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GEOLOGICAL MAPPING,

LITHOGEOCHEMICAL SAMPLING,

TRANSIENT ELECTROMAGNETICS,

<u>AND</u>

DIAMOND DRILLING REPORT

CHU CHUA PROJECT

Chu	Chua	Group	Α
Chu	Chua	Group	В
Chu	Chua	Group	С
Chu	Chua	Group	D

Kamloops Mining Division NTS 92P/8E 51°22'N 120°04'W Lat Long

SUB-RECORDER RECEIVED				
JAN	5	1990		
M.R. # VANCOL	JVE	, B.C.		

Owner/Operator Minnova Inc. 4th Floor, 311 Water Street Vancouver, B.C.

Chris Wild December, 1989

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TABLE OF CONTENTS

1.	Introduction 1.1 General 1.2 Location a 1.3 Topography 1.4 Property a 1.5 History 1.6 Work Done	and Access y and Climate and Ownership	1 1 1 3 3 5
2.	Geology 2.1 Regional (2.2 Chu Chua I 2.3 Lithogeoc 2.3.1 An 2.3.2 Dis	Geology North Geology hemistry alytical Methods scussion	6 6 9 11 11 11
3.	Diamond Drilling 3.1 Program S 3.2 Geology o 3.2. 3.3. 3.3. 3.3. 3.3. 3.4. 3.4. 3.4. 3.4.	ummary f Chu Chua Deposit 1 General 2 Hanging Wall 3 Massive Sulphides 4 Talc-Magnetite 5 Silicic Zone 6 Footwall Basalt 7 Basalt Dykes 1 Procedure 2 Discussion hemistry 1 Analytical Methods 2 Discussion	15 15 15 15 15 18 19 20 21 21 21 21 21 22 22 22 23
4.	Transient EM 4.1 Survey Lo 4.2 Results a	gistics nd Discussion	25 25 25
5.	Conclusions		27
6.	References		28

page

List of Appendices

	page
Itemized Cost Statements	29
Statement of qualifications	30
Chu Chua North Lithogeochemical Results	31
Drill Logs (& Results)	32
Chu Chua Drilling Assay and Lithogeochemical	
Results	33
Transient Electromagnetic Survey, Logistics	
Report and Profiles	34
	<pre>Itemized Cost Statements Statement of qualifications Chu Chua North Lithogeochemical Results Drill Logs (& Results) Chu Chua Drilling Assay and Lithogeochemical Results Transient Electromagnetic Survey, Logistics Report and Profiles</pre>

<u>Table of Figures</u>

		page
Figure 1	Location Map	2
Figure 2	Claim Configuration	4
Figure 3	Compilation Map	7
Figure 4	Drill Hole Plan with Surface Geology	16
Figure 5	Section 101 + 00 mN	17

List of Maps (in pocket)

Map	1	Chu Chua North (West Sheet)	-Geology
Map	2	Chu Chua North (East Sheet)	-Geology
Map	3	Lithogeochemistry (West Sheet)	-Sample Locations
Map	4	Lithogeochemistry (East Sheet)	-Sample Locations
Map	5	Lithogeochemistry (West Sheet)	-SiO ₂
Map	6	Lithogeochemistry (East Sheet)	-SiO ₂
Map	7	Lithogeochemistry (West Sheet)	-TiO ₂
Map	8	Lithogeochemistry (East Sheet)	-TiO ₂
Map	9	Lithogeochemistry (West Sheet)	-K ₂ O
Map	10	Lithogeochemistry (East Sheet)	-K ₂ O
Мар	11	Lithogeochemistry (West Sheet)	-Na ₂ O
Мар	12	Lithogeochemistry (East Sheet)	-Na ₂ O
Мар	13	Lithogeochemistry (West Sheet)	-Ba
Мар	14	Lithogeochemistry (East Sheet)	-Ba
Map	15	Lithogeochemistry (West Sheet)	-Cu
Мар	16	Lithogeochemistry (East Sheet)	-Cu
Map	17	Chu Chua Transient EM -Conductor	Axes

1. INTRODUCTION

1.1 General

The main focus of exploration on the Chu Chua property is a cuperiferous pyritic massive sulphide with minor amounts of Au, Ag, Zn, Pb, and Co. Ore reserves are currently 1 million tonnes grading 3.1% copper. The property was optioned by Minnova Inc. in August, 1985 from Pacific Cassiar Ltd., Quinterra Resources Inc., and Vestor Explorations Ltd. This report covers the Chu chua Group A, B, C, and D mining claims totalling 96, 88, 92, and 88 units respectively. The 1989 work program on this group included geological mapping, lithogeochemical sampling, a transient electromagnetic survey, and diamond drilling.

1.2 Location and Access

The Chu Chua property lies approximately 24 kilometres northeast of Barriere, B.C. on the western flank of Green Mountain (Figure 1). From Barriere, access is easiest along the paved Barriere Lakes Road to the North Barriere Lake and Birk Creek logging roads. A 4x4 dirt road from the end of the Birk Creek road provides access to the Chu Chua deposit and to the Chu Chua North grid, immediately to the north.

