

Daiwan Engineering Ltd.
1030-609 Granville Street, Vancouver, B. C. Canada. V7Y 1G5
Phone: (604) 688-1508

FILMED

**REPORT ON THE
EXPO DRILLING PROGRAM
HUSHAMU AREA
NORTHERN VANCOUVER ISLAND
BRITISH COLUMBIA, CANADA**

NTS 92L/12

Latitude / 50°40' N

Longitude / 127°50' W

For

Moraga Resources Ltd.

1030 - 609 Granville Street
Vancouver, British Columbia
V7Y 1G5

By

E. Harrington, B.Sc.

August 15, 1989

TABLE OF CONTENTS

	<u>Page</u>
SUMMARY	1
INTRODUCTION	2
LOCATION AND ACCESS	2
TOPOGRAPHY AND VEGETATION	2
PROPERTY DESCRIPTION	3
HISTORY	3
REGIONAL GEOLOGY	5
REGIONAL MINERALIZATION	9
GEOLOGY	10
DRILL PROGRAM	10
CONCLUSIONS	12
RECOMMENDATIONS	12
STATEMENT OF COSTS - Hushamu	13
STATEMENT OF COSTS - Regional	14
CERTIFICATE OF QUALIFICATIONS	15
BIBLIOGRAPHY	16

FIGURES

Figure 1	- Location Map	Following Page 2
Figure 2	- Claim Map	Following Page 3
Figure 3	- Major Mineral Deposits on Vancouver Island	Following Page 4
Figure 4	- Regional Geology	Following Page 5
Figure 5	- Drill Locations	Following Page 10
Figure 6	- Regional Sample Locations and Grid Location	In Map Pocket

APPENDICES

Appendix 1 - Claim Data	18
Appendix 2 - Sample Descriptions	19
Appendix 3 - Assay Certificates	27
Appendix 4 - Drill Logs	28

SUMMARY

Between November 11 and November 26, 1988, Tonto Drilling drilled five NQ wireline holes, EC-159-163, for a total of 762.7m (2502 ft). In total 306 split-core samples were assayed by Acme Analytical Laboratories Ltd., Vancouver, B.C.

The drill program substantiated the grade of the Hushamu deposit as indicated by BHP-Utah Mines Ltd.

No new bodies of gold or copper mineralization was indicated but further work in the area is warranted with the possibility of upgrading the tonnage at Hushamu to an economic level.

Between November 1, 1988 and December 22, 1988, Daiwan Engineering Ltd. conducted a regional mapping and sampling program over the entire Expo claim group. The Red Dog, McIntosh, and Hushamu claim groups were the target areas, a total of 125 rock samples were collected and assayed. A total of \$62,918.77 was spent and was divided between the three claim groups.

A total of \$128,483.35 was spent on the Hushamu Claim Group between November 1 and December 22, 1988.

INTRODUCTION

Mr. Maurice Young, president of Moraga Resources Ltd. requested Daiwan Engineering Ltd. to conduct a drilling program on the Expo property in the Hushamu Lake area near Holberg, British Columbia. The program was designed to further investigate gold mineralization in the western section of the Hushamu copper/gold deposit.

The split core samples were shipped by bus to Acme Analytical Labs in Vancouver, B.C. The samples were ground to -80 mesh then .500 gm was digested in 3ml 3-1-2 HCL-HN03-H2O at 95°C for one hour then diluted to 10ml with water. Gold analysis was by acid leach with an AA finish from a 10gm sample.

LOCATION AND ACCESS

The Expo property is located on northern Vancouver Island, approximately 360 km (225 miles) northwest of Vancouver, British Columbia, Canada (Figure 1). Locally this large claim group covers a 20 km (13 mile) stretch of ground immediately north, and parallel to the west end of Holberg Inlet on N.T.S. topographic map 92L/12. The Hushamu claim group consist of 97 contiguous 2-post claims in the centre of the Expo property (see Figure 2). Most areas of the property can be reached by well maintained logging roads and forest tracks. The main access to the claim block is by forest road "Wanokana Main" which commences on the outskirts of Coal Harbour.

Regular Boeing 737 or Dash 7 air service is provided by both Canadian Airlines and Time Air from Vancouver to Port Hardy, each on a twice daily schedule. Alternately, there is good highway access, with travel from Vancouver taking 7 hours.

Port Hardy is the local commercial centre, but there are forestry and fishing centres at Coal Harbour and Holberg.

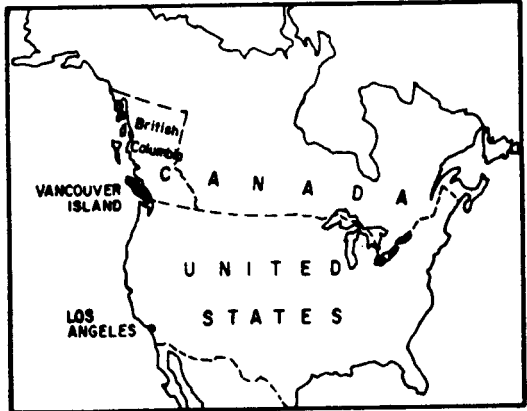
TOPOGRAPHY AND VEGETATION

The property is characterized by many low, northwest to westerly trending hills and ridges bounded by narrow deeply incised valleys and steep slopes. Elevations range from sea level to over 600 metres (2,000 ft). Within the claim block ridge tops are commonly about 300 metres (1,000 ft) above valley bottom. The property is within N.T.S. topographic map 92L/12.



EXPO PROJECT

VANCOUVER ISLAND



MORAGA RESOURCES LTD.		
EXPO PROJECT NANAIMO MINING DIVISION, B.C.		
LOCATION MAP		
DAIWAN ENGINEERING LTD.		
SCALE 1:8,000,000	DATE SEPT., 1989	FIG. 1

The claims are located within an active logging area, consequently forest cover varies from mature stands of fir, hemlock, spruce and cedar to dense second growth or to open clear-cut areas of recent logging. Some of the ridge tops are fairly open with only stunted evergreens. Low areas, especially along creeks have thick brush and berry bushes.

Rock exposure is well defined in the areas of high relief, and on the higher ridges. However, thick humus development on the forested slopes and scattered residual glacial gravels in the valley bottoms restrict geological mapping in these areas.

PROPERTY DESCRIPTION

The Hushamu Group consists of 97 contiguous 2 post claims in the Nanaimo Mining Division. A complete list of the claims can be found in Appendix 1.

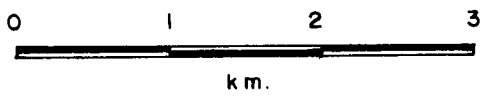
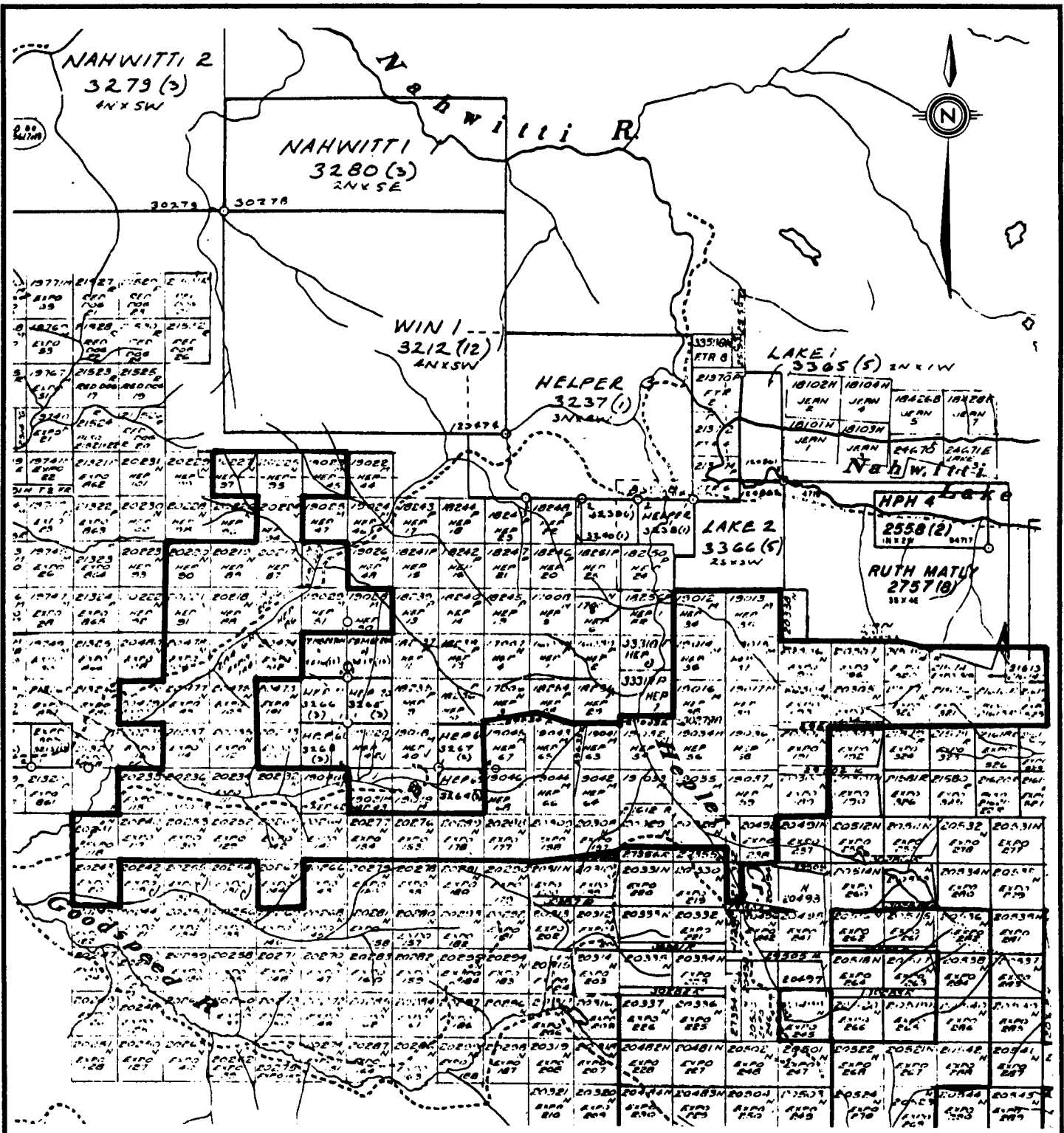
HISTORY

Northern Vancouver Island has been intermittently explored since the early 1800s. Between 1849 and 1920 several attempts were made at mining coal in the Port Hardy area. These operations failed due to the poor quality of coal. In the early 1900s, minor coal was mined at Coal Harbour, located on the north shore of Holberg Inlet.

Copper was discovered in 1911 at Benson Lake, 25 miles southeast of Port Hardy. This property, now known as Coast Copper Mine, was acquired by Cominco in 1916. They carried out considerable underground development work, but closed down in 1931, remaining idle until 1960. It was then actively mined between 1962-1972 producing copper and iron concentrates which were shipped to Japan.

Magnetite occurrences were located in the Benson Lake area in 1897, but were considered of interest only for their copper content until the early 1950s. They were explored for their iron content between 1950-56, then mined until 1967 when the operation ceased. Iron concentrates were also shipped to Japan.

In 1963, the B.C. Department of Mines published the results of a recently completed aeromagnetic survey covering the northern end of Vancouver Island.² Since magnetite deposits were of interest at this time, considerable exploration activity was generated in the area examining all magnetic anomalies of interest.



MORAGA RESOURCES LTD.		
EXPO PROJECT		
NANAIMO MINING DIVISION, B.C.		
HUSHAMU AREA		
CLAIM MAP		
DAIWAN ENGINEERING LTD.		
SCALE	DATE	FIG.
1 : 50,000	SEPT., 1989	2

One magnetic anomaly of fairly large areal extent was recorded on the eastern end of Rupert Inlet. Diligent prospecting in this area located a number of poorly exposed copper occurrences. A large number of claims were located in 1966 and subsequently the property was acquired by Utah Construction and Mining Company, now BHP-Utah Mines Ltd. Over the years, they added to the claim block and conducted extensive geological-geochemical-geophysical surveys and diamond drilling throughout the claim block. This work resulted in locating the large copper-molybdenum deposit which was developed into Island Copper Mine (Figure 3). The mine commenced production in October 1971. Production to 1987 has been in excess of 200 million tonnes milled, for concentrate sales of 753,000 tonnes of copper, 23.1 million grams gold, 168 million grams silver, and 15.3 tonnes molybdenum¹⁴.

With the discovery of significant copper mineralization on the Utah property, a great deal of interest was generated in the area by individuals and companies searching for copper. Many copper occurrences were located but none were found to be economic.

During the height of the exploration activity, Utah Mines Ltd. controlled most of the ground extending from the east end of Rupert Inlet to the west end of Holberg Inlet. Their properties included the large block of claims covering the Island Copper deposit, as well as the favourable geology on trend to the northwest (most of the present Expo group). After exploring the area extensively to 1975, Utah dropped some of the claims but retained the Expo group. Exploration on these claims had located a large area of low grade copper-molybdenum mineralization (the Hushamu zone) estimated to contain 58,420,000 mineable tonnes grading 0.32% Cu, 0.008% Mo and 413 ppb gold with a stripping ratio of 2.21:1. The drill indicated reserve for the deposit is over 100 million tonnes at the same grade, but higher stripping ratio. ⁴

A number of other alteration zones, similar to that at Island Copper Mine and the Hushamu zone, were investigated. While some were mineralized, they were not significant enough at the time to warrant further development.

The Hushamu deposit, and these other alteration zones, are the targets for Moraga's gold and copper exploration. The urgency for developing a further copper deposit in the area is prompted by the expected closure of the Island Copper Mine in 1996 due to the exhaustion of the pit reserves.

Moraga has completed two preliminary phases of exploration since obtaining the property option. The first groundwork was a Downhole Pulse Electromagnetic Survey of DDH EC-158 on Pemberton Hills. This survey indicated a sheet-like sulphide horizon with a significantly more responsive sulphide zone to the north-northwest of the present drillhole. This is awaiting further fieldwork.

MORAGA RESOURCES LTD.

EXPO PROJECT

NANAIMO MINING DIVISION, B.C.

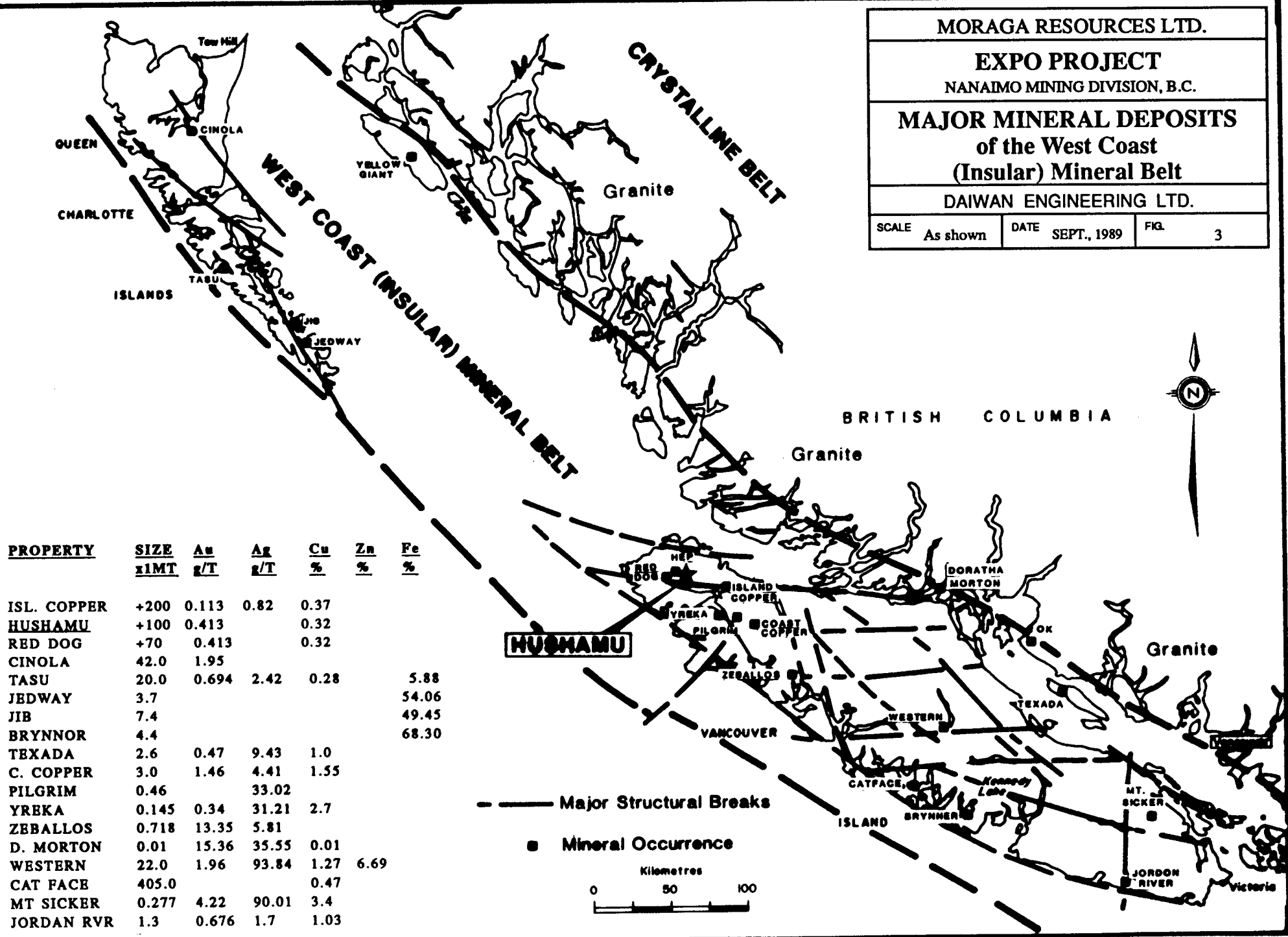
MAJOR MINERAL DEPOSITS
of the West Coast
(Insular) Mineral Belt

DAIWAN ENGINEERING LTD.

SCALE As shown

DATE SEPT., 1989

FIG. 3



PROPERTY	SIZE x1MT	Au g/T	Ag g/T	Cu %	Zn %	Fe %
ISL. COPPER	+200	0.113	0.82	0.37		
HUSHAMU	+100	0.413		0.32		
RED DOG	+70	0.413		0.32		
CINOLA	42.0	1.95				
TASU	20.0	0.694	2.42	0.28		5.88
JEDWAY	3.7					54.06
JIB	7.4					49.45
BRYNNOR	4.4					68.30
TEXADA	2.6	0.47	9.43	1.0		
C. COPPER	3.0	1.46	4.41	1.55		
PILGRIM	0.46		33.02			
YREKA	0.145	0.34	31.21	2.7		
ZEBALLOS	0.718	13.35	5.81			
D. MORTON	0.01	15.36	35.55	0.01		
WESTERN	22.0	1.96	93.84	1.27	6.69	
CAT FACE	405.0			0.47		
MT SICKER	0.277	4.22	90.01	3.4		
JORDAN RVR	1.3	0.676	1.7	1.03		

The second programme commenced in late November of 1988, on the completion of the Company's public financing, and has included regional mapping with road cut sampling, computer modelling of the 1963 airborne geophysical data, and a 762 metre (2,500 feet) drill programme. In addition archived soil sample rejects were recovered from storage and analyzed for gold arsenic, selenium, tellurium, bismuth, and antimony mineralization. These samples had previously been analyzed for copper, lead and zinc.

REGIONAL GEOLOGY

Vancouver Island, north of Holberg and Rupert Inlets, is underlain by rocks of the Vancouver Group. These rocks range in age from Upper Triassic to Lower Jurassic. They are intruded by rocks of Jurassic and Tertiary age and disconformably overlain by Cretaceous sedimentary rocks. The structural setting of Vancouver Island is shown in Figure 3. Figure 4 shows the 1:50,000 geological mapping of the northern part of the Island.

Faulting is prevalent in the area. Large-scale block faults with hundreds to thousands of metres of displacement are offset by younger strike-slip faults with displacements up to 750 metres (2,500 feet).

The Vancouver Group is described as follows:⁶

(a) **Basal Sediment - Sill Unit: Middle and Upper Triassic Age**

The basal sediment-sill unit consists of laminated to graded-bedded black shales and siltstones, silicified and invaded by diabase sills. The entire unit is estimated as 750-900 metres (2,500-3,000 feet) with the sedimentary portion being about 180 metres (600 feet) thick.

