

ASSESSMENT REPORT
DIAMOND DRILLING
on the
RED CAP CLAIMS
Taku River Area
Atlin Mining Division

Cap #2 and Cap #4 Claims
N.T.S. Maps 104K-11W & E, 14W
58°44'N, 133°17'W

Owner
OMNI RESOURCES INC.
900 - 475 Howe Street
Vancouver, B.C.
V6C 2B3

by
Terence M. Elliott, M.S.
Geologist

11 June 1982

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INTRODUCTION

As a result of an extensive soil sampling project on the Red Cap claims in 1980, large anomalies of copper, gold, molybdenum and silver were discovered. During the summer of 1981 a diamond drilling program was initiated to test the anomalous areas for the possibility of locating a large-tonnage body of porphyry copper-molybdenum-gold mineralization.

Drilling commenced July 23, 1981 and terminated September 4, 1981.

LOCATION AND ACCESS

The property is on the southeast side of the Taku River, approximately 120 km south of Atlin, B.C. and 75 km east of Juneau, Alaska. It is located at N.58°44', W.133°17', along the western flanks of Mt. Lester Jones (see Figure 1). Elevations in the area of the geochemical anomalies vary from 800 metres at Red Cap Lake to over 1800 metres on top of the ridge west of Mt. Lester Jones.

Access is from Whitehorse or Atlin by fixed wing airplane to a gravel airstrip that is located near Tulsequah approximately 16 km southwest of the property. Alternatively, a float plane can land on the Taku River approximately 4 km west of the heart of the property. A helicopter can then fly from these points to the showings and drill sites.

Our camp was located northwest of Mount Ogden at Border Lake, about 27 km south of the Red Cap claims. Personnel were flown daily in a Hughes 500D helicopter to and from the drill sites.

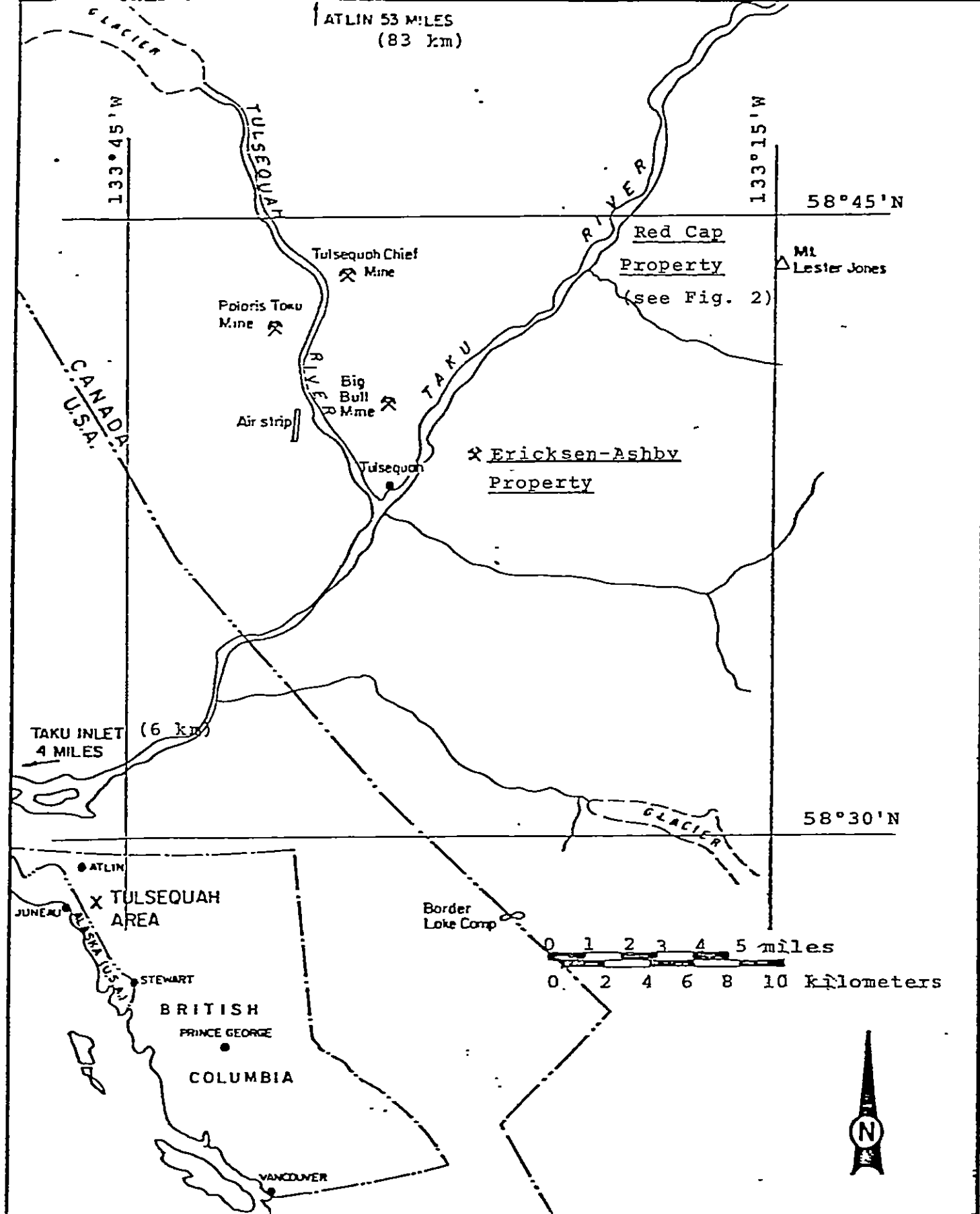
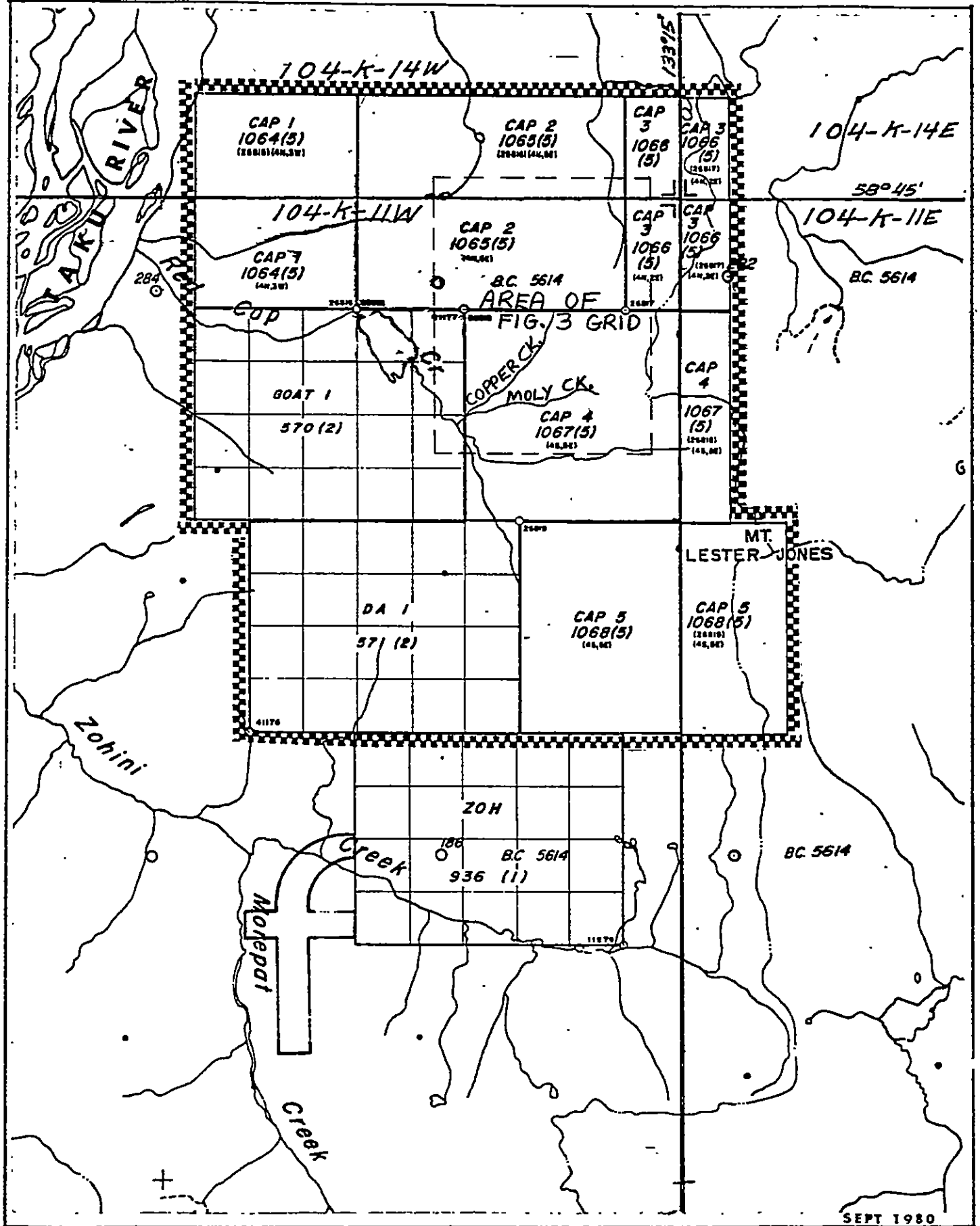