1.3 Topography and Climate

The claims stretch from the relatively gentle Chu Chua and Birk Creek valleys in the south to the rugged steep terrain around Cowell and Dunn Creeks and around Green Mountain. Elevations range from 900m to 2200m. Vegetation varies from clear cut, thick second growth, and dense spruce, pine, and cedar stands at lower elevations, to subalpine and alpine above 1800m. The climate is moderate to extreme with temperatures ranging from -30° C to $+35^{\circ}$ C. Precipitation is moderate; snow can be expected from October to June. Snow is present at higher elevations well into July.



1.4 Property and Ownership

All pertinent claim information for Chu Chua Groups A-D is summarized below. Figure 2 shows claim locations and configurations.

<u>Group</u>	<u>Claim</u>	<u>Units</u>	Record No.	<u>Expiry Date</u>
A	CC1 CC2 CC4	16 4 9	1154 1373 1423	Dec 29, 1998 Dec 29, 1998 Oct 10, 1992
	CC5 CC6 CC7 CC8 CC9	20 9 20 6 12	1455 1456 1457 1424 1458	Oct 24, 1990 Oct 24, 1989 Oct 24, 1990 Oct 24, 1991 Oct 10, 1991 Oct 24, 1989
в	CH2 CC10 CC11 CC1 CC9	96 20 20 16 <u>12</u> 88	1462 1459 1460 1154 1458	Oct 24, 1989 Oct 24, 1991 Oct 24, 1989 Dec 29, 1989 Oct 24, 1989
с	CH1 CH4 CC11 CC1 CC10	20 20 16 16 <u>20</u> 92	1461 1464 1471 1154 1459	Oct 24, 1991 Oct 24, 1991 Oct 24, 1991 Dec 29, 1998 Oct 24, 1991
D	CC1 CC10 CC11 CH2 CH9	16 20 20 20 <u>12</u> 88	1154 1459 1460 1462 1469	Dec 29, 1998 Oct 24, 1991 Oct 24, 1989 Oct 24, 1989 Oct 24, 1989 Oct 24, 1989

1.5 History

A massive sulphide was discovered on the CC1 claim in 1978 by Craigmont Mines Ltd. Between 1978 and 1984, they conducted linecutting, soil geochemistry, geological mapping, VLF-EM, magnetometer, and HLEM surveys over various parts of the property. Over 6000 meters of diamond drilling was completed and a Dighem AEM survey was flown over the area in 1979. Minnova Inc. (Corporation



Falconbridge Copper) acquired the property in August, 1985 and has continued with diamond drilling, soil sampling, rock geochemistry and various geophysical surveys to the present. In fall, 1988, 13 NQ diamond drillholes totalling 1151.6 meters were completed to better define near surface ore reserves (Blackadar, 1989).

1.6 Work Done

During the 1989 field season, an additional 21 diamond drillholes totalling 1662.5m were completed in the deposit area. Also, approximately 42.1 kilometres of grid cut and covered by HLEM survey in 1988 (Lear, 1989), was mapped and 300 rock samples collected for geochemical analysis. Subsequently, a grid over the deposit area was re-established and a transient electromagnetic survey was conducted by Quantech Consulting Inc. of Toronto.

2. GEOLOGY

2.1 Regional Geology

The Chu Chua property lies along the western margin of the Omineca Belt in south central B.C., in a low grade, structurally complex belt of Paleozoic volcanic and sedimentary Rocks of the Fennell Formation and the neighbouring Eagle rocks. Bay Assemblage have been divided into four structural slices separated by southwesterly directed thrust faults (Schiarizza and The lowest structural division contains Fennell Preto, 1987). rocks thrust over Eagle Bay clastic rocks (EBP). These rocks stretch north and west of Clearwater southward across and parallel to the North Thompson River to Barriere (see Figure 3).

The Fennell Formation, divided into upper and lower structural divisions, is cut by imbricated thrust faults that developed prior to metamorphism and large scale deformation. A thrust fault is thought to separate these two divisions based on conodont ages in chert beds in both the Upper and Lower Fennell (Schiarizza and Preto, 1987). Conodont ages range from Early Mississippian to Middle Permian in both divisions. Metamorphic grade is lower greenschist but many primary textures in both volcanic and sedimentary rocks are preserved. Structural deformation is not readily evident in outcrop but regional structures indicate a complex deformational history.

Schiarizza and Preto (1987) suggest that Fennell rocks extensively are imbricated, developing east verging, premetamorphic mesoscopic folds during tectonic emplacement. They further suggest that the variation in stratigraphic facies across these imbricate thrusts determines movement in the order of tens or hundreds of kilometres. Synmetamorphic deformation includes southwesterly directed folds and thrust faults that developed as the Chu Chua Fennell rocks became part of the west limb of the Slate Creek Anticline. The intrusion of the Raft and Baldy Batholiths in the late Cretaceous constituted a major late tectonic West trending folds, kinks and crenulation lineations event. developed near these intrusions but the contact metamorphic aureole



of the Baldy Batholith is limited to a few hundred meters. Finally, northeast and north trending faults developed in Tertiary times. One, the Barriere River Fault, appears to truncate the Fennell Formation at Barriere.

