MnO	MgO	CaO	Na ₂ O
Unit Symbol	%	%	%	%	%	%	%
00-194	61	15.7	6.47	0.12	2.84	5.19	3.57
00-196	63.4	15.7	5.23	0.09	2.02	3.91	3.63
00-197	65	15.6	4.6	0.06	1.61	3.26	3.63
00-199	60.9	15.9	5.89	0.1	2.44	4.52	3.62
00-200	58.2	15.9	6.4	0.07	3.9	5.35	4.66

Central Diorite

Analyte Symbol	SiO ₂	Al ₂ O ₃	Fe ₂ O ₃ (T)	MnO	MgO	CaO	Na ₂ O
Unit Symbol	%	%	%	%	%	%	%
00-128	69.1	15.2	2.79	0.01	1.7	1.41	3.19
00-135	65.4	15.6	4.63	0.02	2.07	2.42	3.41
00-136	64.2	15.2	5.95	0.09	2.64	3.55	3.57
00-178	59.8	18.8	5.84	0.05	3.62	4.21	3.37
00-186	63.5	17.7	5.44	0.03	2.56	2.99	3.57

Dark Diorite

Analyte Symbol	SiO ₂	Al ₂ O ₃	Fe ₂ O ₃ (T)	MnO	MgO	CaO	Na ₂ O
Unit Symbol	%	%	%	%	%	%	%
00-145	57.5	19.2	7.99	0.05	3.58	4.57	3.16
00-160	54.4	18.1	0.86	0.14	4.65	6.67	3.64
00-181	66.4	15.4	4.3	0.02	1.77	2.13	3.46
00-184	56.3	19.3	6.76	0.06	5.2	3.88	4

Equis Monzodiorite

Analyte Symbol	SiO ₂	Al ₂ O ₃	Fe ₂ O ₃ (T)	MnO	MgO	CaO	Na ₂ O
Unit Symbol	%	%	%	%	%	%	%
00-141	66.1	15.3	4.12	0.04	2.3	2.17	3.51
00-146	65.3	17.5	4.97	0.01	1.84	2.01	3.35
00-151	70.9	14.7	1.69	0.01	0.84	1.54	3.61
00-155	69.4	14.8	3.22	0.04	1.15	2.13	3.49
00-173	69.1	15.7	3.11	0.01	1.74	1.1	3.05

South Granodiorite

Analyte Symbol	SiO ₂	Al ₂ O ₃	Fe ₂ O ₃ (T)	MnO	MgO	CaO	Na ₂ O
Unit Symbol	%	%	%	%	%	%	%
00-126	57.4	22.7	2.99	0.61	4.94	3.18	0.63
00-130	64.9	16.5	4.04	0.04	1.88	3.64	4.36
00-156	65.7	16.8	3.52	0.04	1.94	3.29	4.28
00-159	66.7	16.8	2.57	0.02	1.56	3.3	4.5
00-162	65.7	16.8	2.94	0.04	1.34	3.21	4.38

El Abra mine porphyry

Analyte Symbol	SiO ₂	Al ₂ O ₃	Fe ₂ O ₃ (T)	MnO	MgO	CaO	Na ₂ O
Unit Symbol	%	%	%	%	%	%	%
00-138	66.7	16.9	2.83	0.04	1.25	3.07	4.55
00-140	70.4	17	2.16	0.01	0.7	2.14	4.95
00-144	68	16.4	2.38	0.02	0.8	2.33	4.63
00-168	60.9	16.7	6.03	0.07	2.76	4.77	3.42
00-172	66.2	16.7	4.68	0.41	1.03	0.8	4.91

El Abra Apolo leucogranite

Analyte Symbol	SiO ₂	Al ₂ O ₃	Fe ₂ O ₃ (T)	MnO	MgO	CaO	Na ₂ O
Unit Symbol	%	%	%	%	%	%	%
00-120	70.9	15.3	2.55	0.02	1.02	1.11	2.66
00-125	70.3	15.2	1.53	0.01	1.03	1.39	4.48
00-154	71.9	14.5	1.18	0.03	1.11	1.47	3.32
00-171	69.3	15.1	2.93	0.01	1.15	2.23	3.9
00-187	72.2	14.7	1.39	0.01	1.03	0.74	3.47

El Abra Aplite

Analyte Symbol	SiO ₂	Al ₂ O ₃	Fe ₂ O ₃ (T)	MnO	MgO	CaO	Na ₂ O
Unit Symbol	%	%	%	%	%	%	%
00-124	70.6	14.8	2.58	0.01	0.9	1.28	3.31
00-150	71.8	14.2	1.99	0.01	0.83	0.96	2.98
00-158	69.7	14.7	3.29	0.02	1.25	1.94	3.58
00-169	74.5	13.6	1.05	0.01	0.62	0.4	2.93
00-183	72.2	13.7	1.34	0.01	0.9	0.52	3.22
00-188	72.9	14.2	1.93	0.01	0.69	0.65	3.31

Los Picos

Analyte Symbol	K2O	TiO2	P2O5	LOI	Total	Au	Ag
Unit Symbol	%	%	%	%	%	ppb	ppm
98-518	2.9	0.83	0.22		98.68		
98-519	3.16	0.79	0.18		99.53		
98-520	2.88	0.89	0.21		98.75		
98-522	2.8	0.85	0.18		99.37		
98-524	3.08	0.87	0.21		99.81		
98-525	4.87	0.6	0.11		99.66		
98-526	1.99	0.87	0.23		98.56		
98-527	3.18	0.85	0.22		100.09		
98-540	4.6	0.64	0.14		99.75		
98-549	2.97	0.8	0.2		98.96		
98-564	3.77	0.57	0.08		99.12		

Fortuna Gris

Analyte Symbol	K2O	TiO2	P2O5	LOI	Total	Au	Ag
Unit Symbol	%	%	%	%	%	ppb	ppm
98-532	2.95	0.58	0.19		99.14		
98-534	4.65	0.75	0.15		99.78		
98-544	2.88	0.75	0.19		99.82		
98-548	5.12	0.27	0.05		99.51		
98-550	4.46	0.49	0.13		99.42		
98-551	3.12	0.63	0.2		99.73		
98-552	2.88	0.67	0.24		99.97		
98-553	3.3	0.63	0.19		99.08		
98-559	3.16	0.59	0.18		99.28		
98-561	3.43	0.49	0.21		99.84		
98-566	3.38	0.58	0.19		101.19		
98-576	2.95	0.64	0.19		99.79		

Fortuna Clara

Analyte Symbol	K2O	TiO2	P2O5	LOI	Total	Au	Ag
Unit Symbol	%	%	%	%	%	ppb	ppm
98-535	2.73	0.51	0.2		99.14		
98-536	2.84	0.53	0.19		99.55		
98-538	3.23	0.45	0.18		99.45		
98-541	2.45	0.43	0.18		98.23		
98-547	2.88	0.42	0.24		99.82		
98-554	2.74	0.55	0.17		99.7		
98-556	2.82	0.51	0.18		99.26		
98-567	2.57	0.47	0.06		99.93		
98-570	3.51	0.52	0.14		99.73		
98-572	2.98	0.52	0.18		99.73		
98-575	3.02	0.42	0.16		99.59		
98-577	3.3	0.42	0.14		99.79		