FIGURE 1
 Area Location Map
 Omni Resources Inc.
 Red Cap Project
TULSEQUAH B.C. AREA

- Late 1920's Assayable gold, silver and base metals discovered in the present claim area as reported in the 1930 and 1931 Minister of Mines Annual Reports.
- 1946 An unknown prospector (?) staked a claim bearing an initial claim post tag numbered 149450.
- 1959, 1960 J.G. Souther of the Geological Survey of Canada (reported in G.S.C. Memoir 362) mapped veins containing sphalerite, galena, chalcopyrite and arsenopyrite in a large area of pyritization, silicification and carbonate alteration.
- 1970 The property was staked as the "Lynne" claims by Archer and Cathro Ltd. for the Cordero Mining Co. (Sun Oil Co.). Soil, rock and silt sampling outlined some anomalous areas.
- 1971 Five short (deepest hole was 24 feet) x-ray drill holes failed to penetrate the surface oxidation zone. Molybdenum was found to be present in geochemically significant amounts.
- 1979 The Goat #1 and DA #1 claims were staked southwest of the main Red Cap showing area.
- 1980 A soil sampling program resulted in the discovery of strong copper, molybdenum, silver and gold anomalies in an area of over 2.5 square kilometres. Among other Cap group claims, the Cap #2 and Cap #4 claims were staked. Independent geological consultants H. Wahl and G. Rayner each spent approximately 2 weeks mapping different portions of the main zone geochemical anomaly outcrops. H. Wahl recommended a minimum of 7,500 feet of diamond drilling to test the mineralized zone.

CLAIMS AND OWNERSHIP (see Figure 2)

The Cap #2 and Cap #4 claims are wholly owned by Omni Resources Inc. The following list gives pertinent data on the claims:



RED CAP CLAIMS - LOCATION MAP

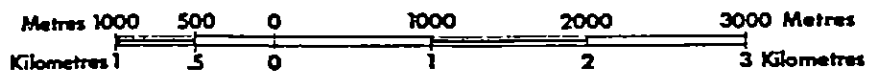


FIG-2

CLAIM	RECORD NO.	UNITS	DATE RECORDED	EXPIRY DATE
Cap #2	1065	20	14 May 1980	14 May 1983
Cap #4	1067	20	14 May 1980	14 May 1983

DRILL HOLE SUMMARY

PURPOSE and PROPERTY GEOLOGY

The purpose of the 1981 drill program was to discover a porphyry copper-molybdenum-gold-silver ore body in the area previously outlined as containing highly anomalous metal-bearing soil samples.

The Red Cap claims are located in a northwest-trending belt of Upper Triassic Stuhini Group volcanic rocks. In the area of anomalous soil geochemistry, these volcanics are acid in composition, whereas Geological Survey of Canada mapping elsewhere in the Tulsequah area has indicated that the Stuhini Group volcanics are andesites and basalts.

G. Rayner has mapped an area coincident with the soil geochemical anomaly which is strongly hornfelsed and weakly mineralized. H. Wahl has mapped the same area as a zone of strong silicification and pyritization of Stuhini(?) rhyolites.

Locally, G. Rayner has mapped areas of brecciation of two types: shatter brecciation near grid coordinates 1045 N. and 1100 W., and a pyritized diatreme breccia which is poorly exposed around 1050 N. and 1140 W.

The Stuhini(?) rhyolites and dacites have been intruded by several igneous rock units. The unit of largest areal extent is a granodiorite to quartz monzonite body which lies along the northwest side of the soil geochemical anomaly. This intrusive body is clearly post-mineralization in age and dykes of it cut Stuhini(?) volcanics along the contact. Other late stage andesite and trachyte dykes also cut the altered rhyolites and dacites.

A small intrusive body which contains disseminated pyrite lies north of drill holes RC81-3, 4, 6 and 7. This diorite porphyry is also later than the mineralizing event.

The 1981 drilling indicated that there are also other small intrusive bodies cutting the acid volcanic sequence. Hence G. Rayner has coined the name for this area as "the Red Cap igneous center" which has "produced a fairly typical volcanic-type porphyry copper event."

WORK DONE

Between July 23 and September 4, 1981, a total of 1203.6 metres (3948 feet) of NQ drilling was completed at 4 drill sites. Seven holes were drilled to depths ranging from 132.3 to 260.0 metres.

Holes RC-1 and 2 were drilled by Arctic Diamond Drilling Company's Longyear 38 drill. Holes RC-3 to 7 were drilled by Arctic Diamond Drilling Co. personnel operating a Longyear Super 38 drill owned by Berglynn Developments Ltd.