The Upper Fennell consists of massive and pillowed basaltic flows intruded by numerous dykes, sills, and discordant fairly continuous stocks of gabbro. Thin, chert and chert/argillite horizons are scattered throughout the sequence. Pillows and hyaloclastic breccias together with chert and minor argillite show this sequence to be submarine. Major element classification suggests these basalts are tholeiitic and AFM plots show a tholeiitic to calc-alkaline composition (Schiarizza and Similarly, discriminant diagrams using immobile Preto, 1987). elements Zr, Y, and Ti suggest these rocks are subalkaline and tholeiitic, most likely related to an ocean floor spreading centre. Aggarwal (1982) argued that immobile elements may have been mobile and that microphenocrysts of augite, amphibole (kaersutite) and Fe-Ti oxides suggest a more alkalic chemistry. He also suggests that Pb isotopes are more radiogenic than mid ocean ridge basalts and more indicative of an ocean island or seamount environment. It seems likely, however, that the Upper Fennell basalts with tholeiitic to transitional compositions formed in a marginal basin.

The Lower Fennell consists of basalts and gabbros very similar to those of the Upper Fennell, as well as chert, argillite, siltstone, sandstone, intermediate to felsic tuffs, quartz-feldspar porphyry domes, and intraformational conglomerate. These rocks are interpreted to be basinal with an ocean floor spreading centre and many off ridge volcanic vents present nearby. Sedimentary rocks dominate farther east, away from these volcanically active centres. Intraformational conglomerate indicate the presence of a high energy environment partially produced by an active tectonic regime.

2.2 Chu Chua North Geology

The north part of the Chu Chua property is underlain by Upper Fennell basalt and gabbro to the west, and mixed basalt, gabbro, and sediments to the east. These rocks strike approximately 340° and generally dip steeply to the west. Deformation features are rare in outcrop. The northeast corner of the property is underlain by Baldy granodiorite to quartz monzonite.

The contact between the upper and lower divisions of the Fennell is uncertain. Schiarizza and Preto (1987) place this contact along an assumed thrust fault on the west side of a large gabbro sill. However, mapping failed to locate any sign of faulting. Furthermore, based on lithology, a more practical division can be made at the top of a predominantly sedimentary sequence 500 meters east of the sills eastern contact.

The principal rock types are as follows (see Maps 1&2): <u>Basalt (1.1/1.2)</u>: Basalt forms massive and pillowed flows, occasionally brecciated, that are typically aphyric to weakly porphyritic with augite, hornblende (kaersutite?), and plagioclase phenocrysts. Pyrite and pyrrhotite content is typically 1-2%. There is a gradation in grain size between aphyric basalt and intrusive gabbroic sills, particularly in the southwest portion of the grid.

Diorite/Gabbro (4.1/4.2): The west side of the grid is dominated by a coarse grained gabbroic sill over 600 meters thick. This sill is apparently the same one that hosts the Windpass and Sweethome gold quartz veins, 4-5 kilometres to the north. This sill has a fairly uniform composition very similar to that of the host basalts. Two large sills are mapped in the Lower Fennell also with basaltic compositions. A hornblende porphyry is mapped immediately hangingwall to an anomolous rusty siliceous unit.

9

<u>Granodiorite/Quartz Monzonite (5.3/5.4)</u>: Coarse grained and equigranular Baldy intrusive rocks form prominent knobs and numerous boulder trains in the northeast part of the grid. A contact metamorphic aureole extends through most of the east half of the grid. Sedimentary rocks show a fine grained brownish recrystallized texture while greenstones are dark grey and fine grained.

<u>Chert/Chert Breccia (6.1/6.2)</u>: Chert varies from pale grey to green to black, is occasionally well bedded, and is usually fractured, brecciated, and mixed with other sediments. Pale, bedded chert forms lenses within basalt while dirty, disrupted cherts are more common in sedimentary sequences.

<u>Argillite (6.3)</u>: Argillite is black, sometimes fissile, usually cherty and finely interlayered with chert beds and coarse clastic horizons.

<u>Siltstone/Greywacke (6.4,6.6)</u>: Medium grained clastic beds are thin and gradational within chert/argillite. Bedding features and grain size are often obscured by thermal metamorphic effects. However, these units are often poorly sorted, usually dark greenish grey to black, and often contain cherty clasts which make them very similar to conglomerate.

<u>Conglomerate/Polylithic Breccia (6.7/6.9)</u>: Most of the clasts within this poorly sorted unit are chert with minor argillite and greenstone. These rocks occur close to the intrusive contact obscuring many details; hence, there may be overlap with chert breccia/argillite units.

10

2.3 Lithogeochemistry

2.3.1 Analytical Methods

Rock samples were collected from most mapped outcrops generally at a 25 meter spacing along grid lines. A total of 300 lithogeochemical samples were collected and sent to Min-En Laboratories of North Vancouver, B.C. for analysis. All samples were analyzed for all major elements plus eight trace elements (Ag, As, Ba, Cu, Pb, Sb, Zn, Au). A standard fusion process with ICP finish was applied for all major elements. Au was determined by wet geochemistry while aqua regia digestion with an ICP finish was used for other trace elements. All results are listed in Appendix III.

2.3.2 Discussion

Major element analysis is useful in determining variations within certain rock types in order to pick out locations of significant alteration. Among the most useful for Chu Chua north geology are SiO_2 , TiO_2 , Na_2O , and K_2O . Trace elements give a count of anomalous metal contents in these rocks. The most useful of these are Ba and Cu. Plots of the above elements with grid lines and geologic contacts are included. Results for all elements are appended.

<u>Al₂O₃</u>: Generally, Al₂O₃ is low in sediments reflecting their cherty nature (2-6%). However, some more detrital argillites and siltstones have higher Al₂O₃. Mafics have mean content of 14.8% with a standard deviation of 2.78%.