San Lorenzo

Analyte Symbol	K2O	TiO2	P2O5	LOI	Total	Au	Ag
Unit Symbol	%	%	%	%	%	ppb	ppm
98-531	3.13	0.45	0.16		99.06		
98-542	2.67	0.42	0.19		97.78		
98-545	3.95	0.26	0.09		99.34		
98-546	3.32	0.37	0.15		99.69		
98-560	3.17	0.9	0.07		96.41		
98-562	3.71	0.35	0		99.03		
98-563	3.46	0.32	0.09		99.42		
98-568	3.35	0.6	0.13		98.33		
98-571	4.91	0.15	0.14		100.1		
98-573	4.84	0.41	0.14		99		

Fortuna Leucocratic Intrusion

Analyte Symbol	K2O	TiO2	P2O5	LOI	Total	Au	Ag
Unit Symbol	%	%	%	%	%	ppb	ppm
98-557	6.01	0.12	0		100.11		
98-558	5.27	0.24	0.04		99.67		
98-569	5.87	0.14	0.01		99.93		

Pajonal diorite

Analyte Symbol	K2O	TiO2	P2O5	LOI	Total	Au	Ag
Unit Symbol	%	%	%	%	%	ppb	ppm
00-194	3.42	0.86	0.24		99.42		
00-196	4.27	0.85	0.18		99.26		
00-197	4.54	0.79	0.17		99.24		
00-199	4.04	0.93	0.21		98.63		
00-200	3.68	0.84	0.25		99.24		

Central Diorite

Analyte Symbol	K2O	TiO2	P2O5	LOI	Total	Au	Ag
Unit Symbol	%	%	%	%	%	ppb	ppm
00-128	4.57	0.84	0.16		98.99		
00-135	4.2	0.85	0.24		98.88		
00-136	2.73	0.74	0.21		98.86		
00-178	2.14	0.92	0.26		99		
00-186	2.23	0.79	0.19		99		

Dark Diorite

Analyte Symbol	K2O	TiO2	P2O5	LOI	Total	Au	Ag
Unit Symbol	%	%	%	%	%	ppb	ppm
00-145	1.9	0.88	0.23		98.99		
00-160	1.41	0.86	0.23		99		
00-181	4.03	0.79	0.16		98.45		
00-184	2.38	0.94	0.26		99		

Equis Monzodiorite

Analyte Symbol	K2O	TiO2	P2O5	LOI	Total	Au	Ag
Unit Symbol	%	%	%	%	%	ppb	ppm
00-141	4.22	0.85	0.18		98.85		
00-146	3.06	0.65	0.29		98.97		
00-151	4.61	0.35	0.04		98.23		
00-155	4.13	0.38	0.12		98.77		
00-173	4.26	0.79	0.14		99		

South Granodiorite

Analyte Symbol	K2O	TiO2	P2O5	LOI	Total	Au	Ag
Unit Symbol	%	%	%	%	%	ppb	ppm
00-126	4.66	0.58	0.19		97.88		
00-130	2.74	0.58	0.21		98.88		
00-156	2.6	0.52	0.19		98.83		
00-159	2.45	0.51	0.19		98.6		
00-162	2.98	0.45	0.19		98.04		

El Abra mine porphyry

Analyte Symbol	K2O	TiO2	P2O5	LOI	Total	Au	Ag
Unit Symbol	%	%	%	%	%	ppb	ppm
00-138	2.66	0.4	0.17		98.64		
00-140	3.11	0.32	0.08		100.83		
00-144	3.11	0.34	0.11		98.07		
00-168	2.91	0.81	0.24		98.67		
00-172	2.53	0.37	0.14		97.8		

El Abra Apolo leucogranite

Analyte Symbol	K2O	TiO2	P2O5	LOI	Total	Au	Ag
Unit Symbol	%	%	%	%	%	ppb	ppm
00-120	4.88	0.45	0.07		99		
00-125	3.51	0.37	0.13		97.9		
00-154	4.86	0.32	0.08		98.8		
00-171	3.68	0.39	0.11		98.74		
00-187	4.98	0.35	0.06		99		

El Abra Aplite

Analyte Symbol	K2O	TiO2	P2O5	LOI	Total	Au	Ag
Unit Symbol	%	%	%	%	%	ppb	ppm
00-124	4.42	0.4	0.04		98.31		
00-150	5.63	0.47	0.03		98.92		
00-158	4.16	0.41	0.07		99.11		
00-169	5.6	0.32	0.04		99		
00-183	5.77	0.52	0.07		98.24		
00-188	4.86	0.35	0.09		99		

Los Picos

Analyte Symbol	As	Ba	Be	Bi	Br	Cd	Co
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-518		686					
98-519		629					
98-520		750					
98-522		621					
98-524		646					
98-525		562					
98-526		541					
98-527		650					
98-540		701					
98-549		624					
98-564							

Fortuna Gris

Analyte Symbol	As	Ba	Be	Bi	Br	Cd	Co
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-532		690					
98-534		714					
98-544		627					
98-548		852					
98-550		644					
98-551		631					
98-552		687					
98-553		697					
98-559		857					
98-561		702					
98-566		698					
98-576		621					

Fortuna Clara

Analyte Symbol	As	Ba	Be	Bi	Br	Cd	Co
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-535		738					
98-536		870					
98-538							
98-541		812					
98-547		832					
98-554		635					
98-556							
98-567		663					
98-570		686					
98-572		719					
98-575		568					
98-577		795					

San Lorenzo

Analyte Symbol	As	Ba	Be	Bi	Br	Cd	Co
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-531		705					
98-542		717					
98-545		472					
98-546		688					
98-560		439					
98-562		787					
98-563		630					
98-568		928					
98-571							
98-573		870					

Fortuna Leucocratic Intrusion

Analyte Symbol	As	Ba	Be	Bi	Br	Cd	Co
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-557		98					
98-558		296					
98-569							

Pajonal diorite

Analyte Symbol	As	Ba	Be	Bi	Br	Cd	Co
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-194		557					
00-196		617					
00-197		585					
00-199		659					
00-200		610					

Central Diorite

Analyte Symbol	As	Ba	Be	Bi	Br	Cd	Co
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-128		618					
00-135		599					
00-136		650					
00-178		487					
00-186		462					

Dark Diorite

Analyte Symbol	As	Ba	Be	Bi	Br	Cd	Co
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-145		503					
00-160		411					
00-181							
00-184		335					

Equis Monzodiorite

Analyte Symbol	As	Ba	Be	Bi	Br	Cd	Co
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-141		574					
00-146		811					
00-151		466					
00-155		625					
00-173		753					

South Granodiorite

Analyte Symbol	As	Ba	Be	Bi	Br	Cd	Co
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-126		405					
00-130		749					
00-156		693					
00-159		654					
00-162		788					

El Abra mine porphyry

Analyte Symbol	As	Ba	Be	Bi	Br	Cd	Co
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-138		758					
00-140		952					
00-144		855					
00-168		689					
00-172		770					

El Abra Apolo leucogranite

Analyte Symbol	As	Ba	Be	Bi	Br	Cd	Co
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-120		482					
00-125		402					
00-154		455					
00-171		506					
00-187		517					

El Abra Aplite

Analyte Symbol	As	Ba	Be	Bi	Br	Cd	Co
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-124		838					
00-150		532					
00-158		670					
00-169		578					
00-183		525					
00-188		701					

Los Picos

Analyte Symbol	Cr	Cs	Cu	Ga	Ge	Hf	Hg
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-518		4.1				4.67	
98-519		3.47				5.18	
98-520		7.87				8.45	
98-522		4.86				4.54	
98-524		4.81				5.79	
98-525		8.17				7.95	
98-526		1.97				3.2	
98-527		6.58				8.96	
98-540		6.64				8.62	
98-549		4.59				5.9	