(b) **Karmutsen Formation: Upper Triassic Age**

Karmutsen Formation consists of 3,000-6,000 metres (10-20,000 feet) of volcanic flows, pyroclastics and minor sediments. It includes three distinct units: a lower pillow lava unit, a middle pillow breccia unit, and an upper lava flow unit. The latter consists of predominantly porphyritic and amygdaloidal basalt flows, individual flows of which range from 1-30 metres (to 100 feet) thick.

Two thin bands of limestone occur near the top of the Karmutsen Formation. The distribution of limestone outcrops is erratic and suggests a series of lenses at the same general stratigraphic horizon rather than one continuous bed.

MORAGA RESOURCES LTD.

EXPO PROJECT
NANAIMO MINING DIVISION, B.C.

REGIONAL GEOLOGY

DAIWAN ENGINEERING LTD.

SCALE As Shown DATE SEPT., 1989 FIG. 4

--- Airborne Magnetics Lineaments
--- Ground Mapping (Faults)
From BCDM Aerial Survey, 1963

PROPERTY

**Hughes
Copper - Gold Deposit**
> 100M Tonnes
0.32% Cu, 0.000% Mo, 0.413gm./t. Au

LEGEND

JURASSIC AND CRETACEOUS

1 Cretaceous Sediments

JURASSIC

Island Intrusions

2 Diorite

3 Quartz Feldspar Porphyry

LOWER JURASSIC

4 Bonanza Volcanics

TRIASSIC

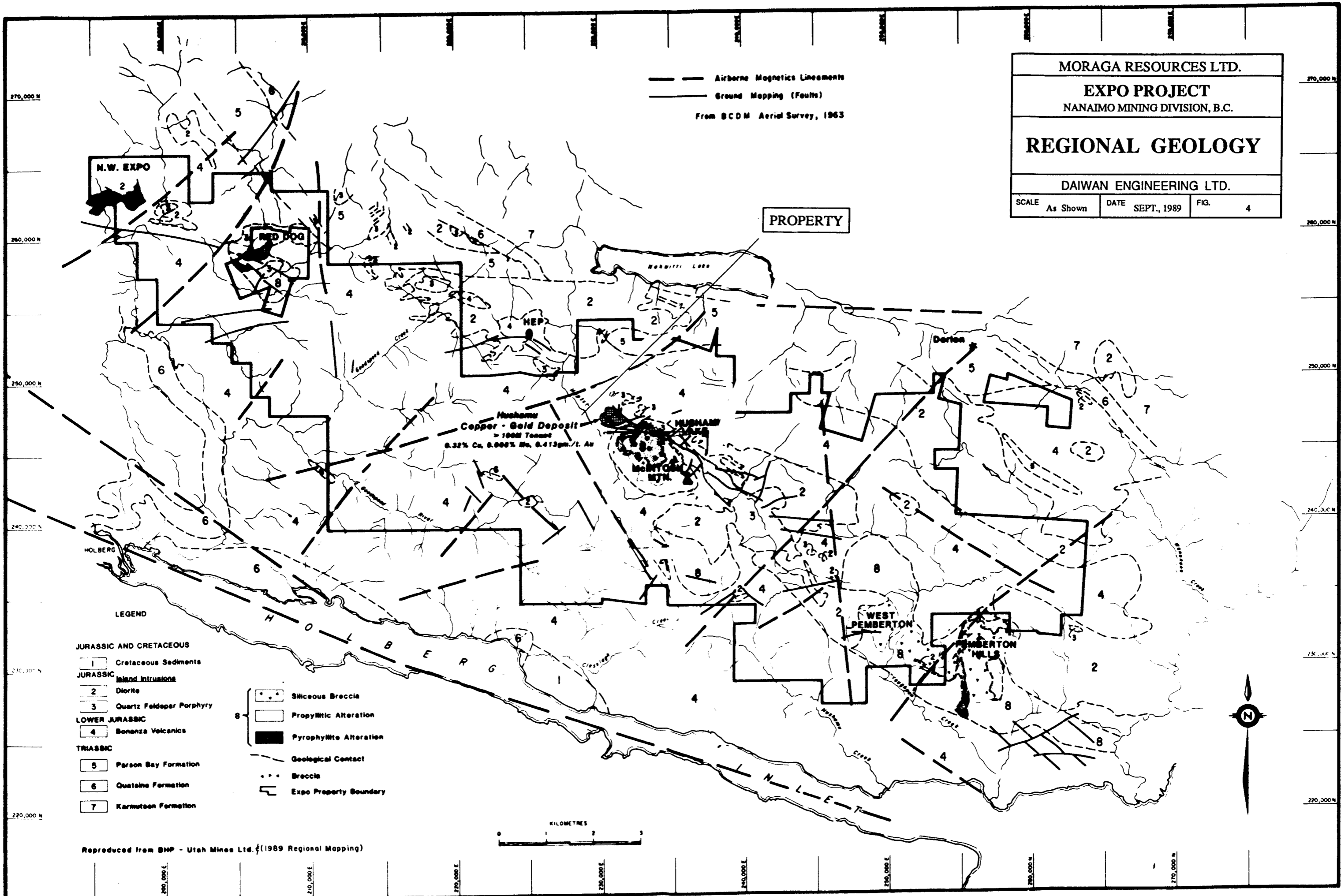
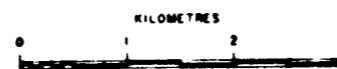
5 Parson Bay Formation

6 Oostana Formation

7 Karmutsen Formation

8 Siliceous Breccia
Propylitic Alteration
Pyrophyllite Alteration
Geological Contact
Breccia
Expo Property Boundary

Reproduced from BHP - Utah Mines Ltd. (1989 Regional Mapping)



The lower contact of the formation has not been observed on the northern part of Vancouver Island. The upper contact with limestone of the Quatsino Formation generally is sharp and easily recognized, although limestones and basalt locally are interbedded over a narrow stratigraphic interval at this contact.

Low-grade metamorphism of the Karmutsen Formation rocks has resulted in pervasive chloritization and amygdules filled with epidote, carbonate, zeolite, prehnite, chlorite, and quartz.

Basaltic rocks along contacts with intrusive stocks are in many places converted to dark-coloured hornblende hornfels. Skarn zones occur sporadically along these contacts, both in the inter-lava limestones and in the basalts.

(c) **Quatsino Formation: Upper Triassic Age**

The Quatsino Formation ranges from 60-1,000 metres (200-3,500 feet) in thickness and consists almost entirely of limestone with a few thin andesite or basalt flows. It has conformable contacts with both the overlying Parson's Bay sediments and the underlying Karmutsen volcanics. The upper contact with the Parson's Bay Formation is gradational with limestone grading upward into carbonaceous argillites.

Within the contact metamorphic/metasomatic aureoles adjacent to intrusive stocks, skarn development and silicification of limestone, accompanied by chalcopyrite-magnetite or galena, sphalerite and silver mineralization has been noted.

(d) **Parson's Bay Formation: Upper Triassic Age**

The Parson's Bay Formation consists of between 60-360 metres (200-1,200 feet) of argillite, minor limestone, agglomeratic and tuffaceous limestone, tuff, quartzite and minor conglomerate. At both its base and top, the unit exhibits gradational contacts with the Quatsino and Harbledown Formations.

On a regional scale, the rocks are unmetamorphosed. Locally, adjacent to intrusive contacts, pyrite-magnetite replacement bands up to one-half inch thick in banded tuffs have been observed.

(e) **Harbledown Formation: Lower Jurassic Age**

The Harbledown Formation consists of 485 metres (1,600 feet), a non-volcanic argillite-greywacke sequence separating the Parson's Bay from the Bonanza Formation.

(f) Bonanza Formation: Lower Jurassic Age

The Bonanza Formation is approximately 1,500 metres (8,500 feet) thick. The lower portion consists of bedded and massive tuffs, formational breccias and rare amygdaloidal and porphyritic flows, in the compositional range andesite to basalt. Porphyritic dykes and sills intrude the lower part of the unit. In the upper part of the Bonanza, rhyodacite flows and breccias become more numerous and are interbedded with andesite and basalt flows, tuffs and tuff breccias.

Regional metamorphism within the Bonanza Volcanics is very low grade, possibly zeolite facies. Plagioclase commonly is albitized and saussuritized. Chlorite, epidote and laumontite occur within the matrix of volcanic breccias, in veinlets, and in amygdules. Coarse intraformational breccias locally are hematized.

Biotite and amphibolite hornfels occur adjacent to stocks which intrude the Bonanza Volcanics.

"Pyrobitumen", a black hydrocarbon erratically distributed within the Bonanza rocks, generally occurs as fracture fillings or in the centre of zeolite-carbonate veins. Its distribution is not related to the position of the intrusive stocks.

Cretaceous Sediments

The Vancouver Group is unconformably overlain by non-marine Cretaceous sediments of the Longarm Formation which are estimated to be about 300 metres (1,000 feet) thick in the Port Hardy area. These sediments, consisting of conglomerate, sandstone, greywacke, and siltstone and some carbonaceous and impure coal seams, occupy local basins. Early coal mining in the district was from several of these basins.

Intrusive Rocks

The Vancouver Group rocks are intruded by a number of Jurassic-aged stocks and batholiths. In the Holberg Inlet area a belt of northwest-trending stocks extend from the east end of Rupert Inlet to the mouth of Stranby River on the north coast of Vancouver Island¹⁵.

Quartz-feldspar porphyry dikes and irregular bodies occur along the south edge of the belt of stocks. Dykes are characterized by coarse, subhedral quartz and plagioclase phenocrysts set in a pink, very fine grained, quartz and feldspar matrix. They are commonly extensively altered and pyritized. At

Island Copper Mine, these porphyries are enveloped by altered, brecciated, mineralized Bonanza wallrocks. The porphyries, too, are cut by siliceous veins, pyritized, extensively altered, and are mineralized where they have been brecciated. The quartz-feldspar porphyries are thought to be differentiates of middle Jurassic, felsic, intrusive rocks.

Other intrusive rocks of lesser significance include felsic dykes and sills around the margins of some intrusive stocks; dykes of andesitic composition, which cut the Karmutsen, Quatsino and Parson's Bay Formation, and represent feeders for Bonanza volcanism; and Tertiary basalt-dacite dykes intruding Cretaceous sediments.

Structure

The structure of the rocks north of Holberg and Rupert Inlets is that of shallow synclinal folding along a northwesterly fold axis. The steeper southwesterly limbs of the folds have apparently been truncated by faults roughly parallel to the fold axis. Failure of limestone during folding may have influenced the location of some of the faulting as indicated by their proximity of the Dawson and Stranby River Faults to the Quatsino horizon. Transverse faulting is pronounced and manifested by numerous north and northeasterly trending faults and topographic lineaments (Figure 3).

The northern part of Vancouver Island lies in a block faulted structural setting with post Lower Cretaceous northwesterly trending faults apparently being the major system (Figure 4). This system causes both repetition and loss of parts of the stratigraphic section, with aggregate movement in a vertical sense in the order of tens to hundreds of metres. The most significant of these fault systems trends west to northwest following Rupert and Holberg Inlets. Near the west end of the Holberg Inlet it splits with the main branch following the Holberg Inlet, the other branch passing through the west side of the Stranby Valley. Another northwesterly to westerly system passes through William Lake and still another smaller system passes through Nahwitti Lake.⁶

Northeasterly trending faults comprise a subordinate fault system. In some cases, apparent lateral displacement, in the order of a several hundred metres, can be measured on certain horizons. Movement, however, could be entirely vertical with the apparent offset resulting from the regional dip of the beds.

Recent computer modelling of the airborne magnetometer data has provided a very clear understanding of the relationship of secondary conjugate sets of northeast and north westerly faults related to the major west-northwest trending breaks.⁷ These conjugate fault sets appear to relate directly to the significant mineralization at the Island Copper, Hushamu, Hep and Red Dog copper/gold deposits. Using these relationships further similar zones are evident on the Expo claim block.

Generally, regional dip of the bedding is gentle to moderate southwesterly. Locally, in the area west of Holberg, dips are much steeper, but these are in close proximity to major faults. There is little folding or flexuring of bedding visible, except along loci of major faults where it is particularly conspicuous in thinly bedded sediments of Lower Bonanza. Bedding is generally inconspicuous in massive beds of Karmutsen, Quatsino and Bonanza rocks, particularly inland where outcrops are widely scattered.

REGIONAL MINERALIZATION

A number of types of mineral occurrences are known on Northern Vancouver Island. These include:

1. Skarn deposits: copper-iron and lead-zinc skarns
2. Copper in basic volcanic rocks (Karmutsen): in amygdules, fractures, small shears and quartz-carbonate veins, with no apparent relationship to intrusive activity
3. Veins: with gold and/or base metal sulphides, related to intrusive rocks
4. Porphyry copper deposits: largely in the country rock surrounding or enveloping granitic rocks and their porphyritic phases.

Figure 3 indicates the large size of the Island Copper - Expo projects and the relationship of mineral deposits in the West Coast Mineral Belt to the major regional structures. Figure 5 shows the location and diversity of the mineral deposits recognized in the North Vancouver Island area⁶, and highlights the extensive central location of the Expo claim holding.

Utah Mines Ltd., in their many years of exploration in the Holberg-Rupert Inlets area, focused their attention on the search for copper porphyry deposits. Their exploration resulted in locating and developing the Island Copper Mine. In addition, they located other areas of porphyry mineralization, as well as two areas anomalous in gold and one area with massive sulphide mineralization within the Expo group claim area. Moraga's efforts in the past 6 months have been directed at verifying the extent of the gold mineralization within the copper deposits; detailing and testing the presently identified adjacent epithermal gold mineralization; and relating the sedimentary sulphide horizons to similar style mineralization presently mined by Western Mines at Buttle Lake to the south.

GEOLOGY

Vancouver Island, north of Holberg and Rupert Inlets, is underlain by rocks of the Vancouver Group consisting of (with decreasing age) Karmutsen Formation, Quatsino Formation, Parson Bay Formation, Harbledown Formation and Bonanza Formation.

In the drilled area the rocks consist predominately of the Bonanza Formation which is generally characterized by andesitic to basaltic bedded and massive tuffs, formational breccias and rare amygdaloidal and porphyritic flows. In the upper Bonanza the flows and breccias become rhyodacitic and are interbedded with andesite and basalt flows, tuffs and tuff breccias.

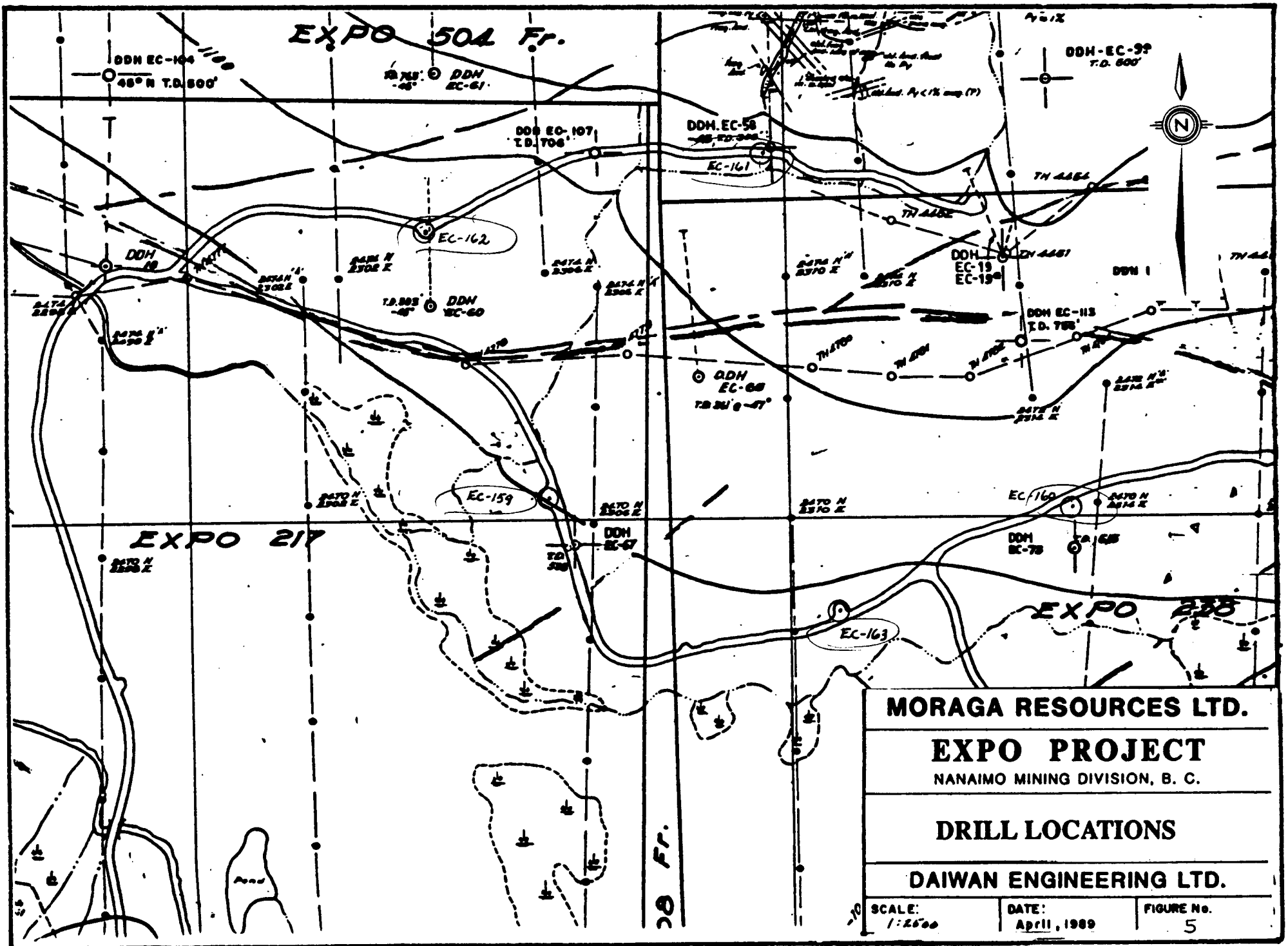
The volcanics are intruded by quartz-feldspar porphyry. These Island Intrusions give rise to localized propylitic and pyrophyllitic alteration.

DRILL PROGRAM

The drill program, using NQ wireline, was carried out from November 11 to November 26, 1988 and consisted of 726.7m (2502 ft) of drilling in five holes. The drilling information is summarized in Table 1. The core is stored at the BHP-Utah Mines Island Copper site at Coal Harbour.

Table 1 Drill Summary

<u>Hole #</u>	<u>Dip</u>	<u>Azimuth</u>	<u>Depth</u>	<u>Reason for Drilling</u>	<u>Location</u>
EC-159	-60°	007°	274.3m/900ft	To test the south boundary of the Hushamu deposit	230591E/247032N
EC-160	-60°	360°	75.6m/248ft	To undercut silicified zone with 200 ppb Au	231492E/247000N
EC-161	-45°	270°	37.5m/123ft	To give E-W section of Hushamu deposit	230984E/247623N
EC-162	-60°	090°	187.2m/614ft	To give E-W section of Hushamu deposit	230394E/247492N
EC-163	-60°	360°	<u>188.1m/617ft</u> 762.7m/2,502ft	To test south boundary of Hushamu deposit	231082E/246843N



MORAGA RESOURCES LTD.

EXPO PROJECT

NANAIMO MINING DIVISION, B. C.

DRILL LOCATIONS

DAIWAN ENGINEERING LTD.

SCALE:
1:2500

DATE:
April, 1989

FIGURE No.
5

EC-159 is composed of andesitic tuffs, flows and breccias with a 7m. fault zone at 93-100m. followed by a 3m. quartz-feldspar porphyry dyke at 101.7 - 104.9m. The hole ends in 7.3m. of granodiorite from 267-274.3m. Pyrite fracture fillings are prominent and quartz, quartz/carbonate and zeolite veining is common.

The highest copper and gold values obtained were 1.1% Cu at 123.0 - 126.0m and 52 ppb Au from 27.0 - 30.0m.

The strongest mineralization occurs between 12.19 - 78.0m where wt.ave. values are .0498% Cu and 26.9 ppb Au.

EC-160 is primarily composed of well fractured silicified andesite with some quartz and quartz/carbonate veining. The hole was abandoned at 75.6m. in fault gouge.