<u>BaT</u>: Mafics contain little total Ba, less than 0.1%. Sediments contain values up to 1% particularly near a mineralized chert horizon in the northeast part of the grid. <u>CaO</u>: Sediments contain low CaO with occasional highs in more detrital rocks. Values are very low within the contact metamorphic aureole. Mafics show some variation; the mean is 8.7%, the standard deviation is 2.58%.

<u>Fe₂O₃</u>: Iron content is variable in sediments but usually between 2-4%. In mafics, the mean is 10.29% and the standard deviation is 2.11%. Values are noticeably lower in the centre of the large gabbroic sill.

<u> K_2O </u>: Potassium is very low throughout most of the mafics. Anomalously high values are present in mafics immediately hangingwall to the siliceous horizon. Some footwall mafics and sediments are also anomalous.

<u>MgO</u>: The mean in mafic rocks is 6.25%, the standard deviation 2.09\%. MgO shows a fairly random pattern, although values are low in cherty sediments (0.36-4%).

<u>MnO₂</u>: Manganese content is generally low; 0.20% in mafics, 0.05% in most sediments. There is very small range of values in mafic rocks but values in chert range up to 1.91% within the siliceous horizon.

<u>Na₂O</u>: Mafics show a good normal distribution with a mean of 3.35% and standard deviation 1.14%. Sodium depletion can be seen in mafics footwall to the siliceous horizon. Sediments within the siliceous horizon are also depleted. Sodium enrichment is possible in some hangingwall rocks.

<u> P_2O_5 </u>: Values show little variation, particularly in mafics. More variation is seen in sediments; depleted values are seen in within the siliceous horizon.

12

<u>SiO</u>₂: In mafics, SiO₂ shows a tight normal distribution with a mean of 51.42% and a standard deviation of 7.83%. These values include a number of samples now believed to be hornfelsed sediments. Sediments show a wide variation; undiluted cherts are easily distinguished. Silica values are slightly higher hangingwall to the siliceous horizon.

<u>TiO</u>₂: In mafics, TiO₂ shows a slightly skewed normal distribution with a mean of 1.59% and a standard deviation of 0.47%. Values are more variable in gabbro sills.

<u>S</u>: Sulphur content is generally low. Elevated values reflect higher sulphide contents in the siliceous horizon.

<u>Ag</u>: Silver values are elevated in the 2.2 - 3.8 ppm range west of the main gabbro sill. Values are much lower within the gabbro sill. East of the sill, silver values are even lower. Values range from 0.3 - 2.3 ppm in the vicinity of the siliceous horizon. Overall, the mean is 1.39 and the standard deviation is 0.76.

<u>As</u>: Arsenic anomalies are scattered in the siliceous horizon and in a package of mafic flows hangingwall to the siliceous horizon. Values range up to 663 ppm in a good log normal distribution.

<u>Ba</u>: Barium values range up to 9433 ppm in the siliceous horizon and in the immediate footwall, generally within cherty sediments. Values are also elevated in mafics adjacent to chert horizons. Barium has a log normal distribution with a mean of 2.28 and a standard deviation of 0.63.

<u>Cu</u>: Copper anomalies are generally restricted to the siliceous horizon and immediate footwall and hangingwall rocks. Mafic rocks show several scattered anomalies. Copper shows a bimodal log normal distribution in mafic rocks with a mean of 1.50 and a standard deviation of 0.42. <u>Pb</u>: Lead values show a normal distribution with a mean of 25.15 ppm and a standard deviation of 10.67 ppm. Anomalous values are scattered in the hangingwall of the siliceous zone. Two anomalies are 362 ppm and 228 ppm.

<u>Sb</u>: Antimony anomalies are rare. Three lie in and above the siliceous horizon. One 46 ppm anomaly lies along a chert-basalt contact.

<u>Zn</u>: Zinc anomalies are scattered throughout sediments in the Lower Fennell. The mean is 53.88 ppm and the standard deviation is 22.86 ppm.

3. DIAMOND DRILLING

3.1 Program Summary

Twenty-one diamond drillholes totalling 1662.5 metres were drilled into the Chu Chua massive sulphide deposit during May The program further delineated ore reserves, and June, 1989. particularly near surface reserves in copper and gold. the core was logged, sulphide intervals were split and assayed, and much of hangingwall footwall sections were sampled for and the lithogeochemical analysis. All analyses were done by Min-En Labs All drillcore is currently being stored at of North Vancouver. Minnova's Barriere warehouse.

3.2 Geology of Chu Chua Deposit

3.2.1 General

The Chu Chua deposit consists of two major and several minor sulphide lenses hosted by massive and pillowed green basalt of the Upper Fennell Formation. Near surface ore reserves are pegged at approximately one million tonnes grading 3.1% copper. The lenses are oriented along a north-south trend dipping from vertical to very steeply west. The principal axes of the lenses appear to plunge gently to the south. The strike extension of near surface mineralization is approximately 300m and total thicknesses for the mineralized zones range up to 80m. Massive sulphide has been intersected as far as 350m below the surface. All 1989 drillholes were drilled east from hangingwall basalts through massive sulphides into a silicic or stockwork zone and finally into unaltered footwall basalts (Figures 4 and 5).