All drill sites are natural benches to which a helicopter can readily transport the drills and personnel. Rototech Helicopters of Delta, B.C. contracted a Hughes 500D helicopter to Omni Resources Inc. for the summer's work. Drillers and other personnel were transported daily to the drill sites from Omni's base camp at Border Lake to the south.

DRILL HOLE SPECIFICATIONS

July-September 1981

<u>Drill Hole</u>	<u>Location</u>	<u>Elevation (m)</u>	<u>Azimuth</u>	<u>Dip</u>	<u>Depth (m)</u>	<u>Total (m)</u>
RC81-1	1080N 1090W	1715	123°	-55°	184.7	184.7
RC81-2	1080N 1090W	1715	300°	-60°	152.4	337.2

<u>Drill Hole</u>	<u>Location</u>	<u>Elevation (m)</u>	<u>Azimuth</u>	<u>Dip</u>	<u>Depth (m)</u>	<u>Total (m)</u>
RC81-3	1004N 1086W	1375	020°	-60°	260.0	597.3
RC81-4	1004N 1086W	1375	197°	-60°	154.0	715.3
RC81-5	1016N 1066W	1475	030°	-60°	152.4	867.7
RC81-6	1010N 1110W	1240	008°	-54°	167.6	1071.3
RC81-7	1010N 1110W	1240	188°	-55°	132.3	1203.6

CORE STORAGE LOCATION

Core from the Red Cap property drilling was flown to the camp at Border Lake where it was logged and stored. The core shack is located at the north side of camp immediately southeast of the heliport. Red Cap core is stored in the east rack nearby core from Island Mining & Exploration Co. Ltd. core from the Ericksen-Ashby property.

RESULTS, INTERPRETATIONS, AND CONCLUSIONS

DDH RC81-1 (see Drill Logs and Soil Survey Grid and Drill Hole Location Map)

Hole RC81-1 cut 184.7 metres of rhyolitic tuffs and tuff breccias which have been intruded by altered feldspar porphyry dykes. The largest of these dykes has an apparent thickness of 21.6 metres (128.8-150.4 m) and is locally silicified and carbonitized. The hole was terminated in a dyke containing 0.1-0.25% pyrite and minor pyrrhotite and arsenopyrite.

Alteration of the rhyolite pyroclastics includes locally pervasive sections of phyllic and propylitic zones alternating with zones of nearly fresh rhyolite. Quartz veining is only locally strong over sections of 1-4 metres.

One section of assayable copper-silver mineralization was intersected in a zone of 3 small feldspar porphyry dykes cutting rhyolite. From 96.3-99.4 m the core assays 0.20% copper and 0.38 oz/ton silver, from 99.4-102.4 m the core

assays 2.95% copper and 3.30 oz/ton silver, and from 102.4-105.5 m the core assays 1.65% copper and 1.59 oz/ton silver (see Appendix III for Assay Results). Visible mineralization includes chalcopyrite, chalcocite, pyrite, and minor arsenopyrite.

Rock geochemical analyses of copper indicate anomalous values. Molybdenum is anomalous from 148 metres to the end of the hole.

DDH RC81-2

Hole RC81-2 cut 152.4 metres of rhyolite tuff and lapilli tuff which has been intruded by altered feldspar porphyry, andesite porphyry, and quartz monzonite to granodiorite. From 118.9-143.9 metres, the rock is an interesting breccia of altered quartz monzonite fragments in a dark gray pyrite, pyrrhotite and chlorite-rich siliceous matrix. A few rhyolite and quartz monzonite fragments contain quartz-pyrite-molybdenite veinlets or hairline molybdenite veinlets.

Pyrite veining is common throughout the hole, but quartz veining is only common between 77 and 105 metres. Chlorite veining and alteration is moderate from 103 metres to the end of the hole, but phyllic alteration is rare (cf. hole RC81-1).

Hole RC81-2 rock geochemical analyses for copper are anomalous over large sections of core. In addition, molybdenum is anomalously high from 122 metres to the end of the hole. These results are similar to the rock geochemistry in hole RC81-1.

DDH RC81-3

Hole RC81-3 intersected a diverse section of igneous rock types. From the collar down to 196 metres the rock contains a low grade intersection of molybdenite mineralization as well as anomalous copper mineralization (see Appendix III

for Assay Results). The molybdenite is found in quartz-molybdenite veins, in association with K-feldspar selvages and veins, and along fractures.

Host rocks for the mineralization are felsite, weakly altered granodiorite, felsite porphyry and granite breccia. From 196 metres to the end of the hole a quartz monzonite porphyry which is post-mineralization cuts off the interesting section above.

DDH RC81-4

Two main rock types were intersected in hole RC81-4: granodiorite to 45 metres in apparent depth and andesite-dacite porphyry for the remainder of the drill hole. Sections of the latter rock type are a breccia.

Propylitic (chlorite-epidote-calcite) alteration is common in this drill hole. Veinlets and fractures associated with this grade of alteration are chlorite-pyrite-calcite, chlorite-epidote-calcite, chlorite, and pyrite.

In hole RC81-4, molybdenite fracture coatings or molybdenite-quartz veins are common, but wide-spaced. Drill core rock geochemistry gave results of anomalous copper and molybdenum, but there are no near-economic grades.

DDH RC81-5

Drill hole RC81-5 cut a sequence of rhyolite-dacite to 114 metres in apparent depth. Thereafter to the end of the hole, the rocks were andesite-dacite porphyry similar to rock encountered in hole RC81-4.

Throughout this hole there were scattered veinlets of quartz, calcite, chlorite and pyrite. In addition, the

following combinations of veinlet type were noted: pyrite-chlorite-calcite, chlorite-calcite, chlorite-pyrite, and quartz-pyrite-molybdenite.

Rock geochemical analyses indicate anomalous values in copper and molybdenum; however, no oregrade intersections were made.

DDH RC81-6

Hole RC81-6 intersected 167.6 metres of granodiorite porphyry which contains phenocrysts of biotite and feldspar.

Molybdenite veinlets and quartz-molybdenite veinlets were common throughout the hole. Alteration of fractures and veinlet development is predominantly propylitic. Veins and veinlets are various combinations of quartz, k-feldspar, epidote, calcite, chlorite, pyrite, biotite, gypsum, anhydrite and molybdenite.

As in other holes on the property, hole RC81-6 is anomalous in both copper and molybdenum.

DDH RC81-7

Hole RC81-7 intersected 92.6 metres of intermediate volcanic rocks cut by a total of 26 metres of granodiorite dykes, and 14 metres of amygdaloidal andesite dykes. The host volcanics include dacite breccia, andesite-dacite breccia, and andesite-dacite.

Mineralization is weak in this hole. Only sparse molybdenite was observed; most of the mineralization consists of disseminated, fracture, and veinlet pyrite.

Rock geochemical analyses of core show anomalous values in copper, but molybdenum values are less than those found in holes RC81-3 to 6.

In summary, the 1981 drill holes intersected a large zone of propylitic alteration characterized by chlorite, epidote, calcite, pyrite and quartz on fractures and in veinlets. In addition, some holes intersected K-feldspar fracture selvages which may be leakage from a higher grade alteration core.