The highest single values were obtained at 15.0 - 18.0m in fault gouge with 410 ppb Au and .3713% Cu.

A 23.4m section from 12.6 - 36.0m returned wt. ave. of 225 ppb Au and .218% Cu. while the 39.6m section, from 36.0 - 75.6m, returned 64 ppb Au and .126% Cu.

EC-161 was abandoned at 37.5m. due to excessive caving. Only 17.5m. of the rock drilled was recovered with the rest lost to triconing. The hole is composed of andesite overlying quartz diorite with the quartz diorite carrying the better values.

The highest values returned were 430 ppb Au and .4648% Cu from a 1.68m section at 24.38 - 26.06m.

There are two separate core intervals divided by a 6.12m. section of triconed lost core. The upper section from 12.19 - 15.24m returned wt. ave. of 167 ppb Au and .1297% Cu while the lower section from 21.36 - 26.06m returned 276 ppb Au and .2443% Cu.

Andesite porphyry, tuff and breccias with strong shearing and gouge zones compose EC-162. The highest values of 910 ppb Au and .9330% Cu were obtained from a .3m sample at 111.1 - 111.4. Overall, the 114.8 m. section from 6.71 - 121.5m returned wt. ave. values of 349 ppb Au and .3027% Cu. Within this section are three mineralized areas of note.

6.71 - 18.0m (11.29m)	644 ppb Au, .4554% Cu.
52.0 - 67.2m (15.2m)	396 ppb Au, .2828% Cu.
110.0 - 116.0m (6.0m)	397 ppb Au, .4209% Cu.

Samples #34722-25 from 107.0 - 112.4m returned values of 127,125,156, and 84 ppm Mo.

EC-163 consists of andesite flows with zones of strong brecciation and gouge. The highest values were obtained at 59.7 - 62.0m with 1.379% Cu and at 75.3 - 76.3m with 480 ppb Au. The best mineralization occurs from 6.198.0m. (91.9m) where wt. ave. values are 134 ppb Au and .198% Cu.

CONCLUSIONS

The drilling has substantiated the gold and copper values indicated by the previous work of BHP-Utah Mines Ltd.

Both gold and copper mineralization appear to be directly related (eg. high Cu = high Au). Values are carried by pyrite and pyrite/chalcopyrite mineralization associated with quartz and quartz/carbonate veining as well as general disseminations in various rock types. Elevated Au and Cu values also appear to be related to rocks exhibiting moderate to strong silicification. Within the Bonanza volcanics drilled there does not appear to be a specific lithologic host for either gold or copper mineralization. Therefore mineralization is likely epigenetic with areas of brecciation, silicification and subsequent quartz veining being prime targets.

RECOMMENDATIONS

Further exploration should be conducted in the surrounding area with special attention paid to areas of increased silicification and quartz veining. These areas could be delineated using ground geophysics and geochemistry with follow-up programs of prospecting and geological mapping.

STATEMENT OF COSTS - Hushamu

1.0	Personnel		
	1 Senior Geologist	17.05 days @ \$380	\$ 6,478.51
	1 Project Geologist	58 days @ \$240	13,920
	1 Field Assistant	8 days @ \$105	<u>840.00</u>
			21,238.51
2.0	Assays		
	324 Core		\$ 5,310.18
3.0	Food & Accommodation		\$ 2,059.44
4.0	Transportation		\$ 5,593.66
5.0	Supplies		\$ 900.00
6.0	Equipment Rental		\$ 823.92
7.0	Cat Contract		\$ 1,145.89
8.0	Office		\$ 248.87
9.0	Miscellaneous		\$ 474.16
10.0	Drilling Contractor		\$ 69,715.80
11.0	Regional Mapping Proportion (see attached schedule)		<u>\$ 20,972.92</u>
			<u>\$128,483.35</u>

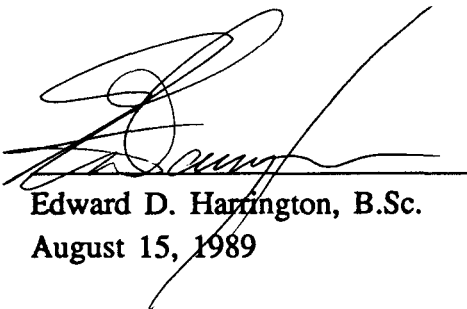
STATEMENT OF COSTS - Regional

1.0	Personnel		
	1 Senior Geologist	19 days @ \$380/day	\$ 7,220.00
	1 Project Geologist	46 days @ \$360/day	17,760.00
	1 Geologist	2 days @ \$260/day	520.00
	1 Geologist	20.1 days @ \$240/day	4,824.00
	1 Draftsperson	4.4 days @ \$200/day	<u>880.00</u>
			31,204.00
2.0	Assays		
	125 Rocks	30 Element I.C.P.	2,248.92
	3 Rocks F.A.		30.60
	7 Rocks Geochem Au		69.00
	4 Rocks Ag, Au F.A.		<u>72.96</u>
			2,421.48
3.0	Food and Accommodation		2,932.63
4.0	Transportation		5,708.22
5.0	Supplies		2,681.93
6.0	Maps		1,792.94
7.0	Office		633.95
8.0	Miscellaneous		833.16
8.0	Drafting		<u>14,709.96</u>
			<u>\$62,918.77</u>

CERTIFICATE OF QUALIFICATIONS

I Edward D. Harrington of the City of Vancouver, British Columbia, do hereby certify that:

1. I am a geologist employed by Daiwan Engineering Ltd. with offices of 1030 - 609 Granville Street, Vancouver, British Columbia, V7Y 1G5.
2. I am a graduate of Acadia University, Wolfville, Nova Scotia, with a B.Sc. degree in Geology (1971).
3. I have practiced my profession as a Geologist for over 13 years.
4. This report is based on my personal knowledge of the Expo Property, a review of geological and geochemical data from previous work and field work under my personal supervision.
5. The work program was carried out from November 11 to 26, 1988.
6. I have no interest in the property of shares of Moraga Resources Ltd. or in any of the companies with contiguous property to their Expo claim blocks, nor do I expect to receive any.



Edward D. Harrington, B.Sc.
August 15, 1989

BIBLIOGRAPHY

1. Jones H.M. 1988 A report on the Expo property Holberg Inlet for Moraga Resources Ltd. Filed V.S.E. Qualifying Report.
2. _____
(1963) G.S.C. Geophysics Paper 1734 incl. maps 1734G, 1738G.
3. Gableman, J.W.
(1982) Geological Occurrence of Gold at Island Copper Internal Report for Utah Mines, Dec. 1982.
4. Pickering, Schmidt Ore Reserve Hushamu. Internal report for Island Copper Mines (1983)Ltd.
5. Northcote, K.E. (1970) - Rupert Inlet - Cape Scott Map - Area in B.C. Dept. of Mines Petrol. Res. G.E.M. 1970, pp. 250-278.

(1972) - Island Copper, In B.C. Dept. Mines Petrol. Res. G.E.M. 1972, pp. 293-298.
6. Muller, J.E.,
Northcote, K.E. and
Carlisle, D. (1974) Geology and Mineral Deposits of Alert Bay - Cape Scott Map - Area, British Columbia, Geol. Surv. Canada Paper 74-8.
7. Utah Mines Ltd. Private company files. Reports by E.S. Rugg, B. Bowen, and B. Richards.
8. Kesler, S.E. (1985) Report on Geological Review of the McIntosh - Pemberton Precious metal Exploration Area. Private Report to BHP-Utah Mines Ltd., June 1985.
9. Magnum, M. (1988) V.S.E. Listing Report Silver Drake Resources Ltd. - July 12, 1988.
10. _____ 1988 V.S.E. News Release 4 Jan 1989 - Crew Natural Resources Ltd. George Cross News Letter - November 23, 1988.
11. Adamson R. (1989) Correspondence to Daiwan Engineering Ltd, and personal discussions April 1989.

12. Sillitoe, R.H. (1980) Styles of low-grade gold mineralization in Volcano-plutonic areas.
13. McMillan et al (1986) Mineral Deposits in British Columbia - A review of their Tectonic settings. Paper in Geoexpo conference Vancouver, B.C. 1986.
14. BCDM Records Island Copper Statistics, B.C. Dept. of Mines records to 1984 with estimates to end of 1986.
15. Carson DJT 1972 The plutonic rocks of Vancouver Island - Page 72-44.
16. Woods, D.V. (1987) Geophysical Report on Reconnaissance Surface and Bore Hole Pulse Electro, magnetic Survey on the Expo Project, Vancouver Island, report for Moraga Resources Ltd.
17. Dasler, P. G. (1989) Report on the Expo Group Northern Vancouver Island for Moraga Resources Ltd.

APPENDIX 1

CLAIM DATA

Schedule - Claims of the Hushamu Group, Nanaimo M.D., B.C.

Claim Name	# of units	record #
Hep 34	1	19012
Hep 35	1	19013
Hep 36	1	19014
Hep 37	1	19015
Hep 38	1	19016
Hep 39	1	19017
Hep 45	1	19023
Hep 49	1	19027
Hep 50	1	19028
Hep 51	1	19029
Hep 54	1	19032
Hep 55	1	19033
Hep 56	1	19034
Hep 58	1	19036
Hep 59	1	19037
Hep 62	1	19040
Hep 63	1	19041
Hep 64	1	19042
Hep 65	1	19043
Hep 66	1	19044
Hep 67	1	19045
Hep 68	1	19046
Hep 86	1	20216
Hep 87	1	20217
Hep 88	1	20218
Hep 89	1	20219
Hep 90	1	20220
Hep 91	1	20221
Hep 94	1	20224
Hep 95	1	20225
Hep 97	1	20227
Expo 102	1	20474
Expo 103	1	20475
Expo 104	1	20476
Expo 105	1	20477
Expo 106	1	20478
Expo 109	1	20232
Expo 111	1	20234
Expo 112	1	20235
Expo 113	1	20236
Expo 114	1	20237
Expo 115	1	20238
Expo 117	1	20240
Expo 118	1	20241
Expo 120	1	20243

Claim Name	# of units	record #
Expo 129	1	20252
Expo 130	1	20253
Expo 139	1	20262
Expo 141	1	20264
Expo 142	1	20265
Expo 144	1	20267
Expo 149	1	20272
Expo 150	1	20273
Expo 151	1	20274
Expo 152	1	20275
Expo 153	1	20276
Expo 154	1	20277
Expo 161	1	20284
Expo 162	1	20285
Expo 163	1	20286
Expo 164	1	20287
Expo 177	1	20288
Expo 178	1	20289
Expo 179	1	20290
Expo 181	1	20292
Expo 183	1	20294
Expo 184	1	20295
Expo 186	1	20297
Expo 188	1	20299
Expo 189	1	20300
Expo 191	1	20302
Expo 193	1	20304
Expo 194	1	20305
Expo 195	1	20306
Expo 196	1	20307
Expo 197	1	20308
Expo 198	1	20309
Expo 217	1	20328
Expo 218	1	20329
Expo 238	1	20492
Expo 317	1	20338
Expo 318	1	20339
Expo 319	1	21574
Expo 320	1	21575
Expo 321	1	21576
Expo 322	1	21577
Expo 503 Fr	1	20612
Expo 521	1	21613
Expo 522	1	21614
Expo 523	1	21615
Expo 524	1	21616
Expo 1008 Fr	1	27352

Don 1 Fr	1	29501
Don 2 Fr	1	29502
Don 3 Fr	1	29633
Don 7 Fr	1	29634
Don 8 Fr	1	29635

Total **97 units**

APPENDIX 2

SAMPLE DESCRIPTIONS

<u>Sample#</u>	<u>Location</u>	<u>Type</u>	<u>Description</u>	<u>Results</u>
33001	Pemberton Main	chip	xstal tuff w sk. hyd. alt.	327 Mn
33002	Pemberton Main	chip	xstal tuff w mo. hyd. alt.	19 ppb Au
33003	Pemberton Main	chip	heavily silic. w Fe stain	
33004	Pemberton Main same site as PED#6	chip	grey volcanoclastic 5-10% Py	
33005	Pemberton Main	chip	rhyolitic flow; breccia 2-3% Py	
33006	Pemberton Main	chip	rotten and.	
33007	Pemberton Main	chip	porph. and. breccia	
33008	Pemberton Main	chip	and. porph. wk. hydro alt.	449 Mn/12 ppb Au
33009	Pemberton Main	chip	and. w. wk. hydro alt.	713 Mn/18 ppb Au
33010	Pemberton Main	chip	bonanza volc. wk. hyd. alt.	1071 Mn.
33011	Pemberton Main	chip	bonanza volc. \leq 1% Py	363 Mn
33012	WN 900 Quarry	chip		233 Mn
33013	WN 900	chip	diorite w feld alt.	394 Mn
33014	WN Main "F"	chip	B. volc. rotten wk. hyd. alt.	388 Mn/10 ppb Au
33015	Cleskagh Ck. @ Rd.	float	siliceous, angular bldr. grey Py patch	
33016	Cleskagh Ck. @ Rd.	float	chalcedonic frag. w Fe stain	
33017	Cleskagh Ck. @ Rd.	float	propylitic bldr. slight kaolin	802 Mn
33018	Cleskagh Ck. @ Rd.	chip	Py breccia w qtz-carb vein	
33019	WN Main Quarry 300m W of WN 100	chip	B. volc w 1cm qtz. vein	1,246 Mn
33020	WN 101B	chip 5m	B. volc wk-mod hydro. alt.	446 Mn
33021	WN 101B	chip 3m	B. volc silic. wk pink alt. in veins	1,053 Mn
33022	Main Rd.	chip	B. volc we-mod alt.	578 Mn

<u>Sample#</u>	<u>Location</u>	<u>Type</u>	<u>Description</u>	<u>Results</u>
33023	Side Rd. off Holberg Connection	chip	B. volc breccia w wk-mod hydro alt.	442 Mn
33024	Side Rd. off Holberg Connection	chip	B. volc breccia w wk-mod hydro alt.	588 Mn
33025	Side Rd. off Holberg Connection	chip	B. volc w wk. hydro alt. 565 Sr/157V	1,262Mn/
33026	Side Rd. off Holberg Connection	chip	B. volc w hydro breccia 1m	342 Mn/104V
33027	Side Rd. off Holberg Connection	chip	B. volc w hydro breccia over 1m	550 Mn/126 Sr/ 119V
33028	Side Rd. off Holberg Connection	chip	B. volc w hydro breccia 3-4 m	816 Mn
33029	Side Rd. off Holberg Connection	chip	B. volc wk hydro alt.	352 Mn/120 Sr
33030	Side Rd. off Holberg Connection	chip	B. volc wk hydro alt	684 Mn/26 ppb Au
33031	Red Dog	chip 1.5 m	B. volc tuffaceous wk. hydro alt. w carb/zeol	898 Mn
33032	Red Dog	float	hydro breccia w carb/zeol	471 Mn
33033	Red Dog	chip 2m	B. volc tuffaceous wk hydro alt.	700 Mn
33034	Red Dog	chip 1m	B. volc tuff. hydro alt. fractures	1,044 Mn
33035	Red Dog (other claim)	chip	intrusive hydro alt.	856 Cu/320 ppb Au
33036	Red Dog (other claim)	chip	intrusive hydro alt.	290 Sr/12 ppb Au
33037	Red Dog (other claim)	chip	intrusive hydro alt.	149 Mn/41 ppb Au
34501	McIntosh Mtn. 20m N. 2440N, 2329E	grab 5m	heavily silicified	82 Mo/56 ppb Au
34502	McIntosh Mtn. NW of 34501	chip	heavily silicified	61 ppb Au
34503	McIntosh Mtn. 12m N. of 34554	grab	heavily silicified w 1/2" qtz. stringers	

<u>Sample#</u>	<u>Location</u>	<u>Type</u>	<u>Description</u>	<u>Results</u>
34504	McIntosh Mtn. 20m N. of 34503	grab	heavily silicified w 1/4" qtz. stringers	19 ppb Au
34505	McIntosh Mtn.	grab	heavily silicified w 1/4" qtz. stringers, minor limonite	70 Mo/113 ppb Au
34506	McIntosh Mtn.	grab	milky qtz. zone 1.5m	26 ppb Au
34507	H1000	grab 3m	massive py	17 ppb Au
34508	H1000	grab 3m	massive Py	429 Mn
34509	H1000	grab 2m	QFP w dissem. Py	475 Mn/11 ppb Au
34510	H1000	chip	QFP	
34511	H1000	grab	QFP w clay	
34512	H1000	grab 3m	QFP w dis. Py grey	
34513	H1000	grab	QFP w dis. Py grey	
34514	H1000	grab	QFP no Py	390 Mn
34515	H1020	grab	QFP w kaolin	
34516	H1020	grab	QFP(?) sil. kaolin Py	120 Mn
34517	H1020	grab		187 Mn
34518	H1020	grab 3m	QFP w Py grey	394 Mn
34519	H1020	grab 3m	QFP w Py grey	
34520	H1031	grab		156 Sr
34521	Quarry Between Pm 200 + PM 300	grab	QFP w Py clay alt.	162 Mn
34522	Pemberton 100		sinter cone kaolin andesite	
34523	Pemberton 100 pit 100m W of Sinter Cones		bedded Py	556 Mn/110V/ 16 ppb Au
34524	Pemberton 100 1100m W Sinter Cones	grab		797 Mn/147V

<u>Sample#</u>	<u>Location</u>	<u>Type</u>	<u>Description</u>	<u>Results</u>
34525	Pemberton 100	grab 3m	bedded py (across bedding)	460 Mn/122V
34526	Pemberton 100	grab 3m	bedded Py W of Sinter Cones (across bedding)	
34527	Pemberton 100	grab	andesite Bx. siliceous kaolin	
34528	Pemberton 100	grab	QFP	1,578 Mn
34529	Pemberton 100		andesite Bx. siliceous	
34530	Holberg Connection		qtz. vein	411 Mn/55 ppb Au
34531			fumarole Bx.	
34532	Rd. W of Hushamu Lake		B. volc dis. Py bleached	
34533	Rd. W of Hushamu Lake 25m E 34532		carb veinlets *limonite	139 Zn/1,060 Mn/ 9 ppb Au
34534	Rd. W of Hushamu Lake		bonanza carb veinlets	700 Mn/270 Sr
34535				15 ppb Au
34536	Rd. SSW of Hushamu Lake		volc grey siliceous Py	1,193 Mn
34537	Rd. SSW of Hushamu Lake			QFP w clay dis. Py
34538	Holberg Connection		Limestone (old sawmill site)	128 Mn/323 Sr
34539	Hushamu Main 250m S Jct. w 1400		hydro alt. kaolinized intrusive	350 Mn
34540	Hushamu Main 250m S Jct. w 1400		hydro alt. kaolinized intrusive	210 Mn
34541	H1500 Jct. H1540		B. volc w carb stringers	463 Mn/117V
34542	WN Main	grab	B. volc kaolinized	373 Mn
34543	WN Main	grab	B. volc Py	

<u>Sample#</u>	<u>Location</u>	<u>Type</u>	<u>Description</u>	<u>Results</u>
34544	WN 1200 Pit SW of Rd.	grab	B. volc Py	
34545	Hushamu Rd. 2K N of Dump		rusty conglomerate	415 Mn/104 Sr
34546 E of Rd.	Red Dog EC-1 Rd.		B. volc sil. bleached Py vein Au	1,509 Cu/290 ppb Au
34547	Red Dog EC-1 Rd.	grab 2.5m	B. volc limonite	1,374 Cu/134V/ 45 ppb Au
34548	Red Dog EC-1 Rd.	grab	surficial conglomerate	595 Cu/128V/ 13 ppb Au
34549	Red Dog EC-1 Rd. 20m S 34548	grab 1.5m	B. volc hydro alt.	581 Cu/440 ppb Au
34550	Red Dog EC-1 Rd.		surficial conglomerate	442 Mn/22 ppb Au
34551	McIntosh Mtn.	chip	heavily siliceous	19 ppb Au
34552	McIntosh Mtn.	chip	siliceous sinter	148 Mo/479 Cu/ 352 As/109 Au
34553	McIntosh Mtn.	chip	stockwork siliceous w qtz. veins ≤ 1/4"	85 ppb Au
34554	McIntosh Mtn.	chip	stockwork siliceous w qtz. veins ≤ 1/4"	
34555	McIntosh Mtn.	grab	wk. sinter w qtz. stringers	46 ppb Au
34556	McIntosh Mtn. SW DDH 154	chip 0.5m	heavily silicified w qtz. stringers ≤ 1 cm	1,205 ppb Au
41501	Red Dog Resample 34546	grab 2.5m	sil. Py vein	63 ppb Au
41502	Red Dog Resample 34547	grab	float selected from sil. Zn east wall of quarry	192 ppb Au
41503	Rd. W End of Nahwitti Lake	grab	float sil. skarn Py	13 ppb Au/206 Mn