3.2.2 Hangingwall Basalt

The hangingwall to massive sulphide consists of unaltered massive and pillowed basalts. Pillow selvages and triple junctions and interpillow hyaloclastic breccias are easily identified. Most of the basalt is aphyric although faint green phenocrysts of augite and/or amphibole and pale grey phenocrysts of albitized plagioclase are discernible in some sections. Thin section and microprobe work





by Aggarwal (1982), has shown the amphibole to be kaersutite which occurs with and sometimes replaces augite. Both kaersutite and augite are often altered to actinolite and sphene, characteristic of lower greenschist metamorphic facies. These basalts are often bleached to a very pale grey and cut by abundant quartz and calcite stringers. Sulphide content increased toward the deposit but the lower contact is very abrupt with only occasional massive sulphide clasts caught up in the base of the overlying flows.

3.2.3 Massive Sulphides

Massive sulphides lie immediately below a very sharp contact with the hangingwall basalts. Pyrite makes up approximately 90% of the massive sulphide, often occurring as coarse anhedral grains displaying annealed textures. Chalcopyrite is the main ore mineral occurring as massive streaks up to 25cm thick, as small inclusions in both pyrite and magnetite, and as fracture fillings and interstices in coarse granular pyrite. These textures suggest a large degree of remobilization. Thin section work (Manley, 1988 -unpublished paper), has shown good triple junctions in granular pyrite with chalcopyrite often occurring in the interstices, as tiny anhedral blebs (50-200 micrometres), and as inclusion trails inside pyrite grains. Megascopically, sections of massive sulphide show good rolled textures and brecciation, indicating either primary collapse structures or, more likely, tectonic activity.

Other economic minerals identified in drillcore include covellite, chalcocite, sphalerite and magnetite. Cubanite ($CuFe_2S_3$) and stannite are also present (Aggarwal, 1982). Covellite occurs in chalcopyrite-rich sections as fracture fillings. Chalcocite occurs as discrete grains within either pyrite or chalcopyrite (Manley, 1988). Sphalerite and possibly trace amounts of galena occur as fine grained and massive blebs usually but not exclusively with copper mineralization. Magnetite content increases toward the footwall occurring as subhedral grains possibly mixed with or replacing pyrite. The matrix is likely quartz and barite. Other

18

metals present in the ore zone include gold (commonly 1 gpt), silver (commonly 15-30 gpt), cobalt (310-475 ppm), and trace amounts of tin (stannite), platinum, and palladium (Aggarwal, 1982).

3.2.4 Talc-Magnetite

Within the massive sulphide lenses lie several lenticular bodies of talc-magnetite. These "sublenses" have the same general shape and orientation as the main sulphide lenses but often appear to cut across the sulphides from hangingwall to footwall. Many sections show two and even three distinct parallel zones of talc and/or magnetite mineralization. Thicknesses range up to 25m but 5-10m is more the norm and the southernmost lens has a down plunge extension of at least 200m.

The talc-magnetite zones can be further broken down into sulphide-magnetite, massive magnetite, talc-magnetite, and massive talc sections, usually in that sequence from hangingwall to footwall. There is much mixing of talc and magnetite but sulphide seldom occurs with talc. Aggarwal (1982) suggested that these rocks are chemical precipitates showing a sequence of deposition at 300° C and increasing fO₂ to be:

talc --> talc + magnetite --> magnetite + pyrite --> pyrite + cpy.

This sequence agrees well with observed mineral assemblages. Furthermore, most of these rocks are fine grained and massive, with few obvious replacement textures.

Alteration of footwall and sulphide horizons also explains the presence of talc-magnetite sequences. Polished section work (Manley, 1988) clearly shows magnetite replacing pyrite in the sulphide-magnetite zone gradually overwhelming the sulphide component to form the massive magnetite zone. Talc is intimately associated with magnetite forming a fine grained matrix. The embayed and pitted nature of the magnetite in the talcmagnetite zone may indicate a further alteration eventually to a massive talc horizon. This alteration crosscuts portions of the massive sulphide lenses and helps to explain the position and variability in the zonation of the talc-magnetite horizon.

3.2.5 Silicic Rocks

A unit of very silicic rocks lies footwall to much of the Main and North Lenses and is best developed between them. They are typically 80% SiO_2 , 0.3% TiO_2 , depleted in Na_2O and CaO, and enriched in Ba. Al_2O_3 , Fe_2O_3 , and MgO are consistently lower than unaltered basalt. Thin sections show the mineralogy to be mainly quartz with fine grained phengitic mica, minor chlorite, pyrite, and chalcopyrite (Aggarwal, 1982).

In drillcore, these silicic rocks appear to be intensely silicified basalts. They have a mottled, altered appearance often with sharp colour and textural contrasts reminiscent of block boundaries, pillow selvages, and basalt breccias. Fracturing and local brecciation is common, chlorite occurs as fracture stringers. Immediate footwall zones often host pyritic stockworks associated with increased quartz-carbonate veining. Similar silicification occurs in later quartz vein selvages that cut an unaltered basaltic dyke within the silicic pile. Minor banding showing warping and truncation of bands is noted occasionally and may indicate the presence of chert horizons within the basalt package.

Aggarwal (1982) suggested that these silicic rocks are primary chemical precipitates, based on lower immobile element compositions and their fine grained nature. However, some original textures have been retained while others have been obliterated by intense quartz flooding. The presence of these rocks only between the two main massive sulphide lenses indicates that a very restricted depression existed to pool a silicic sinter or that this zone constituted an area of intense hydrothermal alteration of footwall basalt.