The principal ore mineral observed extensively in holes RC81-3 to 6 was molybdenite. Molybdenum and copper rock geochemistry of the drill holes indicates a widespread anomalous metal halo which may be the propylitic alteration zone of a large porphyry mineral deposit.

It is the writer's interpretation, however, that deeper drill holes may be warranted to find an orebody. G. Rayner, geologist consultant, has suggested that the main target area between drill holes 1 and 2 to the north and 3-7 to the south has not yet been drilled. This conclusion seems to the writer to be a well-grounded one. In order to accomplish this proposal, drill sites will have to be blasted out of the steep hillside of this target area.

As hole RC81-1 intersected 6.1 metres of ore grade copper-silver mineralization, at least one hole should be drilled to intersect this zone at a deeper level. This zone may be the leakage from a major system of mineralization. Two things were encouraging about this intersection: (1) its high grade of copper and (2) the presence of good grade silver; hence, an associated porphyry copper system may be rich in silver and possibly, gold.

Both holes RC81-1 and 2 are cut by feldspar porphyry dykes which may be associated with a deep, large feldspar porphyry intrusive system which could be extensively mineralized.

APPENDIX I

COST STATEMENT

and

DRILLING BILLS

Red Cap Property

Statement of Costs

(1212 m) July 16 - Sept. 6, 1981

1. Drilling			
Invoice #2217		\$ 31,818.80	
Invoice #2226		25,879.40	
Invoice #2237		70,874.34	
Deduct Credits		<u>(27,793.07)</u>	
Arctic Diamond Drilling			\$ 100,779.47
2. Helicopter			
238.4 hours @ \$324.00/hr		77,241.60	
Fuel 238.4 hours X 22 gal/hr			
X 3.90 gal		<u>20,454.72</u>	97,696.32
3. Salaries			
G. Clouthier, geologist			
20 days @ 125		2,500.00	
T. Elliott, geologist			
29 days @ 140		4,060.00	
B. Hemingway, geologist			
53 days @ 110		5,830.00	
K. Orleski, manager			
20 days @ 100		2,000.00	
D. Bergvinson, assistant			
49 days @ 55		2,695.00	
M. Kamras, assistant			
46 days @ 55		2,530.00	
R. Schmidt, prospector			
25 days @ 80		2,000.00	
D. Schmidt, assistant			
25 days @ 65		1,625.00	
W. Colwell, assistant			
15 days @ 65		975.00	
R. Merrell, assistant			
23 days @ 65		1,495.00	
C. Doulet, cook			
53 days @ 75		<u>3,975.00</u>	
		29,685.00	
+ 15% fringe		<u>4,452.75</u>	
			34,137.75
4. Camp costs			
Food - 4 drilling contractors,			
pilot & salaried employees		12,601.86	
Fuel - drill and camp fuel		9,292.50	
Camp supplies and set up		10,842.00	
Fixed wing support		<u>10,037.67</u>	42,774.33

5.	Assaying, shipping and thin sections Analysis of 580 samples for Cu, Ag, Au, Mo and Zn.	9,493.52
6.	Travel To and from work site	4,403.00
7.	Drafting and report preparation	<u>1,500.00</u>
		\$ <u>290,784.39</u>



ARCTIC DIAMOND DRILLING LTD.

184 Industrial Road, Whitehorse, Yukon Y1A 2V1 (403) 667-6434

INVOICE # 2217
August 7, 1981

Island Mining and Explorations Co. Ltd.
900-475 Howe Street,
Vancouver, B.C.
V6C 2B3

OMNI Red cap

Drilling charges for the period ended 31 July 1981

Hole #RC-81-1-55'xNQ

MOVING

110 Man hours @ 24.00 per hour 2640.00

Standby - Fogged In

145 Man hours @ 24.00 per hr 3480.00 6120.00

Overburden

0-13=13ft @ 20.80 per ft 270.40

Core Drilling

13-600=587ft @ 20.80 per ft. 12209.60

600-606=6ft @ 21.80 per ft 130.80 12340.40

Travel Time

6 Man hours @ 24.00 per hr. 144.00

Standby - Fog

42 Man hours @ 24.00 per hr. 1008.00 19882.80

Hole #RC-81-2-60'xNQ

Moving

30 Man hours @ 24.00 per hr 720.00

Overburden

0-12=12ft @ 20.80 per ft. 249.60

Core Drilling

12-500=488ft @ 20.80 per ft. 10150.40

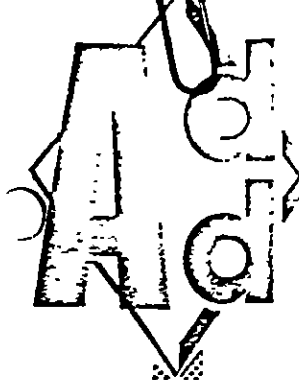
Travel Time

1 Man hour @ 24.00 per hr. 480.00

Standby - Cooking

13 Man hours @ 24.00 per hr. 312.00 792.00 11936.00

31818.80



ARCTIC DIAMOND DRILLING LTD.

184 Industrial Road, Whitehorse, Yukon Y1A 2V1 (403) 667-6434

INVOICE # 2226
September 2, 1981.

IN ACCOUNT WITH:

Island Mining And Exploration Co. Ltd.
Suite 900-475 Howe Street
Vancouver, B.C.
V6C 2B3

Drilling Charges for the Period Ended 15 August 1981.

Hole #RC-81-3-60'xNQ

Moving

145 Man hours @ 24.00 per hr. 3480.00

Overburden

0-35=35ft @ 20.80 per ft. 728.00

Reaming Casing

35-41=6ft @ 12.00 per ft. 72.00

Core Drilling

35-600=565ft @ 20.80 per ft. 11752.00

600-853=253ft @ 21.80 per ft. 5515.40 17267.40

Reaming Through Cave

2 Man hours @ 24.00 per hr. 48.00

1 Machine hour @ 12.00 per hr. 12.00 60.00

Travelling Time

20 Man hours @ 24.00 per hr. 480.00

Standby-Fogged In

116 Man hours @ 24.00 per hr. 2784.00 24871.40

Moving to Next Hole

•16 Man hours @ 24.00 per hr. 384.00

Moving Drill From Camp to Zohini to Airstrip

26 Man hours @ 24.00 per hr. 624.00

25879.40

*Approved **



ARCTIC DIAMOND DRILLING LTD.

184 Industrial Road, Whitehorse, Yukon Y1A 2V1 (403) 667-6434

Invoice # 2237
October 19, 1981

IN ACCOUNT WITH:

Island Mining & Explorations Co. Ltd.
900 475 Howe Street
Vancouver, B.C.
V6C 2B3

Drilling charges for the period ended 6 September 1981.