<u>Sample#</u>	<u>Location</u>	<u>Type</u>	<u>Description</u>	<u>Results</u>
41504	Rd. W End of Nahwitti Lake	grab	float sil. skarn Py	575 Cu/101 As/ 65 ppb Au
41505	Rd. W End of Nahwitti Lake	grab	float sil. volc w ribbon qtz.	609 Mo/622 As/ 88 ppb Au
41506	Rd. W End of Nahwitti Lake 200m S	grab	float black bedded limestone w w calcsil. + Py	316 Mn/143 Sr
41507	Rd. W End of Nahwitti Lake 200m S	grab	float black bedded Lst without calcsil + Py	221 Mn/258 Sr
41508	Red Dog Pit L1 #1	chip		232 Cu/137 Mn/ 108V/10 ppb Au
41509	Red Dog Pit L1 #2	chip		386 Mo/666 Cu/ 75 ppb Au
41510	Red Dog Pit L1 #3	chip		166 Mo/1,947 Cu/ 28 ppb Au
41511	Red Dog Pit L2 #1	chip		364 Cu/459 Mn/ 17 ppb Au
41512	Red Dog Pit L2 #2	chip		367 Cu/352 Mn/ 19 ppb Au
41513	Red Dog Pit L2 #3	chip		140 Mo/32 ppb Au
41514	Red Dog Pit L2 #4	chip		121 Mo/141 ppb Au
41515	Red Dog Pit L3 #1	chip		83 Mo/43 ppb Au
41516	Red Dog Pit L3 #2	chip		207 Mo/200 ppb Au
41517	Red Dog Pit L3 #3	chip		175 Mo/55 ppb Au
41518	Red Dog Pit L4 #1	chip		471 Cu/256 Mn/ 36 ppb Au
41519	Red Dog Pit L4 #2	chip		164 Mo/4,811 Cu/ 550 ppb Au
41520	Red Dog Pit L4 #3	chip		89 Mo/1,900 Cu/ 290 ppb Au

<u>Sample#</u>	<u>Location</u>	<u>Type</u>	<u>Description</u>	<u>Results</u>
41521	Red Dog Pit L4 #4	chip		92 Mo/7,177 Cu/ 1,060 ppb Au
41522	Red Dog Pit L4 #5	chip		192 Mo/1,003 Cu/ 1,140 ppb Au
41523	Red Dog Pit L5 #1	chip		63 Mo/3,150 Cu/ 820 ppb Au
41524	Red Dog Pit L5 #2	chip		407 Cu/146 ppb Au
45125	Red Dog Pit L6 #1	chip		211 Mo/1,862 Cu/ 230 ppb Au
41526	Red Dog Pit L7 #1	chip		68 Mo/110 Cu/ 31 ppb Au
41527	Red Dog Pit L8 #1	chip		3,399 Cu/229 Mn/ 29 ppb Au
41528	Holberg Rd. Quarry 200m up Rd. to LCP WIN 1		epidote Py calc. skarn contact zone granite/Parson Bay	996 Zn/1,330 Mn/ 53 ppb Au/ 22.3 Ag
41529	Holberg Rd. Quarry 200m up Rd. to LCP WIN 1		chloritized carb. veins	422 Mn/135 Sr
41530	Rd. to LCP WIN 1		Parson Bay tuff w limonite siliceous next to granite contacts	816 Mn
41531	Holberg Rd.		B. volc hydro alt. limonite siliceous	40 ppb Au
41532	Holberg Rd.		B. volc hydro alt. clay gouge portion	315 Zn/221 Mn/ 330 ppb Au

APPENDIX 3

ASSAY CERTIFICATES

Daiwan Engineering Ltd.

(604) 688-1508, 1030-609 Granville Street, Vancouver, B.C. (604) 688-1508

ACME ANALYTICAL LABORATORIES LTD. DATE RECEIVED: DEC 12 1988
852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
PHONE(604)253-3158 FAX(604)253-1716 DATE REPORT MAILED: Dec. 13/88

ASSAY CERTIFICATE

- SAMPLE TYPE: Pulp AU** BY FIRE ASSAY FROM 1/2 A.T.

SIGNED BY *C. Long* D. TOYE, C. LRONG, B. CHAN, J. WANG; CERTIFIED B.C. ASSAYERS

DAIWAN ENGINEERING PROJECT EXPO FILE # 88-5978R

SAMPLE#	AU** oz/t
34505	.003
34552	.004
34556	.035

ACME ANALYTICAL LABORATORIES LTD. DATE RECEIVED: DEC 9 1988
852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
PHONE(604)253-3158 FAX(604)253-1716 DATE REPORT MAILED: Dec. 13/88.

ASSAY CERTIFICATE

- SAMPLE TYPE: ROCK AU** AND AG** BY FIRE ASSAY FROM 1 A.T.

SIGNED BY *C. Long*. D.TOYE, C.LEONG, B.CHAN, J.WANG; CERTIFIED B.C. ASSAYERS

DAIWAN ENGINEERING LTD. PROJECT HARRINGTON FILE # 88-6214

SAMPLE#	Ag** OZ/T	Au** OZ/T
A	.01	.001
B	.01	.009
C	.08	.017
D	.01	.001

GEOCHEMICAL ANALYSIS CERTIFICATE

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER.
 THIS LEACH IS PARTIAL FOR NA FE SR CA P LA CR NG BA YI B W AND LIMITED FOR NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM.
 - SAMPLE TYPE: Rock Chips AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

DATE RECEIVED: DEC 21 1988 DATE REPORT MAILED: Dec 23/88 SIGNED BY: *C. Long* D. TOYE, C. LEONG, B. CHAN, J. WANG; CERTIFIED B.C. ASSAYERS

DAIWAN ENGINEERING LTD. PROJECT EXPO File # 88-6355 Page 1

SAMPLE#	Mo PPM	Cu PPM	Pb PPM	Zn PPM	Ag PPM	Ni PPM	Co PPM	Mn PPM	Fe %	As PPM	U PPM	Au PPM	Tb PPM	Sr PPM	Cd PPM	Sb PPM	Bi PPM	V PPM	Ca %	P %	La PPM	Cr PPM	Hg %	Ba PPM	Ti %	B PPM	Al %	Na %	K %	V PPM	AU* PPB
C 33023	1	30	4	36	.1	3	11	442	4.59	2	5	ND	1	46	1	2	3	66	.32	.063	8	3	1.06	219	.01	2	2.70	.05	.06	1	1
C 33024	1	40	6	41	.1	7	10	500	3.54	16	5	ND	1	51	1	2	2	97	2.27	.045	5	11	.04	31	.11	2	4.35	.02	.00	1	2
C 33025	1	66	5	76	.1	2	11	1262	10.63	11	5	ND	1	565	2	2	2	157	1.28	.100	5	2	.95	78	.29	7	3.26	.07	.09	1	1
C 33026	1	32	17	39	.1	2	9	342	5.03	19	5	ND	1	92	2	2	2	104	4.34	.049	4	8	.64	14	.13	3	6.74	.02	.09	1	1
C 33027	1	27	15	43	.1	6	10	550	3.11	22	5	ND	1	126	1	2	3	119	4.95	.044	8	12	.77	26	.09	5	7.00	.02	.10	3	1
C 33028	1	27	5	58	.1	2	13	816	4.05	25	5	ND	1	72	1	2	2	92	2.04	.046	8	9	1.07	11	.10	5	4.61	.02	.05	1	1
C 33029	1	13	9	25	.1	3	5	352	1.61	3	5	ND	1	120	1	4	2	36	5.78	.038	5	4	.23	34	.05	4	7.05	.02	.12	1	1
C 33030	1	288	9	51	.5	4	13	684	3.59	30	5	ND	1	54	1	2	2	34	3.30	.050	5	8	1.19	8	.09	3	3.43	.01	.07	1	26
C 33031	1	28	7	58	.1	3	10	898	3.74	3	5	ND	1	33	1	2	2	79	1.40	.050	3	7	.99	72	.03	5	3.32	.06	.11	1	1
C 33032	1	10	38	38	.1	3	7	471	2.15	6	5	ND	1	74	1	4	3	41	3.90	.030	4	6	.50	45	.01	4	5.43	.03	.11	1	2
C 33033	1	42	29	62	.1	10	10	700	3.41	4	5	ND	1	96	1	2	2	53	2.43	.065	3	11	1.11	155	.10	3	4.77	.02	.12	1	1
C 33034	1	29	15	80	.1	7	13	1044	4.47	23	5	ND	1	59	2	2	2	87	1.72	.045	3	6	1.11	48	.05	3	4.49	.04	.09	1	1
C 33035	72	856	13	37	.3	5	5	58	9.90	3	5	ND	2	3	1	2	2	65	.03	.047	2	6	.43	27	.02	5	1.36	.01	.19	1	320
C 33036	5	403	5	23	.1	5	12	66	2.91	2	5	ND	1	290	1	4	2	61	.82	.089	2	3	.95	115	.02	2	2.37	.09	.07	1	12
C 33037	5	254	8	33	.3	8	11	149	4.66	2	5	ND	1	92	1	2	2	51	3.21	.058	3	8	.97	13	.05	2	5.51	.01	.11	1	41
B 41503	10	83	6	23	.2	26	26	206	5.55	41	5	ND	1	7	1	3	3	18	.10	.018	2	14	.49	15	.04	2	.71	.01	.01	1	73
B 41504	27	575	17	39	.1	13	17	77	13.74	101	5	ND	2	7	1	3	3	38	.04	.010	2	4	.10	9	.01	4	.54	.01	.01	2	65
B 41505	609	88	53	7	.1	6	1	16	2.76	622	5	ND	1	3	1	18	2	15	.03	.005	2	6	.01	2	.01	3	.15	.01	.01	1	88
B 41506	1	15	6	11	.3	3	3	316	1.68	17	6	ND	1	143	1	2	2	2	39.71	.050	2	1	.03	1	.01	2	.08	.01	.01	1	1
B 41507	1	8	2	3	.3	3	3	221	.53	9	7	ND	1	258	1	2	2	1	42.15	.012	2	2	.06	27	.01	2	.23	.02	.01	1	1
B 41508	5	232	11	29	.1	2	5	137	5.16	3	5	ND	17	21	1	2	2	108	.52	.033	4	7	.67	189	.16	2	2.49	.03	.10	1	10
B 41509	386	666	4	14	.1	2	8	15	14.24	12	5	ND	6	3	1	2	2	51	.01	.051	2	1	.01	540	.10	2	.31	.01	.09	1	75
B 41510	166	1947	13	30	.2	7	13	4	35.51	35	5	ND	5	5	2	2	2	68	.41	.059	2	1	.01	18	.01	2	.43	.01	.09	3	28
B 41511	5	364	11	72	.1	6	11	459	3.77	8	5	ND	4	37	1	2	2	69	.42	.038	6	5	1.58	43	.11	5	2.42	.03	.13	1	17
B 41512	3	367	13	49	.1	8	9	352	3.49	6	5	ND	5	11	1	2	2	78	.20	.054	7	6	1.71	39	.11	3	2.70	.03	.12	1	19
B 41513	140	130	2	3	.2	4	1	18	1.77	19	5	ND	1	2	1	2	2	11	.01	.015	2	5	.02	7	.01	2	.11	.01	.04	1	32
B 41514	121	238	2	3	.2	6	1	10	3.28	17	5	ND	2	7	1	2	2	16	.01	.021	2	4	.01	23	.01	2	.34	.02	.09	1	141
B 41515	83	194	2	14	.2	7	7	92	1.84	2	5	ND	1	1	1	2	2	10	.01	.005	2	6	.42	13	.01	4	.81	.01	.08	1	43
B 41516	207	218	4	1	.2	4	1	12	1.76	2	5	ND	2	2	1	2	2	7	.01	.004	3	4	.01	11	.01	2	.20	.01	.07	1	200
B 41517	175	45	2	1	.1	4	1	12	.75	3	5	ND	5	1	1	2	2	3	.01	.013	6	6	.01	4	.01	2	.19	.01	.06	1	55
B 41518	8	471	7	34	.2	8	23	256	5.15	11	6	ND	5	14	1	2	2	78	.28	.037	5	6	1.63	29	.11	4	2.26	.05	.15	1	36
B 41519	140	4911	5	11	.5	6	24	25	3.37	4	7	ND	4	2	1	2	2	19	.02	.024	5	2	.05	20	.01	3	.67	.01	.10	1	550
B 41520	89	1900	2	23	.3	6	11	51	5.45	2	5	ND	4	2	1	2	3	38	.02	.025	6	2	.12	27	.02	2	.83	.01	.15	1	290
B 41521	92	7177	5	34	.6	7	15	90	7.51	5	5	2	4	1	1	3	20	20	.01	.012	2	4	.24	11	.01	5	1.07	.01	.07	1	1060
B 41522	192	1003	2	20	.7	5	8	49	5.85	4	6	2	4	2	1	2	2	20	.01	.008	6	2	.13	16	.01	2	.65	.01	.12	1	1140
B 41523	63	3150	3	40	.4	8	16	76	8.13	2	5	ND	3	2	1	2	2	19	.01	.005	2	2	.10	16	.01	3	.75	.01	.12	1	820
STD C/AN-R	18	59	43	132	7.1	66	30	1048	4.25	43	21	8	40	49	20	16	25	61	.49	.098	41	55	.89	180	.07	38	1.95	.06	.14	11	480

SAMPLE#	Mo PPM	Cu PPM	Pb PPM	Zn PPM	Ag PPM	Ni PPM	Co PPM	Mn PPM	Fe %	As PPM	U PPM	Au PPM	Tb PPM	Sr PPM	Cd PPM	Sb PPM	Bi PPM	V PPM	Ca %	P %	La PPM	Cr PPM	Mg %	Ba PPM	Ti %	B PPM	Al %	Na %	K %	V PPM	Au* PPB
B 41524	50	407	2	5	.1	4	9	17	3.43	2	5	ND	1	1	1	2	4	4	.01	.001	2	5	.01	4	.01	2	.15	.01	.08	1	186
B 41525	211	1062	2	2	.1	2	5	27	2.51	2	5	ND	1	2	1	2	4	6	.01	.004	2	5	.07	7	.01	2	.34	.01	.09	1	230
B 41526	68	110	2	2	.1	4	2	19	1.23	3	5	ND	1	2	1	2	2	3	.01	.005	2	9	.03	5	.01	4	.17	.01	.05	1	31
B 41527	32	3399	11	43	.1	51	14	229	3.98	12	5	ND	1	35	1	2	3	79	.49	.043	4	48	1.97	21	.04	2	3.72	.08	.02	1	29
B 41528	2	237	42	996	22.3	3	58	1330	2.98	28	5	ND	1	73	8	2	23	41	4.10	.053	2	4	.95	1	.07	3	1.22	.01	.01	1	53
B 41529	3	144	9	43	.5	49	21	422	2.80	16	5	ND	1	135	1	2	2	57	4.06	.052	3	85	1.33	48	.09	2	5.26	.04	.05	3	5
B 41530	2	27	3	105	1.0	5	5	816	.87	3	5	ND	2	50	1	2	2	16	12.98	.029	4	9	.42	17	.04	2	.85	.02	.02	1	3
B 41531	3	39	17	265	1.2	1	8	38	10.34	11	8	ND	2	7	3	2	4	51	.11	.196	2	6	.04	35	.15	2	.59	.01	.17	1	40
B 41532	2	68	18	315	6.0	14	32	221	8.77	22	5	ND	1	4	3	2	3	32	.15	.041	2	6	.31	20	.18	2	.80	.01	.14	1	330
STD C/AU-R	19	63	40	134	7.4	72	31	1059	3.99	43	21	8	39	51	19	18	24	60	.51	.092	40	59	.94	179	.07	38	1.95	.06	.14	13	470

3-6

GEOCHEMICAL ANALYSIS CERTIFICATE

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER.
 THIS LEACH IS PARTIAL FOR NH FE SR CA P LA CR NG BA TI B W AND LIMITED FOR NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM.
 - SAMPLE TYPE: ROCK AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

DATE RECEIVED: DEC 2 1988 DATE REPORT MAILED: Dec 7/88 SIGNED BY: *C. Long* D. TOYE, C. LEONG, B. CHAN, J. WANG; CERTIFIED B.C. ASSAYERS

DAIWAN ENGINEERING LTD. PROJECT MORAGA RES.-EXPO File # 88-6115

SAMPLE#	Mo	Cu	Pb	Zn	Ag	W1	Co	Mn	Fe	As	U	Au	Th	Sr	Cd	Sb	Bi	V	Ca	P	La	Cr	Hg	Ba	Ti	B	Al	Na	K	W	Au*
	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	%	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	%	%	PPM	PPM	%	PPM	%	PPM	%	%	%	PPM	PPB
C 34515	3	24	12	6	.1	5	3	5	3.47	12	5	ND	1	23	1	2	2	20	.02	.017	2	8	.02	73	.01	2	.39	.01	.01	1	3
C 34516	1	29	6	13	.1	4	6	120	7.08	13	5	ND	1	14	1	2	2	74	.01	.138	3	7	.39	126	.01	2	1.95	.01	.07	1	1
C 34517	1	66	13	28	.1	11	22	187	7.44	9	5	ND	1	5	1	2	2	46	.03	.061	4	9	.59	13	.01	2	1.56	.01	.04	1	1
C 34518	2	84	15	93	.1	11	19	394	5.23	13	5	ND	1	18	1	2	2	23	.68	.050	7	4	.51	30	.01	3	1.24	.01	.06	1	1
C 34519	4	115	17	33	.1	13	19	38	6.27	13	5	ND	2	2	1	2	2	18	.02	.004	2	6	.10	16	.01	2	.70	.01	.02	1	1
C 34520	2	22	6	5	.1	1	3	71	3.01	19	5	ND	1	156	1	2	2	19	.02	.052	21	2	.19	133	.01	2	1.19	.01	.11	1	2
C 34521	1	53	8	25	.1	27	19	162	6.07	8	5	ND	1	5	1	2	2	18	.05	.031	8	12	.28	21	.01	2	.98	.01	.09	2	1
C 34522	1	13	4	3	.1	1	2	18	.29	2	5	ND	1	25	1	2	2	3	.02	.004	2	2	.02	420	.01	2	.47	.01	.01	1	1
C 34523	1	47	7	57	.1	7	9	556	6.22	8	5	ND	1	20	1	2	2	110	.24	.052	11	19	1.05	61	.05	2	2.45	.04	.05	1	16
C 34524	1	69	14	77	.1	13	13	797	6.15	9	5	ND	1	27	1	2	2	147	.30	.056	12	25	1.72	101	.02	2	3.89	.05	.04	1	1
C 34525	5	53	7	53	.1	16	17	460	5.99	4	5	ND	1	30	1	2	2	122	.46	.040	8	20	1.16	24	.08	3	1.97	.05	.07	1	1
C 34526	5	41	9	9	.1	8	7	13	2.80	77	5	ND	1	5	1	2	2	3	.01	.001	2	4	.01	41	.01	2	.32	.01	.01	2	1
C 34527	3	10	3	2	.1	2	1	12	1.42	23	5	ND	1	11	1	2	2	13	.03	.004	2	3	.02	173	.01	2	.32	.01	.01	1	1
C 34528	1	54	10	73	.1	4	15	1578	4.79	10	5	ND	1	38	1	2	2	61	6.47	.050	13	9	.50	60	.01	2	2.09	.01	.08	1	4
C 34529	6	8	2	1	.1	11	2	28	.83	12	5	ND	1	4	1	7	3	1	.04	.002	2	16	.01	14	.01	3	.02	.01	.01	2	1
C 34530	1	30	60	27	.7	3	1	411	.22	2	5	ND	1	3	1	2	2	2	.12	.020	2	20	.06	1	.01	2	.21	.01	.01	2	55
C 34531	4	156	14	19	.1	1	9	2	17.43	5	5	ND	1	7	1	2	2	91	.01	.061	2	3	.01	270	.01	2	.32	.01	.01	1	1
C 34532	1	70	26	109	.2	5	19	1012	6.09	26	5	ND	1	26	1	2	2	66	.63	.060	4	6	2.14	29	.08	2	2.18	.01	.12	1	2
C 34533	1	116	26	139	.2	12	15	1060	6.10	20	5	ND	1	55	1	2	2	70	1.05	.056	2	24	2.31	45	.14	2	3.46	.01	.05	1	9
C 34534	1	24	8	55	.1	3	13	700	4.85	5	5	ND	1	270	1	2	2	61	.78	.070	2	5	2.04	62	.06	2	2.93	.03	.09	1	1
C 34535	2	55	22	9	.1	3	8	14	5.33	6	5	ND	1	18	1	2	2	10	.02	.005	2	2	.03	26	.01	2	.20	.01	.01	1	15
C 34536	1	38	9	88	.1	5	11	1193	4.33	5	5	ND	1	43	1	2	2	78	.40	.073	4	8	1.96	56	.14	3	2.69	.05	.02	1	1
C 34537	4	26	8	5	.1	26	21	6	5.80	2	5	ND	2	16	1	2	2	6	.01	.003	2	3	.02	20	.01	2	.86	.01	.04	1	1
C 34538	1	5	2	3	.1	1	1	128	.10	5	6	ND	1	323	1	2	2	3	37.54	.008	2	1	.05	8	.01	3	.14	.01	.01	2	1
C 34539	1	29	6	40	.1	7	8	350	2.89	2	5	ND	1	58	1	2	2	54	1.66	.034	5	11	.61	11	.06	2	2.75	.01	.05	1	1
C 34540	1	16	8	24	.1	1	2	210	1.94	8	5	ND	2	40	1	2	2	22	1.04	.022	3	3	.37	13	.05	2	1.94	.01	.03	1	2
C 34541	1	49	8	174	.1	23	32	463	5.37	6	5	ND	1	160	2	2	2	117	3.30	.071	4	18	.71	14	.07	3	5.36	.07	.07	1	1
STD C/AU-R	17	60	38	132	6.6	67	31	1028	4.18	40	21	7	37	47	18	18	21	58	.48	.092	39	57	.91	173	.06	34	1.98	.06	.13	11	510