3.2.6 Footwall Basalt

Footwall alteration in the south part of the Main Lens and all of the North Lens is less than 2 meters and in places appears to be absent. The contact with generally unaltered footwall basalt is sharp to transitional over a few meters. Minor brecciation and quartz-pyrite stockworks die out rapidly with pyrite content down to 1-2%. The basalt becomes less bleached, increasingly competent, and primary structures such as pillow selvages become more apparent. Lower footwall basalt is virtually identical to hangingwall basalt.

3.2.7 Basalt Dykes

The deposit area is intruded by several basaltic dykes and sills. These are most easily seen cutting massive sulphide and silicic zone rock between the North and Main lenses. One dyke cuts the thick silicic package on section 102+25N in CCF-39 with a drill thickness of 12.5 meters. The orientation and continuity of this dyke is uncertain although a possible plane of failure is seen in basalt breccias in the silicic zone 25 meters to the south. Two narrow dykes were intersected in the same sequence 50 meters to the south suggesting a north-south trend. A flat lying dyke appears on section 101+25N at the top of holes CCF-28 and CCF-38. Its extent and orientation in unknown. Another possible dyke lying in the silicic zone between the North and Main lenses appears to dip vertically with an untested strike to the south. South of 100+50N, several conformable basalt flows and sills lie between fingers of massive sulphide.

3.3 Assays

3.3.1 Procedure

All massive and semi-massive sulphide and much talcmagnetite material was collected and assayed for copper, lead, zinc, silver, and gold. Specific gravities were also measured for every assay sample. Sampling intervals were chosen between 0.3 and 1.6m as dictated by visual estimates of grade. Drillcore was split or sawn in half, bagged and tagged and sent to Min-En Labs of North Vancouver for assay. A total of 329 samples were collected and assayed. All data is listed in Appendix V and with drillogs in Appendix IV.

3.3.2 Discussion

Assay results for the deposit reflect the streaky nature of chalcopyrite mineralization within massive pyrite. Highest copper grades are generally found in footwall and hangingwall zones within the massive sulphide. However, high grade streaks appear at any stratigraphic level of the massive lenses. High grade zones appear to be somewhat continuous but individual chalcopyrite streaks probably are not. Grades within high grade zones are also extremely variable depending on sample lengths, attention to mineralization by the sampler, and the amount of copper present from one hole to the next. Higher values of other metals, zinc, silver, and gold, often coincide with high grade copper values.

3.4 Lithogeochemistry

3.4.1 Analytical Methods

Continuous rock samples were taken at 1-10 meter intervals (usually 3 - 5 meters), according to geology, throughout the hangingwall, silicic, and footwall zones. Portions of talcmagnetite were also sampled. A total of 200 lithogeochemical samples were collected and sent to Min-En Laboratories of North Vancouver, B.C. for analysis. All samples were analyzed for all major elements plus eight trace elements (Ag, As, Ba, Cu, Pb, Sb, Zn, Au).

A standard fusion process with ICP finish was applied for all major elements. Au was determined by wet geochemistry while aqua regia digestion with an ICP finish was used for other trace elements. All sample data is provided in Appendix V.

3.4.2 Discussion

In hangingwall basalts several major element trends are Total barium increases dramatically over tens of significant. meters toward the deposit and is the strongest indicator of the CaO shows a subtle and sometimes substantial sulphide body. decrease in the immediate hangingwall. Na₂O becomes rapidly depleted while K,O becomes modestly enriched within a few meters of the sulphide contact. Sulphur increases sharply and Fe₂O₃ increases marginally as pyrite and magnetite contents increase in the hangingwall. Trace elements, particularly copper, lead, zinc, and silver, show relatively high background levels with occasional spiked anomalies. Arsenic and antimony show broad though modest Gold anomalies are rare. increases.

Silicic rocks show wide ranges in both major and trace element contents, reflecting varying degrees of silicification. Al_2O_3 , MgO, SiO₂, and TiO₂ numbers show this variability. Total barium exhibits sharply elevated values at or near the mineralized horizon but wide ranges of values elsewhere. CaO, Na₂O, and MnO₂ are all strongly to moderately depleted with only CaO showing any increases away from the horizon. K₂O shows dramatic increases near the ore horizon while Fe₂O₃ and S show much more subtle increases. P₂O₅ shows more variability in silicic rocks than in unaltered basalt but no real trends. Trace elements such as copper, zinc, lead and arsenic show elevated values throughout the silicic zone particularly near the sulphide horizons. Silver and antimony are very low with only occasional spot anomalies near sulphide zones. Gold anomalies are rare.

Geochemical trends seen in the hangingwall are generally reversed in the footwall. Al_2O_3 , MgO, MnO₂, P_2O_5 , SiO₂, and TiO₂ are all relatively constant. Total barium, K_2O , and S all decrease rapidly as visible footwall stockwork alteration diminishes Fe_2O_3 shows much variability in this transition zone. CaO and particularly Na₂O increase dramatically away from sulphide mineralization. Barium, copper, lead, and zinc concentrations decrease sharply while silver and arsenic concentrations increase as silicification weakens. Gold and antimony show only low background levels.