Hole #RC-81-4-60'xNQ

Moving

16 Man hours @ 24.00 per hour 384.00

Overburden

0-32=32ft @ 20.80 per ft. 665.60

Reaming Casing

32-50=18ft @ 12.00 per ft. 216.00

Core Drilling

32-509=477 ft @ 20.80 per ft. 9921.60

Lack of Water

1 Man hour @ 24.00 per hour 24.00

Travelling Time

6 Man hours @ 24.00 per hr. 144.00 11355.20

Hole # RC - 81 - 5 - 60' x NQ

Moving

80 Man hours @ 24.00 per hr. 1920.00

Overburden

0-20=20ft @ 20.80 per ft. 416.00

Core Drilling

20-500=480ft @ 20.80 per ft. 9984.00

Travelling Time

6 Man hours @ 24.00 per hr. 144.00

Standby - Waiting for Geologist

15 Man hours @ 24.00 per hr. 360.00

Standby-waiting to Move

16 Man hours @ 24.00 per hr. 384.00 744.00 13208.00

c/fwd..... 24563.20

*Drill. **

Hole #81-RC-6-55'xNQ

Moving

93 Man hours @ 24.00 per hr. 2232.00

Overburden

0-64=64ft @ 20.80 per ft. 1331.20

Core Drilling

64-550=486ft @ 20.80 per ft. 10108.80

Travelling Time

8 Man hours @ 24.00 per hr. 192.00

Standby

5 Man hours @ 24.00 per hr. 120.00 13984.00

Hole # RC-81-7-55'xNQ

Moving

51 Man hours @ 24.00 per hr. 1224.00

Overburden

0-30=30ft @ 20.80 per ft. 624.00

Reaming Casing

30-32=2ft @ 12.00 per ft. 24.00

Core Drilling

30-436=406ft @ 20.80 per ft. 8444.80

Reaming through Cave

10 Man hours @ 24.00 per hr. 240.00

5 Machine hrs @ 12.00 per hr. N/C 240.00

Travelling Time

6 Man hours @ 24.00 per hr. 144.00

Standby - No Night Shifts

48 Man hours @ 24.00 per hr. 1152.00 11852.80

Demobilization

Re Clause 12 of Contract.

½ x 2136.00 1068.00

Packing Up & Moving Out

18 Man hours @ 24.00 per hr. 432.00 1500.00

Use of Mud- RC 81 Holes

9 Pails Polydrill 330	1435.50	
1 Bag Quick gel	14.00	320
3 Boxes Quik troll	300.96	✓
90 Man hours @ 24.00 per hr.	2160.00	
45 Machine hrs @ 12.00 per hr.	<u>n/c</u>	3910.46

Equipment Left At Border Lake

to be Paid for as Agreed

• 21 Pails Poly drill 330	3349.50	
• 9 Boxes Quik Troll	902.88	
• 6 Timbers 8"x8"x16"	204.06	
• 3 Timbers 4"x4"x16"	47.52	
• 25 Planks 2"x10"x16'	254.00	
• 12 Timbers 6"x6"x16"	229.56	
• 5 Load Binders 5/16" ^{3/4" Coues}	156.00	
• 1 Mud Mixer Motor	391.60	
• 1 Mud Mixer U Joint	9.85	
• 1 Mud Mixer Pillow Block	19.60	
• 1 Restriction Valve #19037	126.83	
20x2' NW Casing	808.50	
14x5 NW Casing	892.64	22"
5x10 NW Casing	697.15	
1000 ft Aluminum Water Line		
c/w couplings	2226.29	
2x20114 Lock Bolt	38.28	
2x20115 Nut Washers	13.42	
• 1 x 28362 BQ Drive Rod Bushing	53.45	
• 1 x 22089 Bearing	<u>10.28</u>	10431.41
Repair to pumps:-		
Parts.		
1 - A5005 Pinion Cover	13.45	
3 - 1225327 Valve Seats	36.15	
1 - 1265098 Pinion Shaft	<u>40.00</u>	

c/fwd..... 89.60

66241.87

3 - A3434N Packing	24.00	
2 - Reduction Gears	1087.70	
2 - Dodge Couplings	260.32	
2 - 03170400 Conrod Screws	3.50	
1 - 40004600 injection nozzles	25.00	
1 - 00307102 dipstick	3.55	
1 - 00358901 Gasket set	29.75	
1 - 00359000 Gasket set	63.60	
2 - 00359601 Piston sub assembly	336.10	
1 - 00413302 Cylinderhead sub assembly	608.65	
2 - 03201800 Big end bearing	<u>58.30</u>	
	2590.07	
Plus 10%	<u>259.00</u>	
	2849.07	

Labour

25 Man hours @ 24.00 per hr.	<u>600.00</u>	3449.07
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Repairs to Drill in the Field

as per time sheets

36 Man hours @ 24.00 per hr.	864.00	
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Parts etc.

4 - 6308-2RS Bearings	65.20		
1 - B39647 Spacer	5.75		
1 - B39648 Spacer	5.75		
2 - Pails Hydraulic Oil	71.20		
3 - Cases XD3-30 Oil	102.90		
2 - Cases XD#-10 Oil	<u>68.60</u>	<u>319.40</u>	<u>1183.40</u>
			<u>70874.34</u>

APPENDIX II

STATEMENT OF QUALIFICATIONS

AUTHOR'S QUALIFICATIONS

I, Terence M. Elliott, have the following education and work experience:

1. I am a Geologist residing at #309 - 6001 Yew Street, Vancouver, British Columbia, V6M 3Y7.
2. I graduated from the University of British Columbia in 1967 with a Bachelor of Science in Honours Geology. I also received a Master of Science in Geology from Stanford University, California, in 1973.
3. I have practised my profession for 14 years.
4. I logged diamond drill holes RC81-1 and RC81-2. The remaining holes were logged by Mr. B. Hemingway, geologist.

Terence M. Elliott

APPENDIX III

ASSAY RESULTS

OMNI RESOURCES INC.

ASSAY RESULTS, RED CAP PROPERTY, TAKU RIVER AREA, B.C.

<u>D.D. Hole #</u>	<u>Location</u>	<u>Azimuth</u>	<u>Dip</u>	<u>Depth ft(m)</u>	<u>Cu ppm</u>	<u>Mo ppm</u>	<u>Ag ppm</u>	<u>Au ppb</u>	<u>Assay #</u>
RC81-1	1080N 1090W	123°	-55	11(3.4)	535	4	25	5.3	24002
				16(4.9)	1140	4	320	92.6	24003
				26(7.9)	270	3	30	2.0	24004
				36(11.0)	315	3	15	3.0	24005
				44(13.4)	355	3	40	5.5	24006
				54(16.5)	295	1	15	2.1	24007
				64(19.5)	480	2	15	2.5	24008
				73(22.3)	375	6	10	2.3	24009
				83(25.3)	690	6	495	5.2	24010
				93(28.4)	910	7	5	6.4	24011
				103(31.4)	810	4	65	6.8	24012
				113(34.5)	670	6	10	6.0	24013
				124(37.8)	950	7	15	9.5	24014
				133(40.9)	555	10	20	4.2	24015
				144(43.9)	210	3	20	2.2	24016
				156(47.6)	345	4	30	3.5	24017
				166(50.6)	175	6	100	3.6	24018
				176(53.7)	125	10	15	2.0	24019
				186(56.7)	NS	NS	NS	NS	24020
				196(59.8)	330	8	20	3.4	24021
				206(62.8)	130	6	5	1.4	24022
				216(65.9)	170	2	5	2.3	24023
				226(68.9)	120	4	40	1.7	24024
				236(72.0)	345	1	10	2.6	24025
				246(75.0)	285	2	10	3.9	24026
				256(78.0)	255	2	15	4.0	24027
				266(81.1)	225	6	15	2.5	24028
				276(84.1)	220	2	10	2.3	24029
				286(87.2)	570	2	20	4.6	24030
				296(90.2)	680	1	15	9.4	24031
				306(93.3)	300	3	15	5.8	24032
				316(96.3)	0.20*		20	13.8	24033
				326(99.4)	2.95*	1	167	113.1	24034
				336(102.6)	1.65*	1	67	54.5	24035