98-564

Fortuna Gris

Analyte Symbol	Cr	Cs	Cu	Ga	Ge	Hf	Hg
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-532		4.22				4.11	
98-534		7.62				10.11	
98-544		4.51				4.67	
98-548		4.83				4.48	
98-550		8.49				5.68	
98-551		4.45				4.37	
98-552		5.34				2.23	
98-553		6.72				4.24	
98-559		8.27				4.18	
98-561		5.48				4.23	
98-566		5.73				4.52	
98-576		3.98				2.1	

Fortuna Clara

Analyte Symbol	Cr	Cs	Cu	Ga	Ge	Hf	Hg
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-535		4.92				2.9	
98-536		2.71				2.99	
98-538							
98-541		1.57				3.23	
98-547		3.87				4.19	
98-554		3.47				2.93	
98-556							
98-567		3.19				2.07	
98-570		8.73				4.1	
98-572		4.81				8.25	
98-575		4.15				1.91	
98-577		8.22				3.01	

San Lorenzo

Analyte Symbol	Cr	Cs	Cu	Ga	Ge	Hf	Hg
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-531		4.15				3.11	
98-542		1.26				2.8	
98-545		1.71				2.76	
98-546		1.47				3	
98-560		5.75				3.86	
98-562		8.59				5.14	
98-563		6.72				3.19	
98-568		6.3				4.57	
98-571							
98-573		3.74				2.73	

Fortuna Leucocratic Intrusion

Analyte Symbol	Cr	Cs	Cu	Ga	Ge	Hf	Hg
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-557		4.12				2.89	
98-558		3.75				3.43	
98-569							

Pajonal diorite

Analyte Symbol	Cr	Cs	Cu	Ga	Ge	Hf	Hg
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-194		11.82				7.83	
00-196		6.4				8.32	
00-197		8.54				8.43	
00-199		6.67				6.27	
00-200		4.24				4.99	

Central Diorite

Analyte Symbol	Cr	Cs	Cu	Ga	Ge	Hf	Hg
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-128		4.47				4.27	
00-135		5.31				7.3	
00-136		2.49				4.69	
00-178		3.3				3.51	
00-186		4.86				5.09	

Dark Diorite

Analyte Symbol	Cr	Cs	Cu	Ga	Ge	Hf	Hg
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-145		6.28				2.81	
00-160		4				1.96	
00-181							
00-184		14.75				2.19	

Equis Monzodiorite

Analyte Symbol	Cr	Cs	Cu	Ga	Ge	Hf	Hg
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-141		6.27				8.19	
00-146		21.35				4.65	
00-151		2.29				3.6	
00-155		8.82				5.04	
00-173		4.02				6.29	

South Granodiorite

Analyte Symbol	Cr	Cs	Cu	Ga	Ge	Hf	Hg
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-126		8.8				3.37	
00-130		3.33				2.04	
00-156		2.43				2.74	
00-159		2.65				3.12	
00-162		5.22				2.14	

El Abra mine porphyry

Analyte Symbol	Cr	Cs	Cu	Ga	Ge	Hf	Hg
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-138		1.37				2.62	
00-140		1.25				1.98	
00-144		0.97				3.11	
00-168		3.01				4.18	
00-172		1.92				2.77	

El Abra Apolo leucogranite

Analyte Symbol	Cr	Cs	Cu	Ga	Ge	Hf	Hg
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-120		4.17				3.09	
00-125		2.59				4.75	
00-154		7.12				4.59	
00-171		9.31				12.92	
00-187		2.13				3.37	

El Abra Aplite

Analyte Symbol	Cr	Cs	Cu	Ga	Ge	Hf	Hg
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-124		5.03				7.48	
00-150		3.24				6.8	
00-158		9.24				2.53	
00-169		1.92				5.2	
00-183		1.64				7.03	
00-188		2				7.93	

Los Picos

Analyte Symbol	In	Ir	Mo	Nb	Ni	Pb	Rb
Unit Symbol	ppm	ppb	ppm	ppm	ppm	ppm	ppm
98-518				9.81			105
98-519				9.35			121
98-520				11.2			163
98-522				9.2			100
98-524				9.76			113
98-525				13.3			238
98-526				5.92			60
98-527				9.74			117
98-540				12.81			219
98-549				6.59			61
98-564							

Fortuna Gris

Analyte Symbol	In	Ir	Mo	Nb	Ni	Pb	Rb
Unit Symbol	ppm	ppb	ppm	ppm	ppm	ppm	ppm
98-532				8			94
98-534				11.18			178
98-544				7.86			95
98-548				8.81			160
98-550				11.12			202
98-551				10.09			127
98-552				8.47			88
98-553				8.68			128
98-559				7.63			127
98-561				7.27			83
98-566				9.07			152
98-576				7.71			96

Fortuna Clara

Analyte Symbol	In	Ir	Mo	Nb	Ni	Pb	Rb
Unit Symbol	ppm	ppb	ppm	ppm	ppm	ppm	ppm
98-535				6.19			89
98-536				5.67			61
98-538							
98-541				5.89			51
98-547				7.33			75
98-554				8.68			81
98-556							
98-567				4.61			52
98-570				11.6			216
98-572				8.24			108
98-575				6.34			108
98-577				8.2			135

San Lorenzo

Analyte Symbol	In	Ir	Mo	Nb	Ni	Pb	Rb
Unit Symbol	ppm	ppb	ppm	ppm	ppm	ppm	ppm
98-531				7.7			87
98-542				6.63			55
98-545				6.48			125
98-546				6.78			73
98-560				6.86			77
98-562				8.59			162
98-563				6.73			112
98-568				8.65			110
98-571							
98-573				6.19			121

Fortuna Leucocratic Intrusion

Analyte Symbol	In	Ir	Mo	Nb	Ni	Pb	Rb
Unit Symbol	ppm	ppb	ppm	ppm	ppm	ppm	ppm
98-557				9.95			162
98-558				11.31			144
98-569							

Pajonal diorite

Analyte Symbol	In	Ir	Mo	Nb	Ni	Pb	Rb
Unit Symbol	ppm	ppb	ppm	ppm	ppm	ppm	ppm
00-194				13.05			179
00-196				13.01			148
00-197				13.26			180
00-199				12.81			153
00-200				8.29			130

Central Diorite

Analyte Symbol	In	Ir	Mo	Nb	Ni	Pb	Rb
Unit Symbol	ppm	ppb	ppm	ppm	ppm	ppm	ppm
00-128				11.66			105
00-135				10.73			132
00-136				8.19			70
00-178				8.85			78
00-186				9.61			108

Dark Diorite

Analyte Symbol	In	Ir	Mo	Nb	Ni	Pb	Rb
Unit Symbol	ppm	ppb	ppm	ppm	ppm	ppm	ppm
00-145				5.08			72
00-160				4.33			69
00-181							
00-184				5.32			188

Equis Monzodiorite

Analyte Symbol	In	Ir	Mo	Nb	Ni	Pb	Rb
Unit Symbol	ppm	ppb	ppm	ppm	ppm	ppm	ppm
00-141				13.11			140
00-146				10.4			240
00-151				10.96			95
00-155				9.67			163
00-173				11.45			130

South Granodiorite

Analyte Symbol	In	Ir	Mo	Nb	Ni	Pb	Rb
Unit Symbol	ppm	ppb	ppm	ppm	ppm	ppm	ppm
00-126				5.97			286
00-130				8.59			79
00-156				5.73			69
00-159				7.06			67
00-162				8.59			87

El Abra mine porphyry

Analyte Symbol	In	Ir	Mo	Nb	Ni	Pb	Rb
Unit Symbol	ppm	ppb	ppm	ppm	ppm	ppm	ppm
00-138				5.92			46
00-140				5.97			50
00-144				6.26			39
00-168				8.57			96
00-172				6.02			51