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4. TRANSIENT ELECTROMAGNETICS

4.1 Survey Logistics

A transient electromagnetic survey was conducted over the Chu Chua deposit area by Quantech Consulting Inc. of Toronto between September 18 - 28, 1989. The survey was run on 24.3 line kilometres of grid re-established along the trend of the Chu Chua mineralized horizon (Figure 3). The baseline was recut from L87N to L113N and winglines were cut or recut from 9500E to 10400E at 100m intervals. A crew of three directed by geophysicist Sherwood Coulson conducted the survey.

The instruments used in this survey include the Geonics EM-37 transient EM system and the Polycorder data recorder. Two loops approximately 400m x 900m and one loop 400m x 800m were placed to the east of the target area from 104E to 108E. Lines were read east to west and the coil orientation was up and west. A complete description of the theory, instrumentation, computer software and hardware is included in a logistics report written by S.T. Coulson for Minnova Inc. (Appendix VI).

4.2 Results and Discussion

Anomalies interpreted on each line for both the in line horizontal and vertical components were compiled on a 1:2500 grid plan of the survey area. Anomalous trends were then plotted in an attempt to locate and identify continuations of the mineralized horizon and any potential parallel horizons.

The Chu Chua massive sulphide shows very strong anomalies in both horizontal and vertical components. The anomalies become very subtle south of L99N as the deposit quickly plunges in that direction. Massive sulphides have been intersected as far south as L97N at a depth of 400m. No promising targets are indicated south of L97N. Conductivities also diminish rapidly north of known massive sulphides at L103N. However, the horizon remains discernible, though very subtle on lines 104, 105, and 106N. The anomalies become somewhat more evident again of lines 107 - 113N trending off the grid at 340° . This trend appears to be shadowed by an even more subtle anomaly 150m to the west. The more easterly trend appears to present the best potential target for deeper Chu Chua style massive sulphides.

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5. Conclusions

Work on the Chu Chua property is divided into three distinct areas, each at various stages of development.

The Chu Chua North grid is still at a reconnaissance level. Most of the area has been geologically mapped but follow up prospecting and mapping is necessary both on and off the grid. Soil sampling will be useful in better defining potentially mineralized targets. Diamond drilling will be required to fully test these targets.

The Chu Chua deposit, on the other hand, has been tested with approximately 8800 meters of diamond drilling since 1977. Open pittable reserves are pegged at 1 million tonnes grading 3.1% copper. Further shallow target diamond drilling especially south and north of recent drilling could result in a modest expansion of ore reserves. Meanwhile, production feasibility has still to be determined.

A transient electromagnetic survey (T.E.M.) run over the deposit and across possible north and south extensions has shown that any further mineralization is likely to be at depth. The best anomaly, describing a north extension of the Chu Chua horizon, should be followed up with soil sampling and geological mapping. Once the trend is accurately mapped, diamond drilling will be required to test the horizon at depth.

6. References

- Aggarwal, P.K., 1982, Geochemistry of the Chu Chua Massive Sulfide Deposit, British Columbia: Unpublished M.Sc. Thesis, University of Alberta.
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- Blackadar, D.W., 1989, Drilling Report on the CC1, CC2, and CC3 Claims: B.C. Ministry of Energy, Mines, and Petroleum Resources, Assessment Report.
- Lear, S.R., 1989, Geophysical Report Chu Chua Project, Green Mountain Group: B.C. Ministry of Energy, Mines, and Petroleum Resources, Assessment Report.
- Manley, R., 1988, A Petrologic Study of Ore from the Chu Chua Massive Sulphide Deposit: Unpublished paper.
- Schiarizza, P., and Preto, V.A., 1987, Geology of the Adams Plateau-Clearwater-Vavenby Area: B.C. Ministry of Energy, Mines, and Petroleum Resources, Paper 1979-2, 88 pages.

APPENDIX I Itemized Cost Statements

Diamond Drilling

Drilling Costs

Leclerc	Dril]	ling Ltd	., Beaver	rdell,	B.C.			
Longyear	c S38	Drill:	502.6m (9 \$69.	50/m	\$3	34,930.	70

<u>Analytical</u>

Min-En Labs, North Vancouver,	B.C.
99 assays for Cu, Pb, Zn, Ag, and SG \$37.50	Au, \$3,712.50
60 lithos for major and trace @ \$23.50	elements \$1,410.00

<u>Personnel</u> May 22 - June 30, 1989	
D. Heberlein - Project Geologist	
2 days @ \$350/day	\$700.00
C. Wild - Field Geologist	
9 days @ \$300/day	\$2700.00
T. Clarke - Field Geologist	
2 days @\$300/day	\$600.00
A. Lowe - Core Splitter	
7 days @ \$150/day	\$1050.00

Truck

9 days	@ \$50/day	\$450.00
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Food and Accomodation

20	manday	s (9	\$25/day	Υ S	\$500.	00)
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Drafting and Computer

3 days	@ \$200/day	\$600.00

<u>Shipping</u>

159 samples	6	1.00 each	\$159.00
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Miscellaneous

Office Supplies, etc.