<u>D.D.</u> <u>Hole #</u>	<u>Location</u>	<u>Azimuth</u>	<u>Dip</u>	<u>Depth ft (m)</u>	<u>Cu</u> <u>ppm</u>	<u>Mo</u> <u>ppm</u>	<u>Ag</u> <u>ppm</u>	<u>Au</u> <u>ppb</u>	<u>Assay #</u>
RC81-1	1080N 1090W	123°	-55	346 (105.5) - 356 (108.5)	245	2	12.0	50	24036
				356 (108.5) - 366 (111.6)	65	3	0.6	10	24037
				366 (111.6) - 376 (114.6)	115	13	0.5	15	24038
				376 (114.6) - 386 (117.7)	600	16	1.2	55	24039
				386 (117.7) - 396 (120.7)	370	25	1.0	75	24040
				396 (120.7) - 406 (123.8)	440	11	1.3	110	24041
				406 (123.8) - 416 (126.8)	345	14	1.0	40	24042
				416 (126.8) - 426 (129.9)	375	13	1.1	110	24043
				426 (129.9) - 436 (132.9)	365	2	1.2	100	24044
				436 (132.9) - 446 (136.0)	150	2	0.5	170	24045
				446 (136.0) - 456 (139.0)	240	2	0.7	100	24046
				456 (139.0) - 466 (142.1)	125	6	1.9	180	24047
				466 (142.1) - 476 (145.1)	80	7	0.3	70	24048
				476 (145.1) - 486 (148.2)	165	7	0.8	145	24049
				486 (148.2) - 496 (151.2)	400	61	1.2	95	24050
				496 (151.2) - 506 (154.3)	98	6	0.3	120	24051
				506 (154.3) - 516 (157.3)	245	57	0.4	35	24052
				516 (157.3) - 526 (160.4)	61	40	0.2	10	24053
				526 (160.4) - 536 (163.4)	360	55	1.0	20	24054
				536 (163.4) - 546 (166.4)	390	40	1.1	55	24055
				546 (166.4) - 556 (169.5)	510	32	0.8	40	24056
				556 (169.5) - 566 (172.5)	215	67	0.9	30	24057
				566 (172.5) - 576 (175.6)	300	39	1.4	45	24058
				576 (175.6) - 586 (178.6)	225	29	0.8	20	24059
				586 (178.6) - 596 (181.7)	185	28	0.8	35	24060
				596 (181.7) - 606 (184.7)	97	3	1.4	25	24061
RC81-2	1080N 1090W	300°	-60	11 (3.4) - 18 (5.5)	250	2	1.7	10	24062
				18 (5.5) - 26 (7.9)	245	2	1.9	10	24063
				26 (7.9) - 36 (11.0)	370	2	3.0	25	24064
				36 (11.0) - 43 (13.1)	300	4	5.3	110	24065
				43 (13.1) - 53 (16.2)	310	2	5.6	190	24066
				53 (16.2) - 58 (17.7)	260	2	3.4	65	24067
				58 (17.7) - 66 (20.1)	270	2	3.0	130	24068
				66 (20.1) - 74 (22.6)	270	1	2.8	15	24069
				74 (22.6) - 82 (25.0)	275	2	1.6	10	24070
				82 (25.0) - 86 (26.2)	190	2	1.0	5	24071

<u>D.D.</u> <u>Hole #</u>	<u>Location</u>	<u>Azimuth</u>	<u>Dip</u>	<u>Depth ft(m)</u>	<u>Cu</u> <u>ppm</u>	<u>Mo</u> <u>ppm</u>	<u>Ag</u> <u>ppm</u>	<u>Au</u> <u>ppb</u>	<u>Assay #</u>
RC81-2	1080N 1090W	300°	-60	86(26.2)-96(29.3)	480	1	3.0	10	24072
				96(29.3)-102(31.1)	360	2	2.6	15	24073
				102(31.1)-110(33.5)	875	2	4.9	15	24074
				110(33.5)-116(35.4)	415	3	2.3	15	24075
				116(35.4)-126(38.4)	330	1	2.0	15	24076
				126(38.4)-131(39.9)	300	2	3.0	15	24077
				131(39.9)-143(43.6)	225	4	3.6	15	24078
				143(43.6)-150(45.7)	74	1	0.8	15	24079
				150(45.7)-160(48.8)	135	2	1.2	15	24080
				160(48.8)-166(50.6)	35	4	0.2	15	24081
				166(50.6)-176(53.6)	6	4	0.2	15	24082
				176(53.6)-186(56.7)	5	3	0.2	15	24083
				186(56.7)-196(59.7)	109	3	0.8	15	24084
				196(59.7)-206(62.8)	578	2	2.9	10	24085
				206(62.8)-216(65.8)	515	3	2.8	15	24086
				216(65.8)-224(68.4)	310	3	1.9	15	24087
				224(68.4)-236(71.9)	342	4	1.2	15	24088
				236(71.9)-246(75.0)	317	10	0.9	10	24089
				246(75.0)-256(78.0)	428	6	1.1	10	24090
				256(78.0)-264(80.5)	416	4	1.3	10	24091
				264(80.5)-274(83.5)	204	6	0.7	5	24092
				274(83.5)-282(86.0)	222	4	0.6	15	24093
				282(86.0)-291(88.7)	144	5	0.4	15	24094
				291(88.7)-294(89.6)	131	6	0.3	5	24095
				294(89.6)-304(92.7)	252	10	0.6	5	24096
				304(92.7)-313(95.4)	148	6	0.4	5	24097
				313(95.4)-323(98.5)	137	6	0.5	10	24098
				323(98.5)-333(101.5)	215	5	0.6	10	24099
				333(101.5)-341(103.9)	191	4	1.6	10	24100
				341(103.9)-348(106.1)	236	10	0.6	15	24101
				348(106.1)-355(108.5)	282	8	0.8	5	24102
				355(108.5)-366(111.6)	394	8	1.3	15	24103
				366(111.6)-376(114.6)	557	6	1.9	5	24104
				376(114.6)-383(116.7)	405	5	1.8	15	24105
				383(116.7)-390(118.9)	354	2	0.5	5	24106
				390(118.9)-400(121.9)	254	55	0.7	15	24107
				400(121.9)-411(125.6)	112	58	1.2	10	24108