El Abra Apolo leucogranite

Analyte Symbol	In	Ir	Mo	Nb	Ni	Pb	Rb
Unit Symbol	ppm	ppb	ppm	ppm	ppm	ppm	ppm
00-120				3.42			104
00-125				11.37			103
00-154				11.08			156
00-171				10.59			156
00-187				8.55			101

El Abra Aplite

Analyte Symbol	In	Ir	Mo	Nb	Ni	Pb	Rb
Unit Symbol	ppm	ppb	ppm	ppm	ppm	ppm	ppm
00-124				12.6			152
00-150				9.9			129
00-158				12.45			236
00-169				11.52			114
00-183				12.4			92
00-188				13.63			161

Los Picos

Analyte Symbol	S	Sb	Sc	Se	Sn	Sr	Ta
Unit Symbol	%	ppm	ppm	ppm	ppm	ppm	ppm
98-518						519	0.71
98-519						692	0.64
98-520						493	0.97
98-522						546	0.7
98-524						461	0.72
98-525						258	1.18
98-526						865	0.41
98-527						416	0.74
98-540						565	1
98-549						456	0.69

98-564

Fortuna Gris

Analyte Symbol	S	Sb	Sc	Se	Sn	Sr	Ta
Unit Symbol	%	ppm	ppm	ppm	ppm	ppm	ppm
98-532						562	0.61
98-534						339	0.96
98-544						755	0.61
98-548						230	0.86
98-550						352	0.94
98-551						350	0.76
98-552						595	0.71
98-553						495	0.65
98-559						570	0.59
98-561						479	0.9
98-566						567	0.62
98-576						557	0.53

Fortuna Clara

Analyte Symbol	S	Sb	Sc	Se	Sn	Sr	Ta
Unit Symbol	%	ppm	ppm	ppm	ppm	ppm	ppm
98-535						974	0.48
98-536						718	0.44
98-538							
98-541						798	0.44
98-547						737	0.56
98-554						603	0.63
98-556							
98-567						495	0.35
98-570						560	1.18
98-572						626	0.65
98-575						608	0.41
98-577						604	0.82

San Lorenzo

Analyte Symbol	S	Sb	Sc	Se	Sn	Sr	Ta
Unit Symbol	%	ppm	ppm	ppm	ppm	ppm	ppm
98-531						606	0.76
98-542						726	0.45
98-545						459	0.58
98-546						584	0.6
98-560						254	0.46
98-562						419	0.8
98-563						377	0.68
98-568						357	0.72
98-571							
98-573						403	0.71

Fortuna Leucocratic Intrusion

Analyte Symbol	S	Sb	Sc	Se	Sn	Sr	Ta
Unit Symbol	%	ppm	ppm	ppm	ppm	ppm	ppm
98-557						62	1.35
98-558						115	1.36
98-569							

Pajonal diorite

Analyte Symbol	S	Sb	Sc	Se	Sn	Sr	Ta
Unit Symbol	%	ppm	ppm	ppm	ppm	ppm	ppm
00-194						411	1.06
00-196						315	1.02
00-197						282	1.11
00-199						364	0.96
00-200						388	0.69

Central Diorite

Analyte Symbol	S	Sb	Sc	Se	Sn	Sr	Ta
Unit Symbol	%	ppm	ppm	ppm	ppm	ppm	ppm
00-128						276	0.94
00-135						266	0.84
00-136						380	0.58
00-178						461	0.53
00-186						489	0.76

Dark Diorite

Analyte Symbol	S	Sb	Sc	Se	Sn	Sr	Ta
Unit Symbol	%	ppm	ppm	ppm	ppm	ppm	ppm
00-145						566	0.32
00-160						650	0.27
00-181							
00-184						505	0.29

Equis Monzodiorite

Analyte Symbol	S	Sb	Sc	Se	Sn	Sr	Ta
Unit Symbol	%	ppm	ppm	ppm	ppm	ppm	ppm
00-141						2.57	1.06
00-146						406	0.85
00-151						276	1.27
00-155						317	1.01
00-173						245	0.86

South Granodiorite

Analyte Symbol	S	Sb	Sc	Se	Sn	Sr	Ta
Unit Symbol	%	ppm	ppm	ppm	ppm	ppm	ppm
00-126						88	0.48
00-130						605	0.68
00-156						553	0.47
00-159						610	0.5
00-162						698	0.69

El Abra mine porphyry

Analyte Symbol	S	Sb	Sc	Se	Sn	Sr	Ta
Unit Symbol	%	ppm	ppm	ppm	ppm	ppm	ppm
00-138						632	0.47
00-140						680	0.56
00-144						618	0.54
00-168						534	0.69
00-172						398	0.51

El Abra Apolo leucogranite

Analyte Symbol	S	Sb	Sc	Se	Sn	Sr	Ta
Unit Symbol	%	ppm	ppm	ppm	ppm	ppm	ppm
00-120						12	0.3
00-125						293	1.41
00-154						257	1.31
00-171						347	1.32
00-187						221	0.91

El Abra Aplite

Analyte Symbol	S	Sb	Sc	Se	Sn	Sr	Ta
Unit Symbol	%	ppm	ppm	ppm	ppm	ppm	ppm
00-124						308	1.26
00-150						186	0.79
00-158						366	1.25
00-169						165	1.34
00-183						148	1.18
00-188						265	1.53

Los Picos

Analyte Symbol	Th	U	V	W	Y	Zn	Zr
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-518	14.31	3.15			19.97		167
98-519	13.98	4.11			20.6		208
98-520	30.11	8.11			25.5		285
98-522	14.11	3.52			18.49		157
98-524	16.73	4.76			20.04		210
98-525	57.02	14.17			24.84		279
98-526	5.58	1.67			14.53		126
98-527	21.61	4.92			20.61		360
98-540	28.35	5.61			25.22		359
98-549	17.03	4.54			13.89		162

98-564

Fortuna Gris

Analyte Symbol	Th	U	V	W	Y	Zn	Zr
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-532	13.41	3.93			13.82		146
98-534	20.36	6.36			22.53		350
98-544	15.12	2.77			17.39		168
98-548	24.51	4.15			17.18		146
98-550	25.29	5.46			23.51		250
98-551	14.58	4.56			21.93		171
98-552	8.75	2.02			14.56		74
98-553	11.88	2.86			14.88		149
98-559	10.92	3.3			13.18		178
98-561	21.31	4.96			11.72		112
98-566	15.26	3.63			17.61		305
98-576	10.39	2.7			13.09		88

Fortuna Clara

Analyte Symbol	Th	U	V	W	Y	Zn	Zr
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-535	6.7	1.91			8.32		105
98-536	6.28	2.58			8.35		108
98-538							
98-541	5.64	1.64			7.26		121
98-547	7.48	2.64			8.64		157
98-554	11.12	3.19			13.45		99
98-556							
98-567	3.25	0.91			6.82		77
98-570	19.97	5.81			13.28		64
98-572	16.05	5.87			11.74		116
98-575	8.3	2.38			8.74		80
98-577	9.6	3.28			10.65		63

San Lorenzo

Analyte Symbol	Th	U	V	W	Y	Zn	Zr
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-531	13.32	4.01			10.49		112
98-542	5.52	1.79			7.54		117
98-545	15.68	7.02			5.8		77
98-546	12.95	4.3			7.8		93
98-560	11.76	5.02			7.17		140
98-562	18.45	4.39			12.79		188
98-563	21.89	4.41			9.28		85
98-568	19.14	3.11			6		163
98-571							
98-573	19.85	4.15			6.42		86