GEOCHEMICAL ANALYSIS CERTIFICATE

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER.
 THIS LEACH IS PARTIAL FOR NH FE SR CA P LA CR NG BA TI B W AND LIMITED FOR NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM.
 - SAMPLE TYPE: ROCK AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

DATE RECEIVED: NOV 23 1988 DATE REPORT MAILED: Nov 29/88 SIGNED BY: *C. Long* D. TOYE, C. LEONG, B. CHAN, J. WANG; CERTIFIED B.C. ASSAYERS

DAIWAN ENGINEERING LTD. PROJECT EXPO File # 88-5978

SAMPLE#	Mo PPM	Cu PPM	Pb PPM	Zn PPM	Ag PPM	Ni PPM	Co PPM	Mn PPM	Fe %	As PPM	U PPM	Au PPM	Tb PPM	Str PPM	Cd PPM	Sb PPM	Bi PPM	V PPM	Ca %	P %	La PPM	Cr PPM	Hg %	Ba PPM	Ti %	B PPM	Al %	Na %	K %	W PPM	Au* PPB
C 34501	82	61	8	5	.1	1	1	17	1.75	87	5	ND	1	7	1	2	2	2	.01	.007	2	7	.01	1	.01	2	.04	.01	.01	3	56
C 34502	17	18	8	5	.1	8	1	18	.58	2	5	ND	1	9	1	2	2	2	.01	.008	2	7	.01	1	.01	3	.04	.01	.01	1	61
C 34503	25	36	3	5	.1	3	1	33	.81	41	5	ND	1	3	1	2	4	3	.01	.003	2	8	.01	1	.01	3	.03	.01	.01	4	2
C 34504	18	20	5	4	.1	4	1	18	.44	17	5	ND	1	5	1	2	2	1	.01	.003	2	7	.01	1	.01	2	.02	.01	.01	1	19
C 34505	70	88	6	4	.1	1	1	11	1.99	198	5	ND	1	5	1	3	2	15	.01	.003	2	8	.01	1	.01	3	.11	.01	.01	2	113
C 34506	28	12	2	1	.1	1	1	36	.18	3	5	ND	1	1	1	2	2	1	.01	.001	2	2	.01	1	.01	2	.10	.01	.01	1	26
C 34507	4	81	12	5	.1	8	16	5	6.23	2	5	ND	1	11	1	2	2	12	.01	.003	2	2	.01	10	.01	6	.33	.01	.01	1	17
C 34508	2	45	21	30	.1	1	12	429	5.88	10	5	ND	2	3	1	2	2	22	.01	.086	5	1	.90	22	.01	7	2.06	.01	.10	1	2
C 34509	1	74	6	44	.1	8	14	475	6.38	7	5	ND	2	28	1	2	2	57	.97	.061	6	10	.92	18	.21	7	2.40	.02	.11	1	11
C 34510	4	68	2	49	.1	5	13	96	4.35	6	5	ND	2	13	1	2	2	14	.17	.064	11	3	.32	26	.01	5	1.47	.01	.10	1	1
C 34511	1	51	13	17	.1	1	6	77	4.54	4	5	ND	1	33	1	2	2	12	.54	.047	8	1	.14	72	.01	3	1.07	.01	.09	1	5
C 34512	1	91	3	56	.1	10	24	79	4.76	4	5	ND	2	22	1	2	2	18	.40	.051	16	1	.59	16	.01	3	1.71	.01	.09	1	1
C 34513	4	82	10	48	.1	7	16	61	5.18	19	5	ND	2	10	1	2	2	25	.35	.064	18	2	.48	10	.01	4	1.77	.01	.10	1	2
C 34514	16	86	4	36	.1	8	22	390	6.69	3	5	ND	3	2	1	2	2	25	.02	.086	4	4	.71	8	.01	2	1.70	.01	.08	1	1
C 34551	38	29	5	2	.2	1	1	12	.86	3	5	ND	2	10	1	2	2	3	.01	.006	2	2	.01	9	.01	2	.09	.01	.01	1	19
C 34552	148	479	5	16	.1	1	9	3	16.52	352	5	ND	4	2	1	2	2	49	.01	.011	2	4	.01	4	.01	3	.29	.01	.01	1	109
C 34553	90	26	5	4	.1	1	2	6	3.13	2	5	ND	3	3	1	2	2	9	.01	.005	2	2	.01	1	.01	2	.19	.01	.02	1	85
C 34554	10	51	2	4	.1	1	1	4	2.28	31	5	ND	3	5	1	2	2	16	.01	.003	2	2	.01	2	.01	2	.21	.01	.01	1	7
C 34555	5	14	2	1	.3	1	1	2	.22	2	5	ND	3	7	1	2	2	5	.01	.002	2	1	.01	1	.01	2	.20	.01	.01	1	46
C 34556	205	435	24	9	.1	2	5	7	7.56	529	5	ND	4	6	1	2	3	31	.01	.007	2	6	.01	1	.01	8	.18	.01	.01	2	1205
STD C/AU-R	19	62	40	132	6.7	70	31	1048	4.05	42	23	8	41	49	19	16	20	61	.47	.095	42	57	.93	176	.07	41	1.99	.06	.16	13	520

GEOCHEMICAL ANALYSIS CERTIFICATE

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO₃-H₂O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER.
 THIS LEACH IS PARTIAL FOR MN FE SR CA P LA CR MG BA TI B W AND LIMITED FOR NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM.
 - SAMPLE TYPE: ROCK AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

DATE RECEIVED: DEC 11 1988 DATE REPORT MAILED: Dec 15/88 SIGNED BY: *C. Long* D. TOYE, C. LEONG, B. CHAN, J. WANG; CERTIFIED B.C. ASSAYERS

DAIWAN ENGINEERING LTD. PROJECT EXPO File # 88-6230

SAMPLE#	Mo	Cu	Pb	Zn	Ag	Ni	Co	Mn	Fe	As	U	Au	Th	Sr	Cd	Sb	Bi	V	Ca	P	La	Cr	Mg	Ba	Ti	B	Al	Na	K	W	Au*
	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	%	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	PPM	%	%	PPM	PPM	%	PPM	%	PPM	%	%	%	PPM	PPM
C 33001	1	133	15	92	.1	13	14	327	5.44	2	5	ND	1	63	1	2	2	166	.69	.057	10	18	1.88	104	.25	2	4.22	.07	.03	1	2
C 33002	3	52	11	7	.1	2	4	59	2.69	4	5	ND	2	11	1	2	2	21	.04	.032	11	4	.08	118	.06	5	1.04	.01	.15	1	19
C 33003	22	112	2	11	.1	1	4	28	6.49	98	5	ND	1	7	1	2	6	10	.02	.003	2	7	.04	445	.01	2	.17	.01	.01	9	8
C 33004	4	94	26	5	.2	1	4	9	4.69	14	5	ND	1	18	1	2	18	11	.14	.004	2	2	.01	10	.01	2	.29	.01	.04	1	2
C 33005	6	42	17	7	.5	3	11	9	5.00	10	5	ND	1	2	1	2	2	10	.01	.001	2	5	.01	12	.01	2	.29	.01	.01	1	1
C 33006	21	59	9	12	.1	4	6	12	8.87	13	5	ND	1	10	1	2	2	87	.01	.021	2	9	.04	92	.01	2	.30	.01	.01	2	1
C 33007	3	72	27	7	.1	3	8	2	8.00	25	5	ND	2	37	1	4	6	83	.01	.089	2	7	.01	185	.01	2	.56	.01	.05	1	3
C 33008	2	77	5	51	.1	8	8	449	3.74	11	5	ND	1	69	1	2	2	60	1.14	.036	3	10	1.21	68	.05	2	3.03	.03	.11	1	12
C 33009	1	59	36	114	.5	14	10	713	6.15	24	5	ND	1	86	1	3	2	65	1.74	.067	2	23	2.36	24	.21	3	5.21	.03	.06	5	18
C 33010	1	80	25	92	.3	94	44	1071	4.84	12	5	ND	1	69	2	2	2	98	.36	.351	4	65	2.36	104	.25	2	5.42	.04	.02	1	1
C 33011	3	27	13	30	.3	7	10	363	2.85	8	5	ND	1	52	1	3	2	30	1.44	.029	3	5	.79	29	.05	2	2.99	.01	.09	2	2
C 33012	1	18	4	21	.1	4	5	233	1.85	3	5	ND	6	33	1	2	2	31	.90	.025	6	4	.42	25	.05	2	1.44	.02	.07	1	1
C 33013	9	19	115	305	.9	1	7	394	1.91	19	5	ND	1	39	2	2	5	11	1.84	.042	6	3	.29	67	.05	2	2.01	.01	.13	1	6
C 33014	24	26	8	47	.6	8	14	388	4.28	46	5	ND	1	10	1	2	2	38	.39	.076	6	4	.38	26	.04	3	.93	.02	.11	1	10
C 33015	3	31	13	13	.3	3	3	36	2.64	3	5	ND	1	5	1	2	2	4	.05	.002	2	4	.03	192	.01	3	.36	.01	.01	1	1
C 33016	6	12	15	4	.1	3	1	20	.65	2	5	ND	1	4	1	2	2	3	.02	.005	2	5	.01	66	.01	2	.25	.01	.01	1	1
C 33017	1	30	12	51	.1	5	11	802	4.08	66	5	ND	2	92	1	2	2	72	4.75	.072	7	4	1.68	28	.11	6	4.21	.03	.08	2	5
C 33018	1	9	6	1	.1	1	1	24	.21	3	5	ND	1	12	1	2	2	3	.11	.002	2	2	.03	8	.01	3	.64	.01	.01	1	1
C 33019	2	47	10	60	.3	9	14	1246	4.50	3	5	ND	1	71	1	2	2	93	3.54	.055	7	4	.97	61	.18	2	2.59	.05	.06	1	6
C 33020	1	28	6	29	.1	1	8	446	7.81	7	5	ND	2	4	1	2	2	98	.10	.064	2	2	.55	44	.28	2	1.44	.02	.10	1	1
C 33021	1	40	11	67	.1	5	25	1053	5.84	2	5	ND	1	90	1	2	2	201	1.84	.071	11	7	1.80	107	.24	4	4.04	.10	.04	1	1
C 33022	3	51	10	68	.1	15	17	578	5.67	7	5	ND	2	66	1	3	2	99	3.49	.061	10	22	2.58	10	.01	2	3.44	.02	.05	1	1
C 34542	1	41	11	54	.1	39	12	373	6.17	18	5	ND	2	4	1	2	2	96	.06	.017	4	67	.81	72	.01	2	2.14	.01	.09	1	2
C 34543	8	34	2	7	.2	12	11	15	7.14	13	5	ND	1	2	1	2	2	32	.01	.001	2	23	.01	25	.01	2	.59	.01	.01	1	4
C 34544	3	42	9	47	.1	5	10	634	4.85	2	5	ND	1	5	1	2	2	43	.28	.078	8	12	1.01	56	.16	2	1.51	.01	.12	1	2
C 34545	2	60	10	40	.1	19	13	415	5.38	6	5	ND	2	24	1	2	2	104	.38	.055	7	39	.94	127	.19	4	2.38	.02	.06	1	1
C 34546	80	1509	2	11	.1	6	17	26	4.43	7	5	ND	1	1	1	2	2	9	.01	.001	2	8	.04	10	.01	3	.50	.01	.05	1	290
C 34547	44	1374	8	23	.1	2	11	7	26.54	17	5	ND	7	1	3	2	2	134	.01	.071	2	6	.01	50	.02	2	.47	.01	.06	1	45
C 34548	8	595	9	34	.1	3	10	157	19.88	31	5	ND	4	11	2	2	2	128	.05	.097	2	12	.31	41	.18	2	1.25	.02	.05	2	13
C 34549	117	581	2	4	.3	5	5	17	3.27	8	5	ND	1	6	1	2	2	12	.01	.006	2	4	.02	58	.01	3	.32	.01	.05	1	440
C 34550	15	157	6	40	.1	4	11	442	6.45	4	5	ND	1	65	2	2	2	94	.23	.035	4	9	.88	152	.16	2	2.46	.03	.04	1	22
B 41501	164	127	13	3	.1	3	3	23	2.28	8	5	ND	9	47	1	2	2	8	.01	.029	4	7	.89	43	.01	5	.46	.01	.10	1	63
B 41502	78	403	2	4	.1	5	2	18	1.31	2	5	ND	1	2	1	2	2	9	.01	.004	2	36	.02	8	.01	5	.15	.01	.04	1	192
87D C/AU-R	19	63	37	132	6.7	70	31	1040	4.14	42	18	8	38	48	19	16	22	61	.50	.094	40	55	.98	175	.07	32	2.85	.06	.13	12	910

ACME ANALYTICAL LABORATORIES LTD.
852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
PHONE(604)253-3158 FAX(604)253-1716

DATE RECEIVED: JAN 16 1989

DATE REPORT MAILED: *Jan. 19. 89.*

GEOCHEMICAL ANALYSIS CERTIFICATE

- SAMPLE TYPE: ROCK
AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

SIGNED BY. *C. Long*. D. TOYE, C. LEONG, B. CHAN, J. WANG; CERTIFIED B.C. ASSAYERS

DAIWAN ENGINEERING LTD. PROJECT EXPO FILE # 89-0098

SAMPLE#	AU* ppb
B 41551	3
B 41552	1
B 41553	2
B 41554	45
B 41555	1
B 41556	2
B 41557	5

ACME ANALYTICAL LABORATORIES LTD.
 852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
 PHONE(604)253-3158 FAX(604)253-1716

DATE RECEIVED: NOV 25 1988

Dec. 1/88

GEOCHEMICAL ANALYSIS CERTIFICATE

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER. THIS LEACH IS PARTIAL FOR MN FE SR CA P LA CR MG BA TI B W AND LIMITED FOR NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM. - SAMPLE TYPE: Core AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

SIGNED BY *C. Long* D. TOYE, C. LEONG, B. CHAN, J. WANG; CERTIFIED B.C. ASSAYERS

DAIWAN ENGINEERING LTD. PROJECT MORAGA RES./EXPO FILE # 88-6013 Page 1

SAMPLE#	Mo PPM	Cu PPM	Pb PPM	Zn PPM	Ag PPM	Au* PPB
C 34557	9	1000	28	107	.1	46
C 34558	5	394	24	116	.1	32
C 34559	3	427	31	105	.1	45
C 34560	10	1015	23	94	.2	53
C 34561	19	709	18	82	.1	33
C 34562	40	811	20	71	.1	52
C 34563	15	428	21	74	.1	16
C 34564	9	636	24	87	.1	32
C 34565	2	290	19	92	.1	14
C 34566	7	291	22	68	.1	18
C 34567	20	570	32	91	.1	31
C 34568	7	409	14	51	.1	35
C 34569	8	478	20	80	.2	32
C 34570	16	502	32	94	.2	21
C 34571	5	177	20	87	.1	10
C 34572	26	258	16	67	.1	8
C 34573	8	420	23	84	.1	22
C 34574	39	578	13	53	.1	9
C 34575	11	498	19	66	.1	15
C 34576	6	280	23	87	.1	24
C 34577	8	517	14	48	.1	32
C 34578	4	288	24	52	.1	13
C 34579	7	108	13	38	.2	6
C 34580	6	184	17	47	.1	5
C 34581	2	183	15	43	.1	2
C 34582	5	198	18	57	.1	11
C 34583	1	129	20	47	.1	5
C 34584	1	117	11	36	.1	7
C 34585	51	107	8	34	.1	6
C 34586	6	83	13	44	.1	3
C 34587	1	97	17	41	.1	6
C 34588	5	149	9	46	.1	5
C 34589	1	51	9	34	.1	1
C 34590	1	129	16	34	.1	3
C 34591	1	102	14	42	.1	1
C 34592	2	100	13	23	.2	4
STD C/AU-R	19	63	43	132	6.9	480

SAMPLE#	Mo PPM	Cu PPM	Pb PPM	Zn PPM	Ag PPM	Au* PPB
C 34593	3	177	6	38	.1	8
C 34594	136	1135	10	52	.1	15
C 34595	7	267	10	44	.2	9
C 34596	1	158	34	63	.1	12
C 34597	1	138	17	37	.1	6
C 34598	9	153	19	31	.1	9
C 34599	26	627	28	38	.1	28
C 34600	2	81	15	31	.1	8
C 34601	1	99	6	32	.1	7
C 34602	2	217	6	23	.1	5
C 34603	1	24	5	30	.1	1
C 34604	1	26	18	29	.1	1
C 34605	4	51	11	36	.1	2
C 34606	3	26	17	36	.1	9
C 34607	11	141	16	56	.1	12
C 34608	1	191	21	41	.1	8
C 34609	1	95	14	41	.1	6
C 34610	1	35	8	31	.1	2
C 34611	1	88	16	36	.2	7
C 34612	1	14	18	23	.1	2
C 34613	1	41	9	30	.1	1
C 34614	1	79	13	30	.1	11
C 34615	1	57	12	31	.1	2
C 34616	1	90	9	26	.1	8
C 34617	3	56	9	31	.3	1
C 34618	1	75	8	27	.2	3
C 34619	4	95	5	25	.1	4
C 34620	1	132	17	24	.2	2
C 34621	2	86	8	31	.1	3
C 34622	1	53	6	23	.1	4
C 34623	1	113	8	28	.1	4
C 34624	1	77	6	26	.2	6
C 34625	1	38	9	24	.1	1
C 34626	1	24	11	27	.1	2
C 34627	1	45	14	34	.1	1
C 34628	2	151	8	31	.1	8
STD C/AU-R	19	60	37	137	7.4	490

SAMPLE#	Mo PPM	Cu PPM	Pb PPM	Zn PPM	Ag PPM	Au* PPB
C 34629	2	16	14	19	.1	9
C 34630	2	23	14	19	.1	3
C 34631	1	63	24	29	.1	2
C 34632	2	58	19	21	.2	12
C 34633	1	103	22	31	.1	11
C 34634	1	82	18	53	.1	3
C 34635	1	56	15	21	.1	7
C 34636	2	53	19	24	.1	6
C 34637	4	77	20	25	.1	3
C 34638	1	81	11	26	.1	30
C 34639	3	63	12	29	.5	6
C 34640	3	39	15	23	.1	3
C 34641	3	43	20	24	.2	4
C 34642	1	75	12	25	.4	9