<u>D.D.</u> <u>Hole #</u>	<u>Location</u>	<u>Azimuth</u>	<u>Dip</u>	<u>Depth ft(m)</u>	<u>Cu</u> <u>ppm</u>	<u>Mo</u> <u>ppm</u>	<u>Ag</u> <u>ppm</u>	<u>Au</u> <u>ppb</u>	<u>Assay #</u>				
RC81-2	1080N 1090W	300°	-60	411(125.6)-419(127.7)	113	85	1.1	5	24109				
				419(127.7)-429(130.8)	390	52	1.2	5	24110				
				429(130.8)-439(133.8)	162	44	1.0	10	24111				
				439(133.8)-450(137.2)	88	48	0.9	5	24112				
				450(137.2)-460(140.2)	237	50	0.8	15	24113				
				460(140.2)-470(143.3)	118	76	1.1	10	24114				
				470(143.3)-480(146.3)	180	24	0.9	10	24115				
				480(146.3)-490(149.4)	105	6	0.4	10	24116				
				490(149.4)-500(152.4)	72	18	0.3	25	24117				
				RC81-3	1004N 1086W	020°	-60	41(12.5)-49(14.9)	250	560	9.4	40	24118
								49(14.9)-57(17.4)	900	215	36.0	220	24119
								57(17.4)-69(21.1)	1650	160	3.8	20	24120
								69(21.1)-78(23.8)	1350	480	1.3	40	24121
								78(23.8)-83(25.3)	750	165	2.4	20	24122
83(25.3)-92(28.0)	375	245	2.0					80	24123				
92(28.0)-102(31.1)	370	100	3.7					340	24124				
102(31.1)-112(34.1)	320	95	3.5					430	24125				
112(34.1)-125(38.1)	450	89	1.6					30	24126				
125(38.1)-136(41.5)	430	53	0.9					120	24127				
136(41.5)-146(44.5)	360	35	0.5					20	24128				
146(44.5)-156(47.5)	600	135	0.6					40	24129				
156(47.5)-166(50.6)	1300	36	2.0					40	24130				
166(50.6)-176(53.6)	1000	130	1.3					70	24131				
176(53.6)-186(56.7)	750	89	1.2	60	24132								
186(56.7)-196(59.7)	750	45	0.8	40	24133								
196(59.7)-206(62.8)	550	47	0.4	30	24134								
209(62.8)-219(66.8)	650	120	1.1	20	24135								
219(66.8)-229(69.8)	750	110	1.0	50	24136								
230(69.8)-240(73.2)	700	215	2.3	80	24137								
240(73.2)-250(76.2)	1000	88	1.4	40	24138								
250(76.2)-260(79.2)	650	120	0.9	40	24139								
261(79.2)-271(82.6)	550	120	0.7	20	24140								
271(82.6)-282(85.9)	600	157	0.8	20	24141								
282(85.9)-292(89.0)	750	74	0.9	60	24142								
292(89.0)-302(92.1)	540	81	0.5	50	24143								
302(92.1)-310(94.5)	470	140	0.5	30	24144								

<u>D.D.</u> <u>Hole #</u>	<u>Location</u>	<u>Azimuth</u>	<u>Dip</u>	<u>Depth</u> <u>ft(m)</u>	<u>Cu</u> <u>ppm</u>	<u>Mo</u> <u>ppm</u>	<u>Ag</u> <u>ppm</u>	<u>Au</u> <u>ppb</u>	<u>Assay #</u>
RC81-3	1004N 1086W	020°	-60	310(94.5)-320(97.5)	600	160	0.7	60	24145
				320(97.5)-329(100.2)	465	120	0.6	20	24146
				329(100.2)-339(103.3)	490	91	0.1	60	24147
				339(103.3)-347(105.8)	270	220	0.2	20	24148
				347(105.8)-358(109.1)	350	390	0.1	40	24149
				358(109.1)-370(112.8)	410	150	0.5	70	24150
				370(112.8)-382(116.4)	800	100	1.0	70	24201
				382(116.4)-393(119.8)	950	165	1.5	100	24202
				393(119.8)-403(122.8)	1000	175	2.8	10	24203
				403(122.8)-408(124.3)	900	540	2.6	40	24204
				466(142.0)-476(145.1)	350	200	0.7	20	24205
				497(151.5)-509(155.1)	900	270	2.3	66	24206
				509(155.1)-519(158.2)	1200	360	3.0	666	24207
				519(158.2)-529(161.3)	900	160	1.6	66	24208
				529(161.3)-539(164.4)	900	240	1.6	L33	24209
				539(164.4)-549(167.5)	900	420	1.6	1210	24210
				549(167.5)-559(170.6)	700	120	1.6	66	24211
559(170.6)-569(173.7)	500	200	1.6	66	24212				
569(173.7)-579(176.8)	800	260	1.6	66	24213				
579(176.8)-589(179.5)	400	90	1.6	66	24214				
589(179.5)-599(182.6)	500	240	1.6	L33	24215				
599(182.6)-609(185.6)	400	660	1.6	66	24216				
609(185.6)-619(188.7)	900	520	2.0	66	24217				
619(188.7)-629(191.7)	600	360	1.6	66	24218				
629(191.7)-639(194.7)	600	80	1.6	66	24219				
843(256.9)-853(260.0)	400	L10		L100	24220				
RC81-4	1004N 1086W	197°	-60	36(11.0)-46(14.0)	100	170		L100	24221
				46(14.0)-56(17.1)	100	240		100	24222
				56(17.1)-66(20.1)	400	150		L100	24223
				66(20.1)-76(23.2)	200	110		L100	24224
				76(23.2)-86(26.2)	100	70		L100	24225
				86(26.2)-96(29.3)	300	130		L100	24226
				96(29.3)-106(32.3)	300	70		L100	24227
				106(32.3)-116(35.4)	400	80		L100	24228
				116(35.4)-126(38.4)	400	140		L100	24229
				-126(38.4)-135(41.1)	300	210		L100	24230