Fortuna Leucocratic Intrusion

Analyte Symbol	Th	U	V	W	Y	Zn	Zr
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-557	50.82	12.11			13.39		59
98-558	32.34	6.84			13.91		86
98-569							

Pajonal diorite

Analyte Symbol	Th	U	V	W	Y	Zn	Zr
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-194	38.26	11.51			29.56		291
00-196	28.09	8.51			23.7		330
00-197	32.97	7.69			21.5		317
00-199	24.16	6.35			22.88		242
00-200	26.42	7.12			18.94		183

Central Diorite

Analyte Symbol	Th	U	V	W	Y	Zn	Zr
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-128	25.25	5.58			18.8		161
00-135	26.67	7.26			23.21		282
00-136	11.36	3.03			17.73		184
00-178	8.97	2.58			16.49		140
00-186	18.59	5.61			15.25		193

Dark Diorite

Analyte Symbol	Th	U	V	W	Y	Zn	Zr
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-145	4.62	1.55			9.8		111
00-160	3.3	1.11			15.7		71
00-181							
00-184	5.26	2.6			15.73		82

Equis Monzodiorite

Analyte Symbol	Th	U	V	W	Y	Zn	Zr
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-141	31.24	8.76			23.3		309
00-146	17.17	5.4			15.9		162
00-151	37.56	11.25			15.8		103
00-155	29.54	10.11			12.1		166
00-173	32.82	7.21			29.7		246

South Granodiorite

Analyte Symbol	Th	U	V	W	Y	Zn	Zr
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-126	7.49	1.31			9.83		125
00-130	7.68	2.82			12.93		70
00-156	8.73	2.99			7.65		99
00-159	7.61	2.59			9.27		114
00-162	7.73	3.06			8.61		72

El Abra mine porphyry

Analyte Symbol	Th	U	V	W	Y	Zn	Zr
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-138	5.29	1.24			6.61		95
00-140	6.59	2.34			7.9		71
00-144	6.46	1.87			7.95		113
00-168	12.5	4.07			15.35		152
00-172	5.24	1.64			6.2		106

El Abra Apolo leucogranite

Analyte Symbol	Th	U	V	W	Y	Zn	Zr
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-120	3.57	1.07			7.42		114
00-125	28.91	11.7			13.74		139
00-154	33.52	10.32			14.22		128
00-171	25.72	11.06			16.44		472
00-187	39.06	8.9			11.21		104

El Abra Aplite

Analyte Symbol	Th	U	V	W	Y	Zn	Zr
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-124	31.55	20.89			3.07		251
00-150	59.29	25.44			4.07		228
00-158	39.5	11.57			3.23		80
00-169	65.39	20.11			3.6		155
00-183	65.11	7.39			5.01		248
00-188	72.09	19.77			3.52		249

Los Picos

Analyte Symbol	La	Ce	Pr	Nd	Sm	Eu	Gd
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-518	25.5	53.5	6.29	24.8	4.8	1.04	4.08
98-519	23.8	50.6	6.02	24.1	4.78	1.09	4.23
98-520	32.6	68.3	7.89	30.2	5.92	1.2	5.05
98-522	23.8	51.2	6.04	23.8	4.65	1.05	3.85
98-524	24.3	51.3	6.04	23.8	4.76	1.06	4.2
98-525	36.1	78	8.75	32.2	6.04	0.76	4.87
98-526	17.8	37.5	4.49	18.5	3.71	1.05	3.16
98-527	27.7	59.5	6.96	27.5	5.25	1.04	4.42
98-540	29.5	63.8	7.43	28.3	5.37	0.88	4.49
98-549	23.3	48.8	5.65	22.4	4.51	0.98	3.81

98-564

Fortuna Gris

Analyte Symbol	La	Ce	Pr	Nd	Sm	Eu	Gd
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-532	23.1	47.2	5.43	21	3.82	0.92	3.15
98-534	30.6	64.7	7.5	28.3	5.65	0.93	4.44
98-544	22.5	47.4	5.53	22.1	4.39	0.95	3.64
98-548	24.3	49.3	5.54	20	3.58	0.47	3.08
98-550	30.1	63.5	7.34	28.1	5.36	0.96	4.26
98-551	24.5	50.9	5.81	22.7	4.21	0.95	3.29
98-552	23.4	48.8	5.76	22.9	4.38	0.96	3.38
98-553	24.5	49.5	5.66	21.7	3.93	0.91	3
98-559	29.1	55.9	6.14	22.8	3.87	0.88	3.09
98-561	31.7	60.8	6.67	24.6	4.23	0.85	3.32
98-566	25	51.1	5.87	23	4.19	0.91	3.42
98-576	22.1	44.4	5.09	19.6	3.49	0.86	2.88

Fortuna Clara

Analyte Symbol	La	Ce	Pr	Nd	Sm	Eu	Gd
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-535	21	41.5	4.71	18.1	3.1	0.83	2.13
98-536	20.1	39.4	4.41	16.8	2.98	0.79	2.15
98-538							
98-541	20.4	38.2	4.38	16.6	2.77	0.79	1.9
98-547	22.7	43.2	4.84	18	3.13	0.84	2.21
98-554	25.6	54.5	6.37	23.8	4.28	0.95	3.21
98-556							
98-567	16.6	31.6	3.6	13.3	2.48	0.62	1.67
98-570	22.8	45.5	5.36	20.1	3.68	0.8	2.88
98-572	23.9	50.2	5.77	22	3.75	1.01	2.89
98-575	16.1	33.5	3.98	15.5	2.79	0.74	2.03
98-577	17	34.3	4.08	16	3.27	0.73	2.55

San Lorenzo

Analyte Symbol	La	Ce	Pr	Nd	Sm	Eu	Gd
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-531	21.6	43.8	4.94	18.3	3.19	0.81	2.35
98-542	18.9	36	4.09	15.6	2.67	0.69	1.84
98-545	17.7	29.5	3	10.4	1.78	0.45	1.3
98-546	24.6	44.2	4.7	17	2.73	0.73	1.94
98-560	30.9	60.4	6.79	25.4	4.06	0.83	2.24
98-562	22.8	45.8	5.11	18.3	3.52	0.54	2.64
98-563	19.8	38	4.16	14	2.74	0.56	2.14
98-568	28.5	48.8	4.77	16	2.32	0.47	1.46
98-571							
98-573	41.7	60.9	5.63	18.1	2.42	0.56	1.64

Fortuna Leucocratic Intrusion

Analyte Symbol	La	Ce	Pr	Nd	Sm	Eu	Gd
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-557	33.5	67	7.19	24.4	3.95	0.25	2.74
98-558	29.3	57.5	6.06	20.2	3.29	0.34	2.45
98-569							

Pajonal diorite

Analyte Symbol	La	Ce	Pr	Nd	Sm	Eu	Gd
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-194	39.1	86.1	9.95	38.4	7.42	1.07	6.13
00-196	31.6	69	7.83	30.4	5.68	0.94	4.84
00-197	28.8	61.7	7.25	27.8	5.43	0.86	4.35
00-199	29.7	64.6	7.6	29.3	5.74	1.03	4.78
00-200	28.8	60.4	6.94	26.7	4.97	1	4.02

Central Diorite

Analyte Symbol	La	Ce	Pr	Nd	Sm	Eu	Gd
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-128	27.8	60.8	6.84	25.8	4.85	0.78	4.03
00-135	28.9	63.3	7.34	28.2	5.32	0.92	4.63
00-136	23.5	50.4	5.87	23.2	4.5	0.94	3.71
00-178	22.7	47.2	5.57	22.7	4.37	1.15	3.59
00-186	38.6	72.9	7.52	26.4	4.52	1	3.48