\$25.80 \$46,838.00

Geological Mapping and Lithogeochemical Sampling

<u>Personnel</u>	July 21 - 27, 1989 August 1 - 18, 20 - 31, 1989 September 1 - 4, 1989	
D. Heberle	ein - Project Geologist	
	2 days @ \$350/day	\$700.00
C. Wild -	Field Geologist	
	40 days @ \$300/day	\$12,000.00
J. Watkins	s - Field Assistant	
	20 days @ \$150/day	\$3,000.00
S. Noble -	- Field Assistant	
	3 days @ \$150/day	\$450.00

<u>Analytical</u>

Min-En Labs, North Vancouver, B.C. 300 lithos for major and trace elements @ \$23.50 \$7050.00

<u>Truck</u>

40 days	@ \$50/day	\$2,000.00
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Food and Accommodation

62 manday	s @	\$25/day	\$1,	550.00)
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Drafting and Computer

<u> 51,600.</u>	<u>.00</u>
	<u> </u>

\$28,350.00

\$75,188.00

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Leclerc	Drilling I	td., Beaver	dell, B.C	•	
Longyear	: S38 Drill	: 670.2m @	\$69.50/m	\$46,578	3.90

<u>Analytical</u>

Min-En Labs, North Vancouver, B.C.	
133 assays for Cu, Pb, Zn, Ag, Au,	
and S.G. @ \$37.50	\$4,987.50
81 lithos for major and trace elements	
e 23.50	\$1,903.50

Personnel May 22 - June 30, 1989

I.D. Pirie - Senior Exploration Geologist	
1 day @\$500/day	\$500.00
D. Heberlein - Project Geologist	
1 day @ \$350/day	\$350.00
C. Wild - Field Geologist	
16 days @ \$300/day	\$4,800.00
J. Holland - Core Splitter	
8 days @ \$150/day	\$1,200.00

<u>Truck</u>

12 days @	\$50/day	\$600.00

Food and Accommodation

19 mandays	: @	\$25/day	\$475.00

Drafting and Computer

4 days @ \$200/day \$800.00

<u>Shipping</u>

214	samples	6	\$1.00	each	\$214.00
214	Sampres	6	\$T.00	each	ŞZ14.00

<u>Miscellaneous</u>

Office Su	upplies,	etc.	<u>\$42.10</u>
	/		\$62,451.00

Diamond Drilling

Drilling Costs

Leclerc	Drilli	.ng Ltd.,	Beaverd	ell,	B.C.		
Longyeau	c S−38	Drill:	335.1m @	\$69.	.50/m	\$23,289	.45

<u>Analytical</u>

Min-En Labs, North Vancouver, B.C.	
66 assays for Cu, Pb, Zn, Ag, Au,	
and SG @ \$37.50	\$2,475.00
40 lithos for major and trace element	ts
@ \$23.50	\$940.00

<u>Personnel</u> May 22 - June 30, 1989	
D. Heberlein - Project Geologist	
1 day @ \$350/day	\$350.00
T. Clarke - Field Geologist	
7 days @ \$300/day	\$2,100.00
C. Wild - Field Geologist	
1 day @ \$300/day	\$300.00
A. Lowe - Core Splitter	
1^{1} day A \$150/day	\$150.00

	20110 0010	1 day @ \$150/day	\$150.00
J.	Foffonoff -	Core Splitter 3 days @ \$150/day	\$450.00

<u>Truck</u>

6	days @	\$50/day	\$300.00

Food and Accommodation

13	mandays	6	\$25/day	\$325.00

Drafting and Computer

2	days	6	\$200/day	\$400.00

<u>Shipping</u>

106	camples	A	\$1 00	each	\$106.00
T 00	sampies	6	\$1.00	each	\$100.00

<u>Miscellaneous</u>

 Office Supplies, etc.
 \$39.55

 \$31,225.00

Transient Electromagnetics

Quantech Consulting Inc.,	Toronto,	Ont.	
24.3 line kilometres			\$15,000.00
			\$15,000.00

\$46,225.00

Diamond	Drilling

Leclerc	Drilli	ing Ltd.	, Beaver	dell,	B.C.		
Longyeau	r S-38	Drill:	154.6m	@ \$6	9.50/m	\$10,744.	70

<u>Analytical</u>

Min-En Labs, North Vancouver,	B.C.	
31 assays for Cu, Pb, Zn, Ag,	Au, and SG	
@ \$37.50	-	\$1,162.50
19 lithos for major and trace	elements @ \$23.50	\$446.50

Personnel May 22 - June 30, 1989	
T. Clarke - Field Geologist	
3 days @ \$300/day	\$900.00
C. Wild - Field Geologist	
1 day @ \$300/day	\$300.00
D. Feller – Core Splitter	
2 days @ \$150/day	\$300.00

Truck

3	days	6	\$50/day	\$150.00

Food and Accommodation	
4 mandays @ \$25/day	\$100.00
Drafting and Computer	
1 day @ \$200/day	\$200.00
<u>Shipping</u>	
51 samples @ \$1.00 each	\$51.00
Miscellaneous	

Office supplies,	etc	\$43.50
		\$14,400.00

APPENDIX II Statement of Qualifications

Statement of Qualifications

I, Christopher J. Wild, of the City of Burnaby, British Columbia, do hereby certify that:

- 1. I am a geologist residing at 803 5932 Patterson Avenue, Burnaby, British Columbia.
- 2. I graduated from the University of British Columbia with a B.A.Sc. in Geological Engineering in 1984.
- 3. I have worked in mining exploration since 1982.
- 4. I have been employed with Minnova Inc. on a contract basis since March 1, 1989.

Christopher J. Wild



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