1 copy to camp

ACME ANALYTICAL LABORATORIES LTD. DATE RECEIVED: NOV 28 1988
852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
PHONE(604)253-3158 FAX(604)253-1716 DATE REPORT MAILED: Dec. 1/88

GEOCHEMICAL ANALYSIS CERTIFICATE

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER.
THIS LEACH IS PARTIAL FOR MN FE SR CA P LA CR MG BA TI B W AND LIMITED FOR NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM.
- SAMPLE TYPE: Core AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

SIGNED BY... *C. Long* D. TOYE, C. LEONG, B. CHAN, J. WANG; CERTIFIED B.C. ASSAYERS

DAIWAN ENGINEERING LTD. PROJECT-MORAGA RES/EXPO FILE # 88-6042 Page 1

SAMPLE#	Mo PPM	Cu PPM	Pb PPM	Zn PPM	Ag PPM	Au* PPB
C 34643	1	27	9	21	.1	9
C 34644	1	174	12	22	.1	14
C 34645	1	72	11	13	.2	3
C 34646	1	108	7	12	.1	3
C 34647	1	100	6	17	.4	3
C 34648	40	2859	28	141	.6	330
C 34649	61	3713	21	105	.8	410
C 34650	40	1923	17	86	.5	225
C 34651	31	2517	26	132	.5	290
C 34652	37	2197	21	116	.5	260
C 34653	43	1902	15	83	.5	86
C 34654	22	1226	16	123	.4	117
C 34655	34	1367	13	82	.5	116
C 34656	22	1140	9	107	.4	101
C 34657	29	2005	17	133	.6	114
C 34658	24	1239	11	122	.3	86
C 34659	36	2312	32	146	.8	76
C 34660	20	1115	13	141	.4	91
C 34661	19	1508	29	99	.7	61
C 34662	24	1972	15	106	.7	66
C 34663	27	1517	21	117	.6	68
C 34664	27	1305	21	111	.5	74
C 34665	21	1358	21	150	.5	49
C 34666	12	1077	16	109	.5	74
C 34667	23	1186	11	92	.3	98
C 34668	18	1572	13	128	.5	29
C 34669	12	731	9	70	.2	44
C 34670	25	1416	14	89	.4	87
C 34671	15	1176	11	65	.3	94
C 34672	14	548	8	93	.3	32
C 34673	9	670	20	75	.2	34
C 34674	8	660	19	67	.1	47
C 34675	23	5037	22	184	2.2	690
C 34676	21	5067	23	199	2.2	720
C 34677	24	5021	36	282	2.0	750
C 34678	20	2841	37	234	1.0	365

SAMPLE#	Mo PPM	Cu PPM	Pb PPM	Zn PPM	Ag PPM	Au* PPB
C 34679	20	2967	28	162	.7	410
C 34680	13	2239	26	148	.7	390
C 34681	19	2533	24	156	.7	400
C 34682	24	2605	21	143	.7	430
C 34683	28	3426	23	146	1.0	380
C 34684	17	2964	29	156	1.0	410
C 34685	20	2376	31	119	.7	290
C 34686	35	3024	23	157	.8	330
C 34687	30	2308	10	126	.6	290
C 34688	8	1667	8	55	.7	610
C 34689	18	3072	35	153	1.2	600
C 34690	15	1994	19	129	.6	310
C 34691	27	3611	26	206	1.7	510
C 34692	53	2028	33	328	.9	320
C 34693	18	2882	28	212	.9	490
C 34694	21	3004	23	189	.8	540
C 34695	17	3847	17	120	1.0	770
C 34696	142	5063	28	170	1.2	810
C 34697	38	3760	13	130	1.1	620
STD C/AU-R	18	59	37	133	6.9	470

ACME ANALYTICAL LABORATORIES LTD.
 852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
 PHONE(604)253-3158 FAX(604)253-1716

DATE RECEIVED: DEC 1 1988

DATE REPORT MAILED: Dec. 2./88..

GEOCHEMICAL ANALYSIS CERTIFICATE

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER. THIS LEACH IS PARTIAL FOR MN FE SR CA P LA CR NG BA TI B W AND LIMITED FOR NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM.
 - SAMPLE TYPE: Core AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

SIGNED BY *C. Long*. D. TOYB, C. LEONG, B. CHAN, J. WANG; CERTIFIED B.C. ASSAYERS

DAIWAN ENGINEERING LTD. PROJECT MORAGA RES. - EXPO FILE # 88-6109 Page 1

SAMPLE#	Mo PPM	Cu PPM	Pb PPM	Zn PPM	Ag PPM	Au* PPB
C 34698	32	3882	2	125	1.0	650
C 34699	31	2419	15	155	.8	370
C 34700	30	2854	26	312	1.3	450
C 34701	21	2706	5	150	.8	280
C 34702	20	2315	8	125	.7	240
C 34703	17	1656	15	122	.6	260
C 34704	33	3062	8	193	1.1	270
C 34705	38	2502	13	146	1.1	240
C 34706	30	3103	17	177	1.5	220
C 34707	108	2523	12	80	1.0	180
C 34708	30	2189	24	215	.8	108
C 34709	18	2327	14	107	.5	145
C 34710	190	2675	21	56	.9	135
C 34711	26	1793	10	148	.5	125
C 34712	48	2235	18	123	.6	166
C 34713	31	2605	13	168	1.0	167
C 34714	141	2447	19	153	.8	103
C 34715	65	2711	24	201	1.0	159
C 34716	23	2071	12	126	.7	144
C 34717	18	3349	16	120	.9	490
C 34718	61	3321	8	132	1.5	280
C 34719	25	1629	6	134	.3	112
C 34720	19	1673	13	116	.5	113
C 34721	35	2442	4	84	.5	210
C 34722	127	2888	12	124	.6	250
C 34723	125	3134	5	72	.6	270
C 34724	156	9330	10	113	1.5	910
C 34725	84	4350	10	128	.9	380
C 34726	41	3999	11	129	1.0	390
C 34727	53	4435	8	95	1.2	440
C 34728	99	3007	2	79	.7	270
C 34729	57	3198	5	197	.8	125
C 34730	43	2809	2	59	.4	184
C 34731	41	1602	7	60	.3	109
C 34732	28	862	21	115	.4	52
C 34733	9	201	16	68	.2	11
STD C/AU-R	19	63	41	141	7.3	530

SAMPLE#	Mo PPM	Cu PPM	Pb PPM	Zn PPM	Ag PPM	Au* PPB
C 34734	8	201	6	71	.1	12
C 34735	4	179	9	60	.1	14
C 34736	3	220	16	42	.1	21
C 34737	28	2603	11	32	.1	67
C 34738	18	386	12	53	.2	13
C 34739	36	79	8	30	.3	17
C 34740	9	249	6	50	.1	10
C 34741	6	172	3	60	.1	13
C 34742	7	192	9	52	.1	26
C 34743	9	315	8	50	.1	16
C 34744	11	127	17	26	.1	18
STD C/AU-R	18	58	39	128	7.1	485

ACME ANALYTICAL LABORATORIES LTD.
 852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
 PHONE(604)253-3158 FAX(604)253-1716

DATE RECEIVED: DEC 7 1988

DATE REPORT MAILED: Dec. 12/88

GEOCHEMICAL ANALYSIS CERTIFICATE

ICP - .500 GRAM SAMPLE IS DIGESTED WITH 3ML 3-1-2 HCL-HNO3-H2O AT 95 DEG. C FOR ONE HOUR AND IS DILUTED TO 10 ML WITH WATER. THIS LEACH IS PARTIAL FOR MM FE SR CA P LA CR NG BA TI B W AND LIMITED FOR NA K AND AL. AU DETECTION LIMIT BY ICP IS 3 PPM.
 - SAMPLE TYPE: Core AU* ANALYSIS BY ACID LEACH/AA FROM 10 GM SAMPLE.

SIGNED BY: *C. Leong* D. TOYK, C. LEONG, B. CHAN, J. WANG; CERTIFIED B.C. ASSAYERS

DAIWAN ENGINEERING LTD. PROJECT MORAGA RES.-EXPO FILE # 88-6181 Page 1

SAMPLE#	MO PPM	Cu PPM	Pb PPM	Zn PPM	Ag PPM	Au* PPB
C 34745	6	106	6	30	.2	3
C 34746	6	93	11	33	.1	5
C 34747	11	186	10	26	.1	4
C 34748	17	219	5	32	.4	10
C 34749	9	99	6	24	.1	2
C 34750	2	73	5	23	.1	9
C 34751	3	91	11	36	.7	2
C 34752	1	68	8	32	.1	31
C 34753	2	46	8	28	.7	1
C 34754	1	88	8	35	.4	5
C 34755	1	44	5	26	.1	1
C 34756	5	44	7	25	.6	4
C 34757	5	35	5	19	.2	7
C 34758	33	1609	14	98	.8	154
C 34759	18	2326	18	101	1.0	270
C 34760	15	1384	21	94	.6	108
C 34761	12	955	19	91	.9	77
C 34762	13	1365	26	95	1.0	129
C 34763	15	1281	21	174	1.0	105
C 34764	15	1589	23	98	.8	152
C 34765	18	1909	22	148	.4	126
C 34766	31	996	21	89	.1	24
C 34767	10	806	22	102	.5	61
C 34768	7	586	24	94	.4	43
C 34769	6	573	23	73	.7	53
C 34770	14	834	18	93	.5	72
C 34771	20	1027	35	106	.5	107
C 34772	13	1319	20	96	.2	102
C 34773	59	3595	18	122	2.0	78
C 34774	26	1429	29	211	1.3	81
C 34775	18	1807	19	154	.8	200
C 34776	14	1400	17	149	.7	109
C 34777	10	1095	19	119	1.0	97
C 34778	29	1362	61	282	1.7	36
C 34779	15	1413	21	131	1.5	95
C 34780	24	1659	44	166	1.8	53
STD C/AU-R	19	60	41	134	7.4	490

SAMPLE#	Mo PPM	Cu PPM	Pb PPM	Zn PPM	Ag PPM	Au* PPB
C 34781	68	1487	18	123	1.0	31
C 34782	136	2557	26	134	1.5	105
C 34783	34	2097	20	170	.6	142
C 34784	31	1694	26	100	.6	132
C 34785	15	1268	21	86	.5	90
C 34786	12	2064	13	72	.7	210
C 34787	17	1375	16	68	.4	120
C 34788	75	1675	17	98	.7	129
C 34789	17	1832	18	82	.6	197
C 34790	19	1686	13	115	.6	125
C 34791	17	1265	14	244	.5	84
C 34792	22	1744	22	115	.8	132
C 34793	13	3360	14	111	.7	480
C 34794	28	2101	19	99	.5	127
C 34795	20	1375	10	56	.4	84
C 34796	28	2161	11	81	.4	177
C 34797	36	2290	16	90	.5	195
C 34798	33	2518	23	95	.5	120
C 34799	119	4898	29	97	1.1	320
C 34800	68	2836	19	75	.7	210
C 34801	45	2777	15	81	.7	174
C 34802	167	1100	16	48	.4	116
C 34803	50	936	8	48	.3	32
C 34804	29	784	14	38	.2	27
C 34805	27	949	11	36	.3	57
C 34806	26	564	14	36	.3	19
C 34807	29	561	11	47	.2	17
C 34808	107	969	13	33	.2	38
C 34809	5	288	15	33	.2	7
C 34810	12	384	12	30	.3	167
C 34811	15	549	20	45	.4	29
C 34812	13	1102	15	69	.4	26
C 34813	10	634	11	48	.2	19
C 34814	14	966	17	50	.4	35
C 34815	26	899	12	34	.5	24
C 34816	10	1003	19	40	.5	41
STD C/AU-R	19	61	40	133	6.8	520

SAMPLE#	MO PPM	CU PPM	PB PPM	ZN PPM	AG PPM	AU* PPB
C 34817	114	2406	4	41	.6	87
C 34818	36	923	8	43	.4	23
C 34819	35	1469	11	50	.4	53
C 34820	25	885	2	49	.4	22
C 34821	29	1005	5	57	.2	58
C 34822	72	864	8	57	.4	16
C 34823	38	678	7	63	.3	31
C 34824	40	390	7	51	.2	5
C 34825	40	745	12	65	.4	8
C 34826	32	861	12	54	.5	17
C 34827	44	1252	8	55	.6	14
C 34828	32	1646	6	44	1.1	8
C 34829	64	1430	2	46	.8	15
C 34830	22	834	8	50	.6	6
C 34831	49	261	8	34	.2	4
C 34832	196	633	4	51	.3	17
C 34833	65	1911	8	42	.5	59
C 34834	30	930	16	55	.3	28
C 34835	56	3484	2	47	.8	220
C 34836	80	1286	3	41	.5	76
C 34837	56	1353	9	41	.5	49
C 34838	57	766	7	47	.4	16
C 34839	16	586	9	37	.3	14
C 34840	34	816	9	42	.5	23
C 34841	16	280	2	41	.1	15
C 34842	47	284	8	38	.2	18
C 34843	30	326	25	48	.2	10
C 34844	1	152	10	32	.3	12
C 34845	3	124	5	29	.3	10
C 34846	3	153	10	30	.2	14
C 34847	49	257	2	41	.2	21
C 34848	3	70	5	21	.3	8
C 34849	1	51	6	24	.3	10
C 34850	1	78	4	24	.1	15
C 34851	8	227	278	378	.4	20
C 34852	3	119	17	42	.2	7
STD C/AU-R	19	62	42	132	7.0	490

SAMPLE#	Mo PPM	Cu PPM	Pb PPM	Zn PPM	Ag PPM	Au* PPB
C 34853	7	175	21	33	.1	14
C 34854	24	646	39	53	.3	36
C 34855	8	321	23	116	.1	34
C 34856	20	3054	16	231	.4	370
C 34857	11	1780	14	224	.1	260
C 34858	4	395	29	120	.1	37
C 34859	3	140	7	73	.1	10
C 34860	19	1927	17	200	.4	310
C 34861	29	4648	14	194	1.1	430
STD C/AU-R	18	61	43	132	6.8	525

ACME ANALYTICAL LABORATORIES LTD. DATE RECEIVED: DEC 12 1988
852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
PHONE(604)253-3158 FAX(604)253-1716 DATE REPORT MAILED: Dec. 15/88...

ASSAY CERTIFICATE

- SAMPLE TYPE: Pulp AU** BY FIRE ASSAY FROM 1/2 A.T.

SIGNED BY... *C. Long* . D. TOYE, C. LEONG, B. CHAN, J. WANG; CERTIFIED B.C. ASSAYERS

DAIWAN ENGINEERING LTD. PROJECT MORAGA RES. - EXPO FILE # 88-6109R

SAMPLE#	Cu %	Au** OZ/T
C 34698	.51	.017
C 34699	.27	.009
C 34700	.32	.012
C 34701	.29	.008
C 34702	.25	.007
C 34723	.32	.008
C 34724	.97	.027
C 34725	.47	.010
C 34726	.46	.012
C 34727	.48	.011

ACME ANALYTICAL LABORATORIES LTD. DATE RECEIVED: DEC 12 1988
852 E. HASTINGS ST. VANCOUVER B.C. V6A 1R6
PHONE(604)253-3158 FAX(604)253-1716 DATE REPORT MAILED: Dec. 15./88..

ASSAY CERTIFICATE

- SAMPLE TYPE: Pulp AU** BY FIRE ASSAY FROM 1/2 A.T.

SIGNED BY..... *C. Long* D. TOYR, C. LEONG, B. CHAN, J. WANG; CERTIFIED B.C. ASSAYERS

DAIWAN ENGINEERING LTD. FILE # 88-6042R

SAMPLE#	Cu %	Au** OZ/T
C 34648	.30	.009
C 34649	.39	.011
C 34650	.21	.006
C 34651	.27	.007
C 34652	.24	.007
C 34675	.56	.018
C 34676	.58	.018
C 34677	.55	.020
C 34678	.34	.009

APPENDIX 4

DRILL LOGS

HOLE NO.: **NF-88-1**

PROJECT: **MORAGA RES. - EXPLD.**

PAGE NO.: **1** OF **10**

EC-159

COLLAR ELEV.:

GROUND ELEV.:

DATE STARTED: **NOV 12/68**

REF. TO CLAIM CORNER:

COORDINATES:

N.

E.

DATE FINISHED: **NOV 16/68**

SCALE:

INCLINATION: **60**

BEARNG: **007**

TOTAL DEPTH: **274.32 m.**

LOGGED BY: **EDM.**

SECTION	ALTERATION			FRACTURING	MINERAL	GEOLOGY	COMMENTS:	AVE CORE REC'Y / MOLE	SULPHIDES	DRILLING INTERVAL	% CORE RECOVERED	CORE SIZE	SAMPLE INTERVAL	% REC'Y SAMP. INT.	ESTI-MATED					
															PPM					
														Mo	Cu	Pb	Zn	Ag		
12-19							OVERBURDEN 0-12.19													
13							12.19-30.6 <u>Andesite Porphyry</u> - <u>Ag. Phen</u> 4.5mm. (15-20%) GREENISH-GRAY to PINKISH GRAY. MAFIC PHEN. ALT TO CHLORITE. WEAKLY TO MOD. SILICIFICATION TO LATER WEAK ARG ALT SURROUNDING QTE FILLED FRACTURES GENERALLY @ 20° AND UNMINERALIZED. (64m LOCAL Py DIS. TO 5% OCCASIONALLY ASSOCIATED TO HARLINE QTE. AT VARIOUS ANGLES. VERY OCCASIONAL CALCITE TO QTE.													
14							QZ HEALD BECCIA R 10CM. Py 2-5%			46			3455	2.81	9	1000	28	107	0.1	
15										5.0	90									
16													3455	3.0						
17							PHEN 4.3mm 5-10%. Generally R more SILICIOUS.			32					5	377	24	116	0.1	
18																				
19										5.7	84		3455	3.0						
20										45					3	427	31	105	0.1	
21																				
22													3455	3.0						
23										53					10	1015	23	94	0.2	

HOLE NO.: **HP-88-1**

COLLAR ELEV.:

COORDINATES:

INCLINATION:

N. E.

BEARING:

PROJECT:

DATE STARTED:

DATE FINISHED:

TOTAL DEPTH:

PAGE NO.: **3** OF **10**

REF. TO CLAIM CORNER:

SCALE:

LOGGED BY:

SECTION	ALTERATION		FRACTURING	MINERAL	GEOLOGY	COMMENTS:	AVE CORE RECY / HOLE	SULPHIDES	DRILLING INTERVAL	% CORE RECOVERED	CORE SIZE	SAMPLE INTERVAL	% RECY. SAMP. INT.	ESTI-MATED					
	CLAY	SILICIFICATION												ppm					
														Mo	Cu	Pb	Zn	Ag	
53																			
54																			
55																			
56																			
57																			
58																			
59																			
60																			
61																			
62																			
63																			
64																			
65																			
66																			
67																			
68																			
69																			
70																			
71																			
72																			
73																			
74																			
75																			
76																			
77																			
78																			
79																			
80																			
81																			
82																			

NOTE: 53.40 - 58.35 POSSIBLE TUFF LIME-MOD CA VERY SILICEOUS.

VEEY QTE VEIN @ 60° S-5mm DRUSSY QTE. Py 5%

QTE-CARB @ 45° 10mm LAMINATED. VUE TO DRUSSY QTE. MASSIVE Py. 20% ABOVE WK COUSE 20mm HIGH ANGLE LAMINATED VEIN

59.70-64.0 SHORR @ 10-15° LOCAL CALCITE. Py DIS. 1%. SLICKENSIDES. TRACE Mn STAIN.

FRAG 50mm. SUB-ANGULAR TO SUB-ROUNDED Fg (BLACK SILICEOUS MATRIX. DISS. Py 2-3%

CALCITE FRACTURE 2mm @ 10° QTE VEIN 4mm @ 80° Py 10%

DYKE. Fg. EDR. HIGH SILICEOUS. DK CARBON. POSS CHLORINE ALT. LOWER CONTACT @ 50°

WEAK SHEARING TO MINOR COUSE ALONG SURFACES. QTE-CARB VEINS @ 20-30° VUEBY.

QTE-CARB VEIN 10mm @ 20° LAMINATED. VUE TO ACICULAR WHITE ZEBULITE.