Dark Diorite

Analyte Symbol	La	Ce	Pr	Nd	Sm	Eu	Gd
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-145	15.4	30.6	3.52	14.6	2.81	0.88	2.14
00-160	16.7	33.7	4.08	17.4	3.61	0.95	3.31
00-181							
00-184	26.5	50	5.59	21.6	4.31	1.04	3.51

Equis Monzodiorite

Analyte Symbol	La	Ce	Pr	Nd	Sm	Eu	Gd
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-141	32.4	70.4	7.98	30.5	5.71	0.91	4.72
00-146	27.7	56.1	6.18	23.6	4.67	1.09	3.72
00-151	22.9	53.2	5.88	21.7	3.91	0.65	2.94
00-155	28.6	64	7.09	25.7	4.17	0.71	2.66
00-173	26.9	58.3	6.58	25.1	5.25	1.05	5.09

South Granodiorite

Analyte Symbol	La	Ce	Pr	Nd	Sm	Eu	Gd
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-126	26.2	50.1	5.29	19.5	3.2	0.89	2.36
00-130	23.9	48.6	5.31	20	3.39	0.86	2.77
00-156	17.4	32.8	3.57	13.4	2.34	0.68	1.73
00-159	20.3	41.1	4.58	17.2	2.92	0.83	2.18
00-162	22.3	44	4.72	17.1	2.87	0.87	2.06

El Abra mine porphyry

Analyte Symbol	La	Ce	Pr	Nd	Sm	Eu	Gd
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-138	21.1	40.9	4.36	16.1	2.56	0.72	1.71
00-140	20.7	40.5	4.3	15.3	2.7	0.87	2.11
00-144	21.6	41.2	4.4	15.8	2.43	0.72	1.74
00-168	23.4	49.5	5.69	22	4.1	0.97	3.45
00-172	16.8	33.2	3.55	13	2.02	0.54	1.45

El Abra Apolo leucogranite

Analyte Symbol	La	Ce	Pr	Nd	Sm	Eu	Gd
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-120	12.3	24.3	2.57	9.7	1.78	0.4	1.45
00-125	20.5	46.1	5.12	19.1	3.44	0.63	2.63
00-154	28.2	57.7	6.12	21.3	3.71	0.59	2.66
00-171	23	52.4	6.02	21.8	3.81	0.72	2.88
00-187	20.2	42.5	4.59	17	3.07	0.61	2.6

El Abra Aplite

Analyte Symbol	La	Ce	Pr	Nd	Sm	Eu	Gd
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-124	29.2	65.6	7.35	27	4.76	0.85	3.77
00-150	30.1	66.2	7.28	26.9	5.44	0.74	4.6
00-158	28.3	63.7	7.58	28.8	5.29	0.89	3.84
00-169	25.1	56.7	6.38	23.2	4.41	0.49	3.76
00-183	35.2	76.7	8.57	31.9	6.13	0.73	4.91
00-188	23.5	53.4	5.91	22.2	4.08	0.6	3.59

Los Picos

Analyte Symbol	Tb	Dy	Ho	Er	Tl	Tm	Yb
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-518	0.57	3.65	0.71	2.04			2.02
98-519	0.61	3.95	0.75	2.14			2.17
98-520	0.74	4.71	0.93	2.61			2.67
98-522	0.54	3.47	0.68	1.89			1.95
98-524	0.59	3.7	0.73	2			2.14
98-525	0.7	4.41	0.87	2.53			2.72
98-526	0.44	2.65	0.5	1.37			1.36
98-527	0.63	3.94	0.75	2.22			2.32
98-540	0.66	4.19	0.81	2.36			2.44
98-549	0.54	3.4	0.67	1.89			1.95

98-564

Fortuna Gris

Analyte Symbol	Tb	Dy	Ho	Er	Tl	Tm	Yb
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-532	0.41	2.6	0.48	1.33			1.39
98-534	0.66	4.17	0.8	2.29			2.43
98-544	0.52	3.24	0.64	1.78			1.79
98-548	0.45	2.96	0.61	1.8			1.97
98-550	0.61	3.91	0.74	2.16			2.29
98-551	0.46	2.73	0.51	1.43			1.55
98-552	0.46	2.89	0.53	1.4			1.51
98-553	0.42	2.53	0.48	1.35			1.37
98-559	0.4	2.45	0.47	1.28			1.37
98-561	0.47	2.81	0.52	1.48			1.66
98-566	0.49	2.95	0.57	1.5			1.55
98-576	0.41	2.37	0.43	1.26			1.21

Fortuna Clara

Analyte Symbol	Tb	Dy	Ho	Er	Tl	Tm	Yb
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-535	0.27	1.6	0.3	0.84			0.81
98-536	0.27	1.57	0.28	0.78			0.82
98-538							
98-541	0.24	1.4	0.27	0.71			0.77
98-547	0.28	1.58	0.31	0.82			0.93
98-554	0.43	2.55	0.49	1.36			1.44
98-556							
98-567	0.23	1.31	0.24	0.62			0.63
98-570	0.4	2.42	0.47	1.32			1.51
98-572	0.39	2.35	0.45	1.26			1.43
98-575	0.27	1.63	0.29	0.82			0.9
98-577	0.35	2.19	0.39	1.2			1.29

San Lorenzo

Analyte Symbol	Tb	Dy	Ho	Er	Tl	Tm	Yb
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-531	0.31	1.92	0.36	1.02			1.17
98-542	0.23	1.37	0.25	0.68			0.72
98-545	0.17	1.06	0.2	0.57			0.74
98-546	0.26	1.52	0.28	0.8			0.88
98-560	0.28	1.48	0.25	0.74			0.9
98-562	0.39	2.24	0.43	1.26			1.58
98-563	0.27	1.75	0.31	0.98			1.2
98-568	0.19	1.16	0.22	0.67			0.89
98-571							
98-573	0.19	1.12	0.21	0.59			0.6

Fortuna Leucocratic Intrusion

Analyte Symbol	Tb	Dy	Ho	Er	Tl	Tm	Yb
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
98-557	0.38	2.24	0.44	1.3			1.66
98-558	0.35	2.23	0.45	1.41			1.86
98-569							

Pajonal diorite

Analyte Symbol	Tb	Dy	Ho	Er	Tl	Tm	Yb
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-194		5.43	1.06	2.99			3.18
00-196		4.33	0.84	2.46			2.56
00-197		4.07	0.79	2.19			2.34
00-199		4.25	0.84	2.36			2.38
00-200		3.59	0.7	1.93			1.96

Central Diorite

Analyte Symbol	Tb	Dy	Ho	Er	Tl	Tm	Yb
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-128		3.68	0.67	1.94			2.02
00-135		4.25	0.81	2.26			2.26
00-136		3.32	0.62	1.8			1.84
00-178		3.14	0.61	1.74			1.74
00-186		2.95	0.57	1.7			1.73

Dark Diorite

Analyte Symbol	Tb	Dy	Ho	Er	Tl	Tm	Yb
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-145		1.85	0.34	1.03			1.01
00-160		2.95	0.59	1.7			1.71
00-181							
00-184		3.15	0.6	1.65			1.76

Equis Monzodiorite

Analyte Symbol	Tb	Dy	Ho	Er	Tl	Tm	Yb
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-141		4.35	0.84	2.39			2.45
00-146		2.98	0.57	1.61			1.61
00-151		2.68	0.54	1.65			1.86
00-155		2.13	0.41	1.28			1.58
00-173		5	0.99	2.8			2.75

South Granodiorite

Analyte Symbol	Tb	Dy	Ho	Er	Tl	Tm	Yb
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-126		1.9	0.34	1.02			1.16
00-130		2.51	0.46	1.27			1.36
00-156		1.4	0.27	0.73			0.84
00-159		1.72	0.32	0.95			0.98
00-162		1.61	0.29	0.87			0.97

El Abra mine porphyry

Analyte Symbol	Tb	Dy	Ho	Er	Tl	Tm	Yb
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-138		1.26	0.22	0.63			0.67
00-140		1.77	0.33	0.9			1.05
00-144		1.36	0.25	0.73			0.75
00-168		2.91	0.56	1.51			1.61
00-172		1.14	0.21	0.6			0.66

El Abra Apolo leucogranite

Analyte Symbol	Tb	Dy	Ho	Er	Tl	Tm	Yb
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-120		1.28	0.26	0.76			0.9
00-125		2.34	0.47	1.38			1.72
00-154		2.35	0.47	1.38			1.79
00-171		2.74	0.55	1.69			2.11
00-187		2.2	0.42	1.15			1.24

El Abra Aplite

Analyte Symbol	Tb	Dy	Ho	Er	Tl	Tm	Yb
Unit Symbol	ppm	ppm	ppm	ppm	ppm	ppm	ppm
00-124		16.22	0.59	1.62			1.84
00-150		19.5	0.76	2.03			2.08
00-158		17.92	0.62	1.8			2.08
00-169		20.79	0.71	2.06			2.36
00-183		27.46	0.98	2.85			2.89
00-188		19.93	0.69	2.06			2.47

Los Picos

Analyte Symbol	Lu	Mass	Ga	Pb	Sn	Nb	Rb
Unit Symbol	ppm	g	ppm	ppm	ppm	ppm	ppm
98-518	0.29			11			
98-519	0.31			8			
98-520	0.4			19			
98-522	0.29			25			
98-524	0.32			39			
98-525	0.39			18			
98-526	0.19			13			
98-527	0.34			46			
98-540	0.37			13			
98-549	0.3			21			
98-564							

Fortuna Gris

Analyte Symbol	Lu	Mass	Ga	Pb	Sn	Nb	Rb
Unit Symbol	ppm	g	ppm	ppm	ppm	ppm	ppm
98-532	0.21			10			
98-534	0.36			10			
98-544	0.27			59			
98-548	0.3			36			
98-550	0.32			17			
98-551	0.22			76			
98-552	0.22			9			
98-553	0.21			41			
98-559	0.22			10			
98-561	0.25			17			
98-566	0.22			14			
98-576	0.18			8			

Fortuna Clara

Analyte Symbol	Lu	Mass	Ga	Pb	Sn	Nb	Rb
Unit Symbol	ppm	g	ppm	ppm	ppm	ppm	ppm
98-535	0.12			8			
98-536	0.12			17			
98-538							
98-541	0.11			12			
98-547	0.14			17			
98-554	0.2			7			
98-556							
98-567	0.1			14			
98-570	0.23			7			
98-572	0.22			10			
98-575	0.14			7			
98-577	0.2			9			

San Lorenzo

Analyte Symbol	Lu	Mass	Ga	Pb	Sn	Nb	Rb
Unit Symbol	ppm	g	ppm	ppm	ppm	ppm	ppm
98-531	0.17			11			
98-542	0.11			11			
98-545	0.13			9			
98-546	0.13			11			
98-560	0.14			15			
98-562	0.23			5			
98-563	0.16			13			
98-568	0.13			62			
98-571							
98-573	0.09			9			

Fortuna Leucocratic Intrusion

Analyte Symbol	Lu	Mass	Ga	Pb	Sn	Nb	Rb
Unit Symbol	ppm	g	ppm	ppm	ppm	ppm	ppm
98-557	0.27			15			
98-558	0.29			14			
98-569							

Pajonal diorite

Analyte Symbol	Lu	Mass	Ga	Pb	Sn	Nb	Rb
Unit Symbol	ppm	g	ppm	ppm	ppm	ppm	ppm
00-194	0.46			14			
00-196	0.36			15			
00-197	0.33			15			
00-199	0.36			16			
00-200	0.27			6			

Central Diorite

Analyte Symbol	Lu	Mass	Ga	Pb	Sn	Nb	Rb
Unit Symbol	ppm	g	ppm	ppm	ppm	ppm	ppm
00-128	0.28			12			
00-135	0.33			10			
00-136	0.26			6			
00-178	0.26			8			
00-186	0.33			9			

Dark Diorite

Analyte Symbol	Lu	Mass	Ga	Pb	Sn	Nb	Rb
Unit Symbol	ppm	g	ppm	ppm	ppm	ppm	ppm
00-145	0.16			4			
00-160	0.24			4			
00-181							
00-184	0.24			8			

Equis Monzodiorite

Analyte Symbol	Lu	Mass	Ga	Pb	Sn	Nb	Rb
Unit Symbol	ppm	g	ppm	ppm	ppm	ppm	ppm
00-141	0.36			9			
00-146	0.25			11			
00-151	0.28			3			
00-155	0.24			7			
00-173	0.38			13			

South Granodiorite

Analyte Symbol	Lu	Mass	Ga	Pb	Sn	Nb	Rb
Unit Symbol	ppm	g	ppm	ppm	ppm	ppm	ppm
00-126	0.18			54			
00-130	0.19			7			
00-156	0.13			11			
00-159	0.15			4			
00-162	0.15			8			

El Abra mine porphyry

Analyte Symbol	Lu	Mass	Ga	Pb	Sn	Nb	Rb
Unit Symbol	ppm	g	ppm	ppm	ppm	ppm	ppm
00-138	0.09			9			
00-140	0.15			7			
00-144	0.12			9			
00-168	0.23			6			
00-172	0.1			54			

El Abra Apolo leucogranite

Analyte Symbol	Lu	Mass	Ga	Pb	Sn	Nb	Rb
Unit Symbol	ppm	g	ppm	ppm	ppm	ppm	ppm
00-120	0.14			1			
00-125	0.26			19			
00-154	0.26			8			
00-171	0.31			10			
00-187	0.18			6			

El Abra Aplite

Analyte Symbol	Lu	Mass	Ga	Pb	Sn	Nb	Rb
Unit Symbol	ppm	g	ppm	ppm	ppm	ppm	ppm
00-124	0.27			10			
00-150	0.31			10			
00-158	0.32			14			
00-169	0.36			12			
00-183	0.44			14			
00-188	0.37			15			

Appendix 6 Data from Arnott (2003) Thesis

The following section contains a data table from Arnott (2003). This whole rock geochemical data was used to compare MMH with components of the Chuquicamata Intrusive Complex.