WEAK FRACTURING @ 10-20° TO IRRREGULAR QTE, QTE-CARB AND ZEBULITE FILLING. VUEBY. Py STRINGS OCC. MASSIVE.

10cm COUSE IRRREGULAR SILICEOUS MATE. SIMILAR TO PORPHYRYIC SECTIONS. 10cm WELL FRACTURED TO MINOR

58.35 - 83.0 BRECCIA. MOD-HIGH SILICIFICATION. FRAG 5 10mm COMPRISE ≈ 20% OF RX SET IN Fg. MATRIX, ANGULAR TO SUB-ROUNDED. DIS. Py 0.5-2% LOCALLY.

HOLE NO.: HF-88-1.

PROJECT:

PAGE NO.: 7 OF 10

COLLAR ELEV.:

GROUND ELEV.:

DATE STARTED:

REF. TO CLAIM CORNER:

COORDINATES:

N. E.

DATE FINISHED:

SCALE:

DIP INCLINATION:

BEARING:

TOTAL DEPTH:

LOGGED BY:

SECTION	ALTERATION		FRACTURING	MINERAL	GEOLOGY	COMMENTS:	AVE CORE REC'Y / HOLE	SULPHIDES	DRILLING INTERVAL	% CORE RECOVERED	CORE SIZE	SAMPLE INTERVAL	% REC'Y. SAMP. INT.	ESTI-MATED					
	Py	SiO ₂												ppm	Mo	Cu	Pb	Zn	Ag
173																			
174						QTE-CARRS VEIN 5-15MM @ 5°		2			34612	3.0	1	14	18	23	0.1		
175						Py 1% WEAK SLICKENSIDES AT WALL BY CONTACT TO MINOR MM. STAIN.													
176																			
177						R1 LESS SILICEOUS		1	5.1	94	34613	3.0	1	41	9	30	0.		
178																			
179						R1 MOTTLED BROWN + PALE GREEN (SIDITE/CHLORITE + CRIDITE)													
180								11			34614	3.0	1	79	13	30	0.1		
181																			
182																			
183								2	5.7	96	34615	3.0	1	57	12	31	0.1		
184																			
185						HEALED BRECCIA W 2-3MM QTE-CARRS VEIN @ 5°. MINOR WHITE ZEOLITE													
186						QTE VEIN 40MM @ 15-20°													
187						Py 30-40%.		8			34616	3.0	1	90	9	26	0.1		
188						WK-MUD CONG. OCCASIONAL QTP													
189						BLEBS W 2-5% Py.													
190						QTE STRINGERS 1-2mm @ 10°		1	5.7	95	34617	3.0	3	56	9	31	0.1		
191						2-3% Py. BLEBS OF CALCITE.													
192						QTE VEIN 2-5mm @ 5-10°													
193						W ANGULAR WALL RT.		3			34618	3.0	1	75	8	27	0.2		
194																			
195						CORE FRACTURED AND CALCITE HEALED.													
196								4	5.7	88	34619	3.0	4	95	5	25	0.1		
197						CORE SHATTERED W MINOR GOUGES.													
198						6CM BRECCIA CALCITE HEALED													
199						CROSSCUT BY 8MM QTE VEIN @ 40°													
200						IRREGULAR CARR-QTE @ 5° W		2			34620	3.0	1	132	17	24	0.2		
201						WK CONG.													
202																			
203						10MM QTE VEIN W 20% Py													
204																			
205						IRREGULAR GROUNDING MINOR BLEBS (5-15°) QTE CARRS. IRREGULAR		3			34621	3.0	2	86	9	31	0.1		

HOLE NO.: HF 88-1

PROJECT:

PAGE NO.: 8 OF 10

COLLAR ELEV.:

GROUND ELEV.:

DATE STARTED:

REF. TO CLAIM CORNER:

COORDINATES:

N. E.

DATE FINISHED:

SCALE:

DIP INCLINATION:

BEARING:

TOTAL DEPTH:

LOGGED BY:

SECTION	ALTERATION	FRACTURING	MINERAL	GEOLOGY	COMMENTS:	AVE CORE REC'Y / HOLE	SULPHIDES	DRILLING INTERVAL	% CORE RECOVERED	CORE SIZE	SAMPLE INTERVAL	ESTIMATED							
												ppm							
												Mo	Cu	Pb	Zn	Ag			
203	CLAY S-K/CA						Asph												
204							4	5.6	98	34622	3.0	1	53	6	23	0.1			
205							4			34623	3.0	1	113	8	28	0.1			
206					QZ-CARB VEIN @ 25' 10-15mm MINOR VEGS. 15-20% Py.														
207																			
208																			
209																			
210					QZ HEALED FRACTURES 5-15° "TUFFACEOUS"		6	5.5	98	34624	3.0	1	77	6	26	0.2			
211																			
212																			
213							1			34625	3.0	1	38	9	24	0.1			
214																			
215					QZ VEIN (MINOR CARB) < 5mm @ 5-15° WEAKLY LAMINATE c. 1% Py. ZEOOLITES IN VEG.		2	5.4	94	34626	3.0	1	24	11	27	0.1			
216																			
217																			
218																			
219							1			34627	3.0	1	45	14	34	0.1			
220																			
221																			
222							8	5.9	93	34628	3.0	2	157	8	31	0.1			
223																			
224					INTRUSIVE AEG- QZ DYKE. CONTACTS IRREGULAR @ 20-30°. CUT BY QZ-CARB VEIN 55mm @ 5° D MINOR Py.		9			34629	3.0	2	16	14	19	0.1			
225																			
226					QZ VEIN 2-6mm @ 5-10° Py 50% Pr. 10% Cp 20%														
227					Py STRUNGERS														
228					Py ALGAS + STRUNGERS. 226.8-227.05		3	5.5	98	34630	3.0	2	23	14	19	0.1			
229																			
230																			
231					Py STRUNGERS @ 15°		2	5.4	100	34631	3.0	1	63	24	29	0.1			
232																			
233																			

230.0 - 248.0 - WEAK TO MODERATE LEACHING SURROUNDING VEINS AND FRACTURES.

HOLE NO.: HF88-1

COLLAR ELEV.:

COORDINATES:

INCLINATION:

GROUND ELEV.:

N. E.

BEARING:

PROJECT:

DATE STARTED:

DATE FINISHED:

TOTAL DEPTH:

PAGE NO.: 10 OF 10

REF. TO CLAIM CORNER:

SCALE:

LOGGED BY:

SECTION	ALTERATION		FRACTURING	MINERAL	GEOLOGY	COMMENTS:	AVE CORE RECY / HOLE	SULPHIDES	DRILLING INTERVAL	% CORE RECOVERED	CORE SIZE	SAMPLE INTERVAL	% RECY. SAMPLE INT.	ESTI-MATED					
	CLAY	SILICA												PPM					
														Mn	Cu	Pb	Zn	Ag	
263																			
264						QZ VEIN. ≤ 15 MM @ 20° Py 2%.					34643	3.0							
265						Py STRINGS @ $5-10^\circ$													
266								14	5.7	96	34644	3.0	1	174	12	22	0.1		
267																			
268						MODERATE TO STRONG CROSS HATCHED QZ + QZ CARB STRINGERS.													
269						30 MM COUCC @ 35° .		3			34645	3.0	1	72	11	13	0.1		
270																			
271									55	96									
272						LAYERED QZ VEIN 5.6 MM @ 80°		3			34646	3.0	1	108	7	12	0.1		
273																			
274								3	2.92	95	34647	1.32	1	100	6	17	0.4		
						274.74													

DESCRIPTIVE GEOLOGY

266.9 - 274.32 - INTRUSIVE (GRANODIORITE?) MED GRAINED.
MOTTLED GREEN. FELD - 30-40% BIOT - 10%
CALCITE - 15-20% OTHER MINERALS - 10% QZ - 10-20%
OCCASSIONAL CHALCEDONIC LAYERS QZ VEINS
MINOR WHITE TO PINK CALCITE. DIS Py S. 5%
MINOR Py MS W QZ STRINGERS. SILICEOUSNESS
DECREASES W DEPTH. SCATTERED IRREGULAR
PATCHES OF SOFT BRICK-RED MATERIAL (SIDORITE?)

HOLE NO.: HF-88-2

PROJECT:

PAGE NO.: 2 OF 3

COLLAR ELEV.:

GROUND ELEV.:

DATE STARTED:

REF. TO CLAIM CORNER:

COORDINATES:

N. E.

DATE FINISHED:

SCALE:

INCLINATION:

BEARING:

TOTAL DEPTH:

LOGGED BY:

SECTION	ALTERATION			FRACTURING	MINERAL	GEOLOGY	COMMENTS:	AVE CORE REC'Y / HOLE	SULPHIDES	DRILLING INTERVAL	% CORE RECOVERED	CORE SIZE	SAMPLE INTERVAL	% REC'Y. CAMP. INT.	ESTI-MATED				
	Si	Ca	Other												Mo	Cu	Pb	Zn	Ag
40							MOD. CONCR. 80% GRAY QZ L		76			34661	1.5	36	2312	32	146	0.8	
41							3-4% Py. ON FRACT.		91			34662	0.5	20	1115	13	141	0.4	
42							MINOR SOFT MILKY WHITE KALIN.		61			34663	1.5	19	1578	29	99	0.7	
43							5% Py ON FRACT.												
44							10mm GRAY QZ VEIN @ 10°	45.0 - 75.59. WHITE CALCITE IS COMMON	66			34662	2.5	24	1972	15	106	0.7	
45							Py 6 1/2 ON FRACT.	FRACTURE COATING.											
46																			
47									68	5.45	60	34663	3.0	27	1517	21	117	0.6	
48							SOFT QZ PRIS IN GOUCE												
49									74			34664	3.0	27	1305	21	111	0.5	
50							50-60° GOUCE. ONE CONTACT @ 40°												
51																			
52																			
53							QZ VEIN 5mm @ 30°		49	4.93	65	34665	3.0	21	1352	21	150	0.5	
54							2-3% MAGNETITE												
55							SILICIOUS 2-3% Py.												
56									74			34666	3.0	12	1077	16	109	0.5	
57																			
58									98	5.3	68	34667	1.83	23	1186	11	92	0.3	
59							QZ VEIN 20mm @ 45°												
60									29			34668	0.67	18	1572	13	128	0.5	
61							MOD. TO STRONG GOUCE CALCITE ON FRACTURE												
62									44			34669	2.5	12	731	9	70	0.2	
63																			
64							STRONG GOUCE MINOR CALCITE ON FRACT.		87	5.6	71	34670	3.0	25	1416	14	89	0.4	
65																			
66																			
67									94	6.9	60	34671	2.0	15	1120	11	65	0.3	
68							68.58-69.49. GROUND CORE. ONLY 10CM. REMAINS												
69									32			34672	3.0	14	808	9	92	0.2	

HOLE NO.: HF 88-2

PROJECT:

PAGE NO.: 3 of 3

COLLAR ELEV.:

GROUND ELEV.:

DATE STARTED:

REF. TO CLAIM CORNER:

COORDINATES:

N. E.

DATE FINISHED:

SCALE:

DIP/SLURRY:

BEARING:

TOTAL DEPTH:

LOGGED BY:

SECTION	ALTERATION		FRACTURING	MINERAL	GEOLOGY	COMMENTS:	AVE CORE REC'Y / HOLE	% SULPHIDES	DRILLING INTERVAL	% CORE RECOVERED	CORE SIZE	SAMPLE INTERVAL	% REC'Y SAMP. INT.	ESTI-MATED					
	Qz	St												ppm	Mo	Cu	Pb	Zn	Ag
70																			
71																			
72																			
73																			
74																			
75																			

Qz vein 10cm @ 30°
black metallic fracture
filling (magnetite).

ALTHOUGH CORE IS STRONGLY COULED THE
FRAGMENTS OF ANDESITE ARE STILL MOD-STRONGLY
SILICEOUS.

34	3.79	48	34622	3.0					
34			34623	2.0	9	670	20	75	0.2
47			34624	2.59	8	660	19	67	0.1

75.59 EOH.

HOLE NO.: HF 88-4

COLLAR ELEV.:

COORDINATES:

DIP: -60°

GROUND ELEV.:

N. E.

BEARING: 090°

PROJECT:

DATE STARTED: NOV 20/88

DATE FINISHED: NOV 23/88

TOTAL DEPTH: 187.15

PAGE NO.: 1 OF 7

REF. TO CLAIM CORNER:

SCALE: METRIC

LOGGED BY: EDN

EC-162

SECTION	ALTERATION	FRACTURING	MINERAL	GEOLOGY	COMMENTS:	AVE CORE REC'Y / HOLE	SULPHIDES	DRILLING INTERVAL	% CORE RECOVERED	CORE SIZE	SAMPLE INTERVAL	% REC'Y SAMP INT.	ESTI-MATED				
													PPM				
													Mo	Cr	Pb	Zn	Ag
					OVERBURDEN TO 6.71 M												
					MINOR Fe STAINING ON FRACT.												
8					6.71 - 15.5 <u>ANDESITE PORPHYRY</u> . Fg LIGHT SILICEOUS GROSS MATRIX W/ DARK-GREEN IRREGULARLY SHAPED PHENO ≤ 5MM. WEAK TO MOD MAGNETIC. PHENO ARE SOFT AND MAJOR MICACEOUS (CLORITE). Py ≤ 1% (SUBHEDRAL TO ANHEDRAL) IS DISS. AND APPEARS TO OCCUR PRIMARILY WITHIN THE PHENO. VERY LITTLE Qtz VEINING EXCEPT FOR OCCASIONAL HAIRLINE STRINKERS @ 5-45°	690			34625	2.2	23	5037	22	184	2.2		
9						720	5.4	87	34676	3.0	21	5067	23	199	2.2		
10						750			34677	3.5	24	5081	36	282	2.0		
11						365	5.6	95	34678	2.5	20	2841	37	234	1.0		
12						410			34679	3.0	20	2467	28	162	0.7		
13						390			34680	3.0	13	2234	26	148	0.7		
14						400	5.2	85	34681	3.0	19	2533	24	156	0.7		
15						430			34682	2.9	24	2605	21	143	0.7		
16						380	4.8	98	34683	2.1	28	3426	23	146	1.0		
17						410				2.0	17	2464	24	152	1.0		
18																	
19																	
20																	
21																	
22																	
23																	
24																	
25																	
26																	
27																	
28																	
29																	
30																	
31																	
32																	
33																	
34																	

OVERBURDEN TO 6.71 M

MINOR Fe STAINING ON FRACT.

CALCITE VEIN 6.3MM @ 45°

VALLEY CALCITE STRIKER 6.2MM X-CUT Qtz STRINKERS W/ MINOR DISPLACEMENT.

Qtz. USM 4.0CM @ 50° IRREGULARLY LAMINATED W/ CP 5-10% W/ TRACE BARNITE

3.2 STRIKERS @ 25-30° SURROUNDED BY 1-2MM OF SOFT DARK GREEN ALT.

Qtz-CARB (PINK) VEIN 5.0MM @ 40°

Qtz-CARB VEIN ≤ 4mm @ 25-30° CUTS 5CM DYKE AND PORPH.

- COUPE 5CM @ 45°.

PINKISH Qtz-CARB STRINKERS ≤ 3mm @ 5-15°. FINE TRACE Qtz STRINKERS SURROUNDING

LESS SILICEOUS.

DESCRIPTIVE GEOLOGY

6.71 - 15.5 ANDESITE PORPHYRY. Fg LIGHT SILICEOUS GROSS MATRIX W/ DARK-GREEN IRREGULARLY SHAPED PHENO ≤ 5MM. WEAK TO MOD MAGNETIC. PHENO ARE SOFT AND MAJOR MICACEOUS (CLORITE). Py ≤ 1% (SUBHEDRAL TO ANHEDRAL) IS DISS. AND APPEARS TO OCCUR PRIMARILY WITHIN THE PHENO. VERY LITTLE Qtz VEINING EXCEPT FOR OCCASIONAL HAIRLINE STRINKERS @ 5-45° STILL LESS COMMON ARE STRINKERS OF WHITE CALCITE.

15.5 - 29.9 - ANDESITE BRECCIA.

BARY TO GREEN. Fg. SPARKLY SILICEOUS OCCASIONAL PATCHES OF PORPHYRY. PATCHY CALCITE ALT. 1-2%. WEAK CALCITE FRACTURE FILLING BY BARRON OF SULFIDES EXCEPT FOR LOCAL DENS AND FRACTURE FILLINGS OF CP ≤ 5%

29.9 - 42.0 ANDESITE PORPHYRY. W/ WK-MOD BRECCIATION LOCALLY. SIMILAR TO 6.91-15.5. Py APPEARS TO BE MORE EVENLY DISSEMINATED THROUGHOUT Rf AND MORE COMMON AS FRACTURE COATING.

HOLE NO.: HF 88-4

PROJECT:

PAGE NO.: 2 of 7

COLLAR ELEV.:

GROUND ELEV.:

DATE STARTED:

REF. TO CLAIM CORNER:

COORDINATES:

N. E.

DATE FINISHED:

SCALE:

INCLINATION:

BEARING:

TOTAL DEPTH:

LOGGED BY:

SECTION	ALTERATION	FRACTURING	MINERAL GEOLOGY	COMMENTS:	AVE CORE RECY / MOLE	SULPHIDES	DRILLING INTERVAL	% CORE RECOVERED	CORE SIZE	SAMPLE INTERVAL	ESTI-MATED				
											ppm				
											Mo	Cu	Pb	Zn	Ag
34	CLAY SILICA	MOD		RY FRACTURED QTE-CARB HEALED Py 5-10% DIS + ON FRACT.		290			34687	0.6	20	2370	31	119	0.5
35				330	5.5	95	34688	0.7	35	3024	23	157	0.5		
36				290			34689	2.7	30	2308	10	126	0.6		
37				610			34690	1.0	8	1667	8	55	0.7		
38				600	4.9	98	34691	1.0	18	3022	35	153	1.2		
39				310			34692	1.0	15	194	19	129	0.6		
40				510			34693	1.0	27	3611	26	206	1.7		
41				320			34694	1.0	53	3028	33	328	0.9		
42				490	5.0	92	34695	3.0	18	2882	28	212	0.9		
43				540			34696	3.0	21	3004	23	189	0.8		
44				720			34697	1.0	17	3047	17	120	1.0		
45				810	5.3	97	34698	1.0	142	5064	28	170	1.2		
46				620			34699	1.0	38	3760	13	130	1.1		
47				600			34700	3.0	32	3882	2	125	1.0		
48				370	5.2	89	34701	3.0	31	2449	15	155	0.8		
49				450			34702	3.0	30	2854	26	312	1.3		
50				280	5.2	91	34703	3.0	21	2704	5	150	0.8		

42.0 - 68.4 - ANDESITE w LOCAL BRECCIA VERY SILICEOUS w MINOR LOCAL CLAY ACT.

NOTE: CORE WK - MOD FRACTURED. HEALED TO QTE, CARB. AND PINKISH EROLITE.

9% UGIN 5CM @ 5-10% Py 3-4%

HOLE NO.: HF 88-4

CELLAR ELEV.:

COORDINATES:

INCLINATION:

GROUND ELEV.:

N. E.

BEARING:

PROJECT:

DATE STARTED:

DATE FINISHED:

TOTAL DEPTH:

PAGE NO.: 7 OF 7

REF. TO CLAIM CORNER:

SCALE:

LOGGED BY:

SECTION	ALTERATION	FRACTURING	MINERAL GEOLOGY	COMMENTS:	AVE CORE REC'Y / HOLE	% SULPHIDES	DRILLING INTERVAL	% CORE RECOVERED	CORE SIZE	SAMPLE INTERVAL	ESTI-MATED								
											ppm								
											Mo	Cu	Pb	Zn	Ag				
64	CU, PY																		
65																			
66																			
67																			
68																			
69																			
70																			
71																			
72																			
73																			
74																			
75																			
76																			
77																			
78																			
79																			
80																			
81																			
82																			
83																			
84																			
85																			
86																			
87																			
88																			
89																			
90																			
91																			
92																			
93																			
94																			

CORE STRONGLY FRACTURED AND HEALED TO QTZ, CARB + PINK FELDS. WK. SKARN? SURF LAYER CONTACT @ 5-10°. Py 4-10%. 68.4-69.0- Py 25% + FRACT 3-4%.