	SiO2 %	Al2O3 %	Fe2O3(T) %	MnO %	MgO %	CaO %	Na2O %	K2O %	TiO2 %
<u>Ester</u>									
Cu 769	70.2	16.88	2.43	0.04	0.07	3.03	4.1	1.72	0.32
<u>Banco</u>									
Cu-1333	68.06	17.36	1.36	0.01	0.37	1.01	3.71	4.06	0.3
Cu-1333	68.02	17.88	1.13	0.01	0.47	0.94	3.58	4.15	0.29
Cu-93-514	69.2	16.9	2.26	0.04	0.63	1.81	5.29	3.17	0.29
<u>Oeste</u>									
Cu93-516	71	15.5	0.74	0.02	0.12	0.31	5.17	4.03	0.2
<u>EsteK</u>									
Cu-1112	70.64	15.74	0.76	0.01	0.46	0.32	5.77	4.09	0.32
Cu-1115	69.94	14.95	2.02	0.04	0.76	0.21	4.1	4.79	0.24
<u>EsteSer</u>									
Cu-020	68.26	11.1	2.63		0.29	3.51		3.43	0.26
Cu-070	83.6	6.42	1.17	0.01	0.49	1.92		1.95	0.1
Cu-093	86.52	4.84	1.03		0.08	1.56		1.46	0.07
Cu-102	80.87	11.74	0.89		0.22	0.05		3.58	0.19
Cu-1144	65.56	15.53	0.53	0.23	0.28	1.69	0	10.43	0.18
<u>EsteProp</u>									
Cu-746	69.01	16.06	1.52	0.09	0.82	1.94	4.99	2.83	0.31
<u>Ksil</u>									
Cu-1334	68.96	15.19	0.5	0.01	0.49	1.42	3.92	5.79	0.23
<u>Fiesta</u>									
Cu-524	64.4	14.9	3.18	0.07	1.52	3.69	3.55	3.87	0.56
Cu-526	64.6	16.6	3.61	0.08	1.31	4.13	2.48	4.41	0.43

	P2O5 %	LOI %	Total %	Au ppb	Ag ppm	As ppm	Ba ppm	Be ppm	Bi ppm
<u>Ester</u>									
Cu 769	0.11	0.44	99.97		0		885		
<u>Banco</u>									
Cu-1333	0.08	2.41	98.73		2		799		
Cu-1333	0.05	2.82	99.48						
Cu-93-514	0.04	0.8	100.4						
<u>Oeste</u>									
Cu93-516	0.06	1.7	98.8						
<u>EsteK</u>									
Cu-1112	0.06	0.42	98.59				415		
Cu-1115	0.08	1.19	98.32				625		
<u>EsteSer</u>									
Cu-020	0.08	8.97	98.56	<1			129.5		
Cu-070	0.03	3.77	99.56		1		80		
Cu-093	0.07	3.3	99.03		11		203		
Cu-102	0.09	1.99	99.67		6		234		
Cu-1144	0.13	4.08	98.64						
<u>EsteProp</u>									
Cu-746	0.15	1.09	98.81						
<u>Ksil</u>									
Cu-1334	0.09	2.94	99.64						
<u>Fiesta</u>									
Cu-524	0.16	0.85	96.75				649		
Cu-526	0.21	1.97	99.83				704		

	Hg ppm	In ppm	Ir ppb	Mo ppm	Nb ppm	Ni ppm	Pb ppm	Rb ppm	S %
<u>Ester</u>									
Cu 769					7	0		65	
<u>Banco</u>									
Cu-1333					5	20		107	
Cu-1333									
Cu-93-514									
<u>Oeste</u>									
Cu93-516									
<u>EsteK</u>									
Cu-1112					18			98	
Cu-1115					12			154	
<u>EsteSer</u>									
Cu-020					7	35		90.4	
Cu-070					6	110		44.8	
Cu-093					1	60		32.6	
Cu-102					6	30		91	
Cu-1144									
<u>EsteProp</u>									
Cu-746									
<u>Ksil</u>									
Cu-1334									
<u>Fiesta</u>									
Cu-524						4		177	
Cu-526						5		75	

	Sb ppm	Sc ppm	Se ppm	Sn ppm	Sr ppm	Ta ppm	Th ppm	U ppm	V ppm
<u>Ester</u>									
Cu 769				0	709	1	2	2	35
<u>Banco</u>									
Cu-1333				7	400	0	3	3	25
Cu-1333									
Cu-93-514									
<u>Oeste</u>									
Cu93-516									
<u>EsteK</u>									
Cu-1112					158		6	2	
Cu-1115					116				
<u>EsteSer</u>									
Cu-020				14	108 <0.5		4	1.5	60
Cu-070				7	94.4 <0.5		1 <0.5		60
Cu-093				6	212 <0.5		7	0.5	30
Cu-102				13	72.9 <0.5		2	0.5	70
Cu-1144									
<u>EsteProp</u>									
Cu-746									
<u>Ksil</u>									
Cu-1334									
<u>Fiesta</u>									
Cu-524					367				
Cu-526					638				

	W ppm	Y ppm	Zn ppm	Zr ppm	La ppm	Ce ppm	Pr ppm	Nd ppm	Sm ppm
<u>Ester</u>									
Cu 769	0	7	30	125	24	43.5	5.1	19	2.6
<u>Banco</u>									
Cu-1333	17	9	435	188	12	24	2.6	9	1.5
Cu-1333									
Cu-93-514									
<u>Oeste</u>									
Cu93-516									
<u>EsteK</u>									
Cu-1112		12		108	15.5	30	3.5	12	2.1
Cu-1115		14		102					
<u>EsteSer</u>									
Cu-020	28	4	145	207	15.5	30.5	3.2	11	1.6
Cu-070	12	2.5	15	48	4.5	8.5	1	3.5	0.6
Cu-093	9	2.5	20	46	21.5	35	3	9	1.2
Cu-102	21	2	15	138.5	7.5	15.5	1.6	5.5	0.9
Cu-1144									
<u>EsteProp</u>									
Cu-746									
<u>Ksil</u>									
Cu-1334									
<u>Fiesta</u>									
Cu-524		27		315	28	64	6.9	26	4.5
Cu-526		9		156	22	47	4.8	16	3.2

	Eu ppm	Gd ppm	Tb ppm	Dy ppm	Ho ppm	Er ppm	Tl ppm	Tm ppm	Yb ppm
<u>Ester</u>									
Cu 769	0.8	2.4	0.3	1.2	0.2	0.8		0.1	0.8
<u>Banco</u>									
Cu-1333	0.4	1.7	0.2	1.3	0.3	0.9		0.1	0.9
Cu-1333									
Cu-93-514									
<u>Oeste</u>									
Cu93-516									
<u>EsteK</u>									
Cu-1112	0.4	1.4	0.4	0.7	<0.1	0.3	<0.1		0.3
Cu-1115									
<u>EsteSer</u>									
Cu-020	0.4	1.1	0.1	0.7	0.1	0.4	<0.1		0.4
Cu-070	0.1	0.5	<0.1	0.3	<0.1	0.2	<0.1		0.2
Cu-093	0.3	0.9	0.1	0.5	0.1	0.3	<0.1		0.2
Cu-102	0.1	0.6	<0.1	0.3	<0.1	0.2	<0.1		0.1
Cu-1144									
<u>EsteProp</u>									
Cu-746									
<u>Ksil</u>									
Cu-1334									
<u>Fiesta</u>									
Cu-524	1	4.3	0.6	3.4	0.8	1.7		0.3	1.9
Cu-526	1.1	1.9	0.3	1.4	0.4	0.7		0.1	0.7

	Lu ppm	Mass g	Ga ppm	Pb ppm	Sn ppm	Nb ppm	Rb ppm
<u>Ester</u>							
Cu 769	0.1			5			
<u>Banco</u>							
Cu-1333	0.1			15			
Cu-1333							
Cu-93-514							
<u>Oeste</u>							
Cu93-516							
<u>EsteK</u>							
Cu-1112	<0.1						
Cu-1115							
<u>EsteSer</u>							
Cu-020	<0.1			20			
Cu-070	<0.1			15			44.8
Cu-093	<0.1			35			
Cu-102	<0.1			20			
Cu-1144							
<u>EsteProp</u>							
Cu-746							
<u>Ksil</u>							
Cu-1334							
<u>Fiesta</u>							
Cu-524	0.3						
Cu-526	0.1						