5-10cm GORGE @ 5-10°. QTZ-VEIN > 2cm @ 0-5° LOWER CONTACT @ 30°

CORE LT GRAY ANDGAS SILICEOUS BUT IS MOD SOFT WHICH SCRATCHED. Py DIS. 6-2%. QTZ VEIN 4mm @ 20° Py 5-10% ALONG LINEAR CONTACT.

WK. 20cm GORGE @ 30°

MID-STROKE 30cm GORGE. CORE GRAY WK-MOD FELD. ACT. QTZ VEIN 2-10cm. DIS Py + Py ON FRACT 5%.

CROSSMATCHED FRACTURING HEALED TO QTZ + QTZ-CARB (FeCO₃ COMMON) CORE STILL SPOTTY BUT MORE MARK (BASALT?).

QTZ VEIN 15cm @ 45° GRAY CORE. VARIABLE SILICEOUSNESS. 2-3% DIS Py.

WK-MOD CROSS HATCH STRANGLER QTZ + QTZ-CARB (FeCO₃).

68.4 - 111.0 ANDESITE PORPHYRY
DISS. Py 1% TO LOCAL CONCENTRATIONS ON FRACTURE SURFACES 5%.

DEPTH	% SULPHIDES	DRILLING INTERVAL	% CORE RECOVERED	CORE SIZE	SAMPLE INTERVAL	% RECY. STAMP. INT.	Mo	Cu	Pb	Zn	Ag
240	Aug 85			34702	3.2	20	2315	8	125	0.7	
260	6.0		88	34703	1.2	17	1654	15	122	0.6	
270				34704	0.6	33	3062	8	192	1.1	
240				34705	2.0	38	2502	13	146	1.1	
220				34706	1.3	30	3109	17	177	1.5	
180	5.2		94	34707	3.4	108	2523	12	90	1.0	
108				34708	2.3	30	2189	24	215	0.8	
145	5.6		91	34709	2.3	18	2327	14	107	0.5	
135				34710	1.3	190	275	21	56	0.9	
125				34711	1.4	26	1793	10	148	0.5	
166	5.7		91	34712	3.0	48	2225	18	123	0.6	
167				34713	1.9	31	2205	13	168	1.0	
103	5.3		94	34714	1.56	141	2442	19	153	0.8	
159				34715	1.24	65	2911	24	201	1.0	
144				34716	3.3	23	2001	12	126	0.7	

HOLE NO.: HF 88-4

COLLAR ELEV.:

COORDINATES:

INCLINATION:

GROUND ELEV.:

N. E.

BEARING:

PROJECT:

DATE STARTED:

DATE FINISHED:

TOTAL DEPTH:

PAGE NO.: 4 of 7

REF. TO CLAIM CORNER:

SCALE:

LOGGED BY:

SECTION	ALTERATION		FRACTURING	MINERAL	GEOLOGY	COMMENTS:	AVE CORE REC'Y / HOLE	SULPHIDES	DRILLING INTERVAL	% CORE RECOVERED	CORE SIZE	SAMPLE INTERVAL	ESTI-MATED							
	CLAY	SILICA											ppm							
													No	Cu	Pb	Zn	Ag			
94																				
95																				
96						WK CONC. 2-3% Py.		490	5.4	94	34717	2.4	18	3349	16	120	0.9			
97																				
98						IRREGULAR PARTING QTE VEIN GREENISH-GREY SANDY MATERIAL (PYROPHILITE?)		280			34718	2.3	61	3321	8	132	1.5			
99																				
100								112	5.6	93	34719	2.3	25	1629	6	134	0.2			
101																				
102						101.0 - 124.0 - <u>ANDESITE</u> Fg. DARK GREY. MOD-STRONGLY SILICEOUS. WEAKLY MAGNETIC. PARTING CHLORITE ACT. 3-5%. POORLY FRACTURED AT ALL ANGLES AND QTE HEAVY. QTE STRAINING OCCASIONALLY SHOW LAMINAE.														
103								113			34720	3.0	19	1673	13	116	0.5			
104						2cm CONC														
105								210	5.4	94	34721	3.0	35	2442	4	84	0.5			
106																				
107																				
108																				
109						UGLY.														
110						WK-MOD CONC. Py 1-2%.		250			34722	3.0	127	2500	12	124	0.6			
111																				
112						NO BLES ON MANT.		270	6.3		34723	1.1	125	3134	5	72	0.6			
113						STRONGER @ 35° TO Cp 2-3%. HEMATITE 10-15% Py 5-10%. MORE STRONGLY MAGNETIC THAN WALL RT.		910		92	34724	0.7	156	3130	10	114	1.5			
114								380			34725	1.0	84	4350	10	128	0.9			
115																				
116						Mo also.		390			34726	3.1	41	3999	11	129	1.0			
117						5cm. QTE-CARB. @ .10°		440	5.5	95	34727	0.9	53	4425	8	95	1.2			
118						52% Cp. Finely Dis.		270			34728	0.7	99	3007	2	79	0.7			
119						66% CORE. LOCALLY SILICEOUS LOCAL CONC.		125			34729	0.6	57	3190	5	197	0.3			
120								184			34730	1.7	43	2809	2	59	0.4			
121						STRONGLY FRACTURED + QTE, QTE-CARB, FeO1 HEAVY.		109	5.0	90	34731	2.5	41	1602	7	60	0.3			
122						MOD-STRONG CONC. MOD MAGNETIC.														
123						DIS Py 1-2%.		52			34732	2.5	28	862	21	115	0.4			

HOLE NO.: HF 88-4.

PROJECT:

PAGE NO.: 5 of 7

COLLAR ELEV.:

GROUND ELEV.:

DATE STARTED:

REF. TO CLAIM CORNER:

COORDINATES:

N. E.

DATE FINISHED:

SCALE:

INCLINATION:

BEARING:

TOTAL DEPTH:

LOGGED BY:

SECTION	ALTERATION		FRACTURING	MINERAL GEOLOGY	COMMENTS:	AVE CORE RECY / HOLE	% SULPHIDES	DRILLING INTERVAL	% CORE RECOVERED	CORE SIZE	SAMPLE INTERVAL	% RECY. SAMP. INT.	ESTIMATED				
	CLAY	SILICA											P.P.M.				
					DESCRIPTIVE GEOLOGY								Mo	Cu	Pb	Zn	Ag
124					STAIN GOUGE												
125						124.0-139.5 - <u>ANDESITE TUFF.</u>											
126						Fg. MOD-DR. GRAY. VERY SILICEOUS. FRAG 3mm	11			34733	3.0	9	201	16	68	0.2	
127						FINE QZ VEINS (GENERALLY $\leq 3\mu m$ @ 10-40°)		5.9	93								
128						DIS. SULPHIDES (Py) 5%. OCCASIONALLY CONCENTRATED				34734	3.0	8	201	6	71	0.1	
129						ALONG FRACTURES. LK WHITE CARBONATE ON	12										
130						OCCASIONAL FRACTURES.											
131						FRAG 4mm.											
132						5mm QZ-CARB VEIN @ 20°	14			34735	3.0	4	179	9	60	0.1	
133																	
134						5mm QZ-CARB VEIN @ 20°	21	5.4	96	34736	1.5	3	220	16	42	0.1	
135						IRREGULAR PATCH (2.5cm)	67			34737	0.5	28	203	11	32	0.1	
136						GREEN SOAPY MATERIAL (PYRROPHILITE?)											
137						SULPHIDE (Py) CROSS 10-15%.	13			34738	3.0	18	306	12	53	0.2	
138																	
139						QZ VEIN @ 40° Py CROSS & STRIKERS 2-4%	17	5.6	95	34739	0.8	36	77	8	20	0.3	
140																	
141						3mm QZ VEIN @ 20° MINOR GOUGE.	10			34740	3.2	9	249	6	50	0.1	
142																	
143																	
144						CORE STILL GRAY BUT W/ IRREGULAR GREEN BLOTCHES $\leq 3\mu m$.	13	5.3	98	34741	3.0	6	172	3	60	0.1	
145						SOFTER THAN MATRIX.											
146						2mm Py STRIKER @ 90°											
147						20mm GOUGE.	26			34742	3.0	7	192	9	52	0.1	
148						2mm Py STRIKER @ 90°											
149						20cm WK-MOD GOUGE.											
150							16	5.7	98	34743	3.0	9	315	8	50	0.1	
151						MID SECTION W/ CARBONATE HEALED FRACTURES ABOVE.											
152							18			34744	1.0	11	127	17	26	0.1	
153						WK. GOUGE. MIXTURE QZ & WALL RT (50/50). Py 5% @ 50°.	3			34745	3.0	6	106	6	30	0.1	

HOLE NO.: HF 88-5

PROJECT:

PAGE NO.: 3 OF 7

COLLAR ELEV.:

GROUND ELEV.:

DATE STARTED:

REF. TO CLAIM CORNER:

COORDINATES:

N. E.

DATE FINISHED:

SCALE:

INCLINATION:

BEARING:

TOTAL DEPTH:

LOGGED BY:

SECTION	ALTERATION	FRACTURING	MINERAL	GEOLOGY	COMMENTS:	AVE CORE REC'Y / HOLE	% SULPHIDES	DRILLING INTERVAL	% CORE RECOVERED	CORE SIZE	SAMPLE INTERVAL	ESTIMATED								
												PAM								
												Mo	Cu	Pb	Zn	Ag				
60	SILICA				<p>60-61 - VUGGY QZ-CARB VEIN 5 CM @ 0-10°</p> <p>62-63 - 4CM QZ VEIN UNDERLYING 2CM STAINING SURGE.</p> <p>64-65 - 6.3% - MANGANESE BLEBS ALONG FRACTURE.</p> <p>66-67 - QZ VEIN 1CM @ 75° Py-1% Cp 1% CUT BY QZ-CARB VEIN @ 10° DISPLACEMENT 2.5MM.</p> <p>70-71 - QZ VEIN 5 CM @ 0-5° SULFIDES < 2% (Py/Cp SP/ST) SURROUNDING CORE W/ LEACHING TO GREEN ALT (EPIDOTE?) ADJACENT TO VEIN.</p> <p>75-76 - STAINING SURGE 7CM.</p> <p>85-86 - QZ VEIN 5CM @ 20° CP 5.5% BLEBS W/ MOD ACT. SURFACED (EPIDOTE)</p> <p>87-88 - 87.5-87.9 - MOD LEACHED W/ 5% Py</p> <p>88-89 - FRACTURE @ 5° W/ MANGANESE STAIN W/ SILICATES.</p> <p>89-90 - STAINING SURGE @ 50°</p>	<p>62.6-63.3 - TUFFACEOUS DS-SULFIDET 3-5% Py 1-3% Cp 1-2%</p> <p>75.3-76.2 - CORE DR. GREEN AND SOFTER. FROM 75.9-76.2 CORE APPEARS LEACHED W/ BLOTCHES OF SOFT PALE-GREEN Waxy MATERIAL (Pyrophyllite?). 1-2% Py.</p> <p>87.5-90.5 - W/ LEACHED ZONE W/ LOCAL BRECCIATION AND COARSE.</p>	120	5.1	94	34787	2.3	17	1675	16	68	0.4				
61							129			34788	3.0	75	1675	17	98	0.7				
62							197			34789	3.0	17	1675	18	82	0.6				
63							5.2			92	25	34790	3.0	19	1675	13	115	0.6		
64											84	34791	2.0	17	1675	14	244	0.5		
65											132	34792	2.3	22	1675	22	115	0.8		
66											480	34793	1.0	13	1675	14	111	0.7		
67											127	34794	2.7	28	1675	19	99	0.5		
68											5.3	96	84	34795	3.0	20	1675	10	56	0.4
69													177	34796	3.0	28	1675	11	81	0.4
70													195	34797	2.4	36	1675	16	90	0.5
71													120	34798	0.5	33	1675	23	95	0.5
72													320	34799	2.6	118	1675	28	0.7	

HOLE NO.: HF 88-5

CELLAR ELEV.:

COORDINATES:

INCLINATION:

GROUND ELEV.:

N. E.

BEARING:

PROJECT:

DATE STARTED:

DATE FINISHED:

TOTAL DEPTH:

PAGE NO.: 4 OF 7

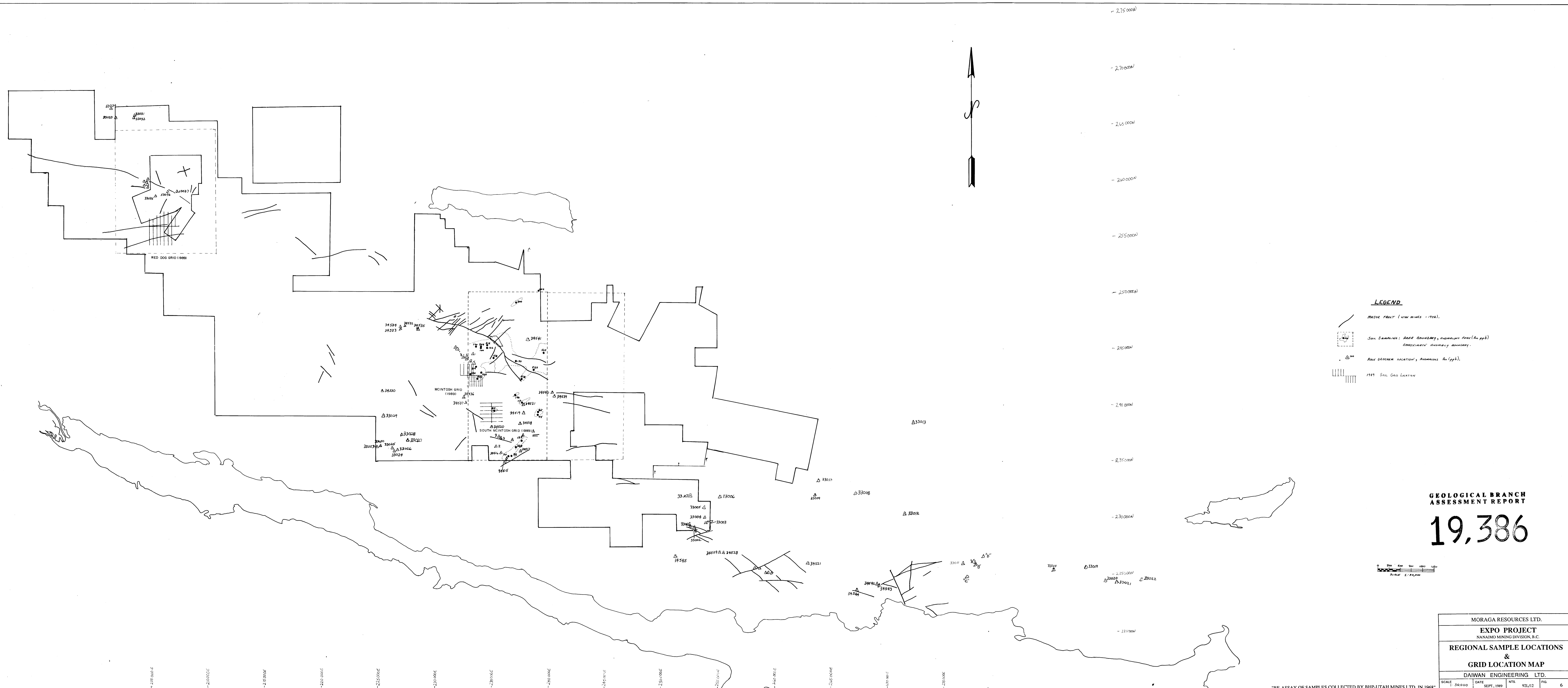
REF. TO CLAIM CORNER:

SCALE:

LOGGED BY:

SECTION	ALTERATION	FRACTURING	MINERAL	GEOLOGY	COMMENTS:	AVE CORE RECY / HOLE	SULPHIDES	DRILLING INTERVAL	% CORE RECOVERED	CORE SIZE	SAMPLE INTERVAL	ESTIMATED							
												ppm							
												Mo	Cu	Pb	Zn	Ag			
90	SILICA				90.1-90.5 - D.K. BRECCIA QTR-CARB VEINING @ 45°. WR-MOD LEACHED.		As ₂ S ₃			34800	2.0	68	2036	19	75	0.7			
91																			
92								LN CALCITIC FRACTURES.					34801	1.8	45	2077	15	81	0.7
93													34802	1.5	167	1100	16	48	0.4
94								98.3-95.8 - MOD-STRAIN GOUGE 3CM QTR-CARB VEIN @ 0-10°	95.6-95.8 - UNDULATING BAND OF TUFF (FINE BRECCIA?) @ 30°. MOD SOFT. FANG ANGULAR - ROUNDED				34803	2.2	50	936	8	48	0.3
95								10MM QTR VEIN @ 80°	≤ 4mm (MAY 10MM).				34804	1.0	29	704	14	38	0.2
96								98.4-98.8 - LN BRECCIA HEALED TO QTR AND MINOR CARB. P ₁ 2-3%.					34805	3.0	27	949	11	36	0.3
97								4mm QTR VEIN @ 0-5°. CORE W/ LEACHED					34806	1.0	26	564	14	36	0.3
98													34807	0.5	29	561	11	47	0.2
99								103.1-103.3 - LEACHED P ₁ STRINGS @ 45°.					34808	1.5	107	469	13	33	0.2
100													34809	1.5	5	200	15	33	0.2
101								MOD RECOVERED TO 100% (P ₁ 2-3% (LEACHED 4-5%)). QTR-CARB HEALED STRINGS SILICIFICATION @ 5-10°					34810	1.5	12	304	12	30	0.3
102								4mm QTR VEIN @ 0-5°. CORE W/ LEACHED					34811	3.0	15	544	20	45	0.4
103								103.1-103.3 - LEACHED P ₁ STRINGS @ 45°.					34812	3.0	13	102	15	64	0.4
104													34813	1.8	10	634	11	48	0.2
105								MOD RECOVERED TO 100% (P ₁ 2-3% (LEACHED 4-5%)). QTR-CARB HEALED STRINGS SILICIFICATION @ 5-10°					34814	1.0	14	966	17	50	0.4
106								TRACER OF QTR + QTR-CARB HEALED FRACTURES.					34815	1.0	26	394	12	34	0.5
107								IRREGULAR PATCHES AND STRINGS OF QTR AND PINK RESID. NO MINERALIZATION.					34816	1.0	10	103	19	40	0.5
108								114.2-114.5 - MOD GOUGE @ 10°					34817	1.0	10	103	19	40	0.5
109								115.2-115.3 - MOD GOUGE @ 40°					34818	1.0	10	103	19	40	0.5
110									34819	1.0	10	103	19	40	0.5				
111				115.7-116.35 - IRREGULAR PATCHES QTR AND CARB. P ₁ BLEBS 3-4% MINOR MAGNETITE ALONG QTR FRACTURES.					34820	1.0	10	103	19	40	0.5				
112									34821	1.0	10	103	19	40	0.5				
113									34822	1.0	10	103	19	40	0.5				
114									34823	1.0	10	103	19	40	0.5				
115									34824	1.0	10	103	19	40	0.5				
116									34825	1.0	10	103	19	40	0.5				
117									34826	1.0	10	103	19	40	0.5				
118									34827	1.0	10	103	19	40	0.5				
119									34828	1.0	10	103	19	40	0.5				
120									34829	1.0	10	103	19	40	0.5				

1190-1230 P₁ content 3-4%

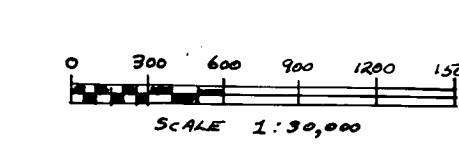


LEGEND

- MORAGA FAULT (UTAH MINES - 1988)
- SOIL SAMPLING: DATA BOUNDARY, ANOMALOUS AREA (Aa pp6)
ANOMALOUS AREA BOUNDARY
- △ ANOMALOUS AREA LOCATION, ANOMALOUS AREA (Aa pp6)
- ||||| 1989 SOIL GRID LOCATION

**GEOLOGICAL BRANCH
ASSESSMENT REPORT**

19,386



MORAGA RESOURCES LTD.			
EXPO PROJECT NANAIMO MINING DIVISION, B.C.			
REGIONAL SAMPLE LOCATIONS & GRID LOCATION MAP			
DAIWAN ENGINEERING LTD.			
SCALE: 1:30,000	DATE: SEPT. 1989	NTS: 92L/12	PG: 6

RE-ASSAY OF SAMPLES COLLECTED BY BHP-UTAH MINES LTD. IN 1968