<u>D.D.</u>	<u>Hole #</u>	<u>Location</u>	<u>Azimuth</u>	<u>Dip</u>	<u>Depth ft (m)</u>	<u>Cu ppm</u>	<u>Mo ppm</u>	<u>Ag ppm</u>	<u>Au ppb</u>	<u>Assay #</u>
	RC81-4	1004N 1086W	197°	-60	135 (41.1) - 145 (44.2)	800	1000		L100	24231
					145 (44.2) - 155 (47.2)	800	170		L100	24232
					155 (47.2) - 165 (50.3)	900	210		L100	24233
					165 (50.3) - 175 (53.3)	500	170		L100	24234
					175 (53.3) - 185 (56.4)	300	180		L100	24235
					185 (56.4) - 195 (59.4)	300	90		L100	24236
					195 (59.4) - 205 (62.5)	600	180		L100	24237
					205 (62.5) - 215 (65.5)	800	210		L100	24238
					225 (68.6) - 235 (71.6)	500	100		L100	24239
					245 (74.7) - 255 (77.7)	400	140		133	24240
					265 (80.7) - 275 (83.8)	500	90		233	24241
					281 (85.6) - 285 (86.9)	300	160		100	24242
					285 (86.9) - 295 (89.9)	300	60		L100	24243
					305 (92.9) - 315 (96.0)	400	290		166	24244
					325 (99.1) - 335 (103.1)	600	40		L100	24245
					345 (105.2) - 355 (108.2)	300	80		L100	24246
					365 (111.3) - 375 (114.3)	400	140		L100	24247
					385 (117.3) - 395 (120.4)	500	200		133	24248
					405 (123.4) - 415 (126.5)	300	230		L100	24249
					425 (129.5) - 435 (132.6)	300	120		L100	24250
					445 (135.6) - 455 (138.6)	400	130		L100	24451
					465 (141.7) - 475 (144.8)	500	100		L100	24452
					485 (147.8) - 492 (149.9)	400	110		L100	24453
					492 (149.9) - 501 (152.7)	1400	110		200	24454
					501 (152.7) - 505 (153.9)	300	80		L100	24455
	RC81-5	1016N 1066W	030°	-60	21 (6.4) - 31 (9.5)	300	80		L100	24456
					31 (9.5) - 41 (12.5)	500	190		L100	24457
					41 (12.5) - 51 (15.5)	700	70		L100	24458
					51 (15.5) - 61 (18.6)	600	120		L100	24459
					61 (18.6) - 71 (21.6)	800	250		L100	24460
					71 (21.6) - 81 (24.7)	600	90		333	24461
					81 (24.7) - 91 (27.7)	500	150		L100	24462
					91 (27.7) - 101 (30.8)	400	80		L100	24463
					101 (30.8) - 111 (33.8)	500	120		L100	24464
					111 (33.8) - 121 (36.9)	1600	300		100	24465
					121 (36.9) - 131 (39.9)	300	330		L100	24466

<u>Hole #</u>	<u>Location</u>	<u>Azimuth</u>	<u>Dip</u>	<u>Depth ft (m)</u>	<u>Cu ppm</u>	<u>Mo ppm</u>	<u>Ag ppm</u>	<u>Au ppb</u>	<u>Assay #</u>
RC81-5	1016N 1066W	030°	-60	131(39.9) -141(42.9)	300	130		L100	24467
				141(42.9) -151(46.0)	500	170		L100	24468
				151(46.0) -161(49.1)	300	70		L100	24469
				161(49.1) -171(52.1)	400	50		L100	24470
				171(52.1) -181(55.2)	500	340		L100	24471
				181(55.2) -191(58.2)	400	170		L100	24472
				191(58.2) -201(61.3)	400	80		L100	24473
				201(61.3) -211(64.3)	600	190		L100	24474
				211(64.3) -221(67.4)	500	170		L100	24475
				221(67.4) -230(70.1)	500	80		L100	24476
				230(70.1) -240(73.3)	700	130		L100	24477
				240(73.3) -250(76.2)	600	230		L100	24478
				260(79.2) -270(82.3)	500	220		L100	24479
				270(82.3) -280(85.3)	600	240		L100	24480
				280(85.3) -290(88.4)	600	130		L100	24481
				300(91.4) -310(94.4)	500	110		L100	24482
				320(97.5) -330(100.6)	500	190		L100	24483
				330(100.6) -340(103.6)	700	140		L100	24484
				340(103.6) -350(106.7)	700	110		L100	24485
				350(106.7) -360(109.7)	600	70		333	24486
				360(109.7) -370(112.8)	1500	240		2066	24487
				370(112.8) -380(116.1)	800	90		1266	24488
				380(116.1) -390(118.8)	700	140		L100	24489
				400(121.9) -410(124.9)	600	90		L100	24490
				420(128.0) -430(131.1)	700	200		L100	24491
				440(134.1) -450(137.2)	600	320		L100	24492
				460(140.2) -470(143.3)	700	140		L100	24493
				480(146.3) -490(149.4)	700	120		L100	24494
				490(149.4) -500(152.4)	800	120		L100	24495

<u>D.D.</u> <u>Hole #</u>	<u>Location</u>	<u>Azimuth</u>	<u>Dip</u>	<u>Depth ft(m)</u>	<u>Cu</u> <u>ppm</u>	<u>Mo</u> <u>ppm</u>	<u>Ag</u> <u>ppm</u>	<u>Au</u> <u>ppb</u>	<u>Assay #</u>
RC81-6	1010N 1110W	008°	-54°	64(19.5)	270	38		380	24496
				74(22.6)	410	51		40	24497
				84(25.6)	860	79		160	24498
				94(28.7)	550	32		40	24499
				104(31.7)	430	9		40	24500
				114(34.7)	490	66		20	24251
				134(40.8)	475	33		60	24252
				154(46.9)	355	49		30	24253
				170(51.8)	480	76		60	24254
				180(54.9)	460	57		40	24255
				190(57.9)	410	31		40	24256
				200(60.9)	350	92		10	24257
				210(64.0)	320	110		10	24258
				220(67.1)	260	110		L10	24259
				242(73.8)	160	72		10	24260
				252(76.8)	310	71		40	24261
				262(79.9)	415	130		10	24262
				272(82.9)	325	165		40	24263
				282(85.9)	350	170		30	24264
				292(89.0)	500	190		20	24265
				302(92.0)	520	430		L10	24266
				312(95.1)	560	240		20	24267
				322(98.1)	330	165		L10	24268
				332(101.2)	360	360		L10	24269
				342(104.2)	350	120		L10	24270
				352(107.3)	700	230		10	24271
				362(110.3)	790	370		16	24272
				372(113.4)	1100	180		330	24273
				382(116.4)	420	370		10	24274
				392(119.5)	480	190		10	24275
				402(122.5)	230	110		L10	24276
				412(125.6)	350	135		L10	24277
				422(128.6)	280	120		L10	24278
				432(131.7)	290	400		L10	24279
				442(134.7)	300	165		10	24280
				462(140.8)	280	105		L10	24281
				490(149.4)	400	310		20	24295
				500(152.4)	220	130		L10	24296
				540(164.6)	310	80		10	24297

<u>D.D. Hole #</u>	<u>Location</u>	<u>Azimuth</u>	<u>Dip</u>	<u>Depth ft(m)</u>	<u>Cu ppm</u>	<u>Mo ppm</u>	<u>Ag ppm</u>	<u>Au ppb</u>	<u>Assay #</u>
RC81-7	1010N 1110W	188°	-55°	33(10.1)-43(13.1)	310	72		80	24282
				43(13.1)-53(16.2)	130	34	0.5	10	24283
				53(16.2)-63(19.2)	500	33	2.0	10	24284
				63(19.2)-73(22.3)	680	82	2.5	40	24285
				73(22.3)-83(25.3)	340	91	1.8	20	24286
				83(25.3)-93(28.3)	580	64	2.5	30	24287
				93(28.3)-103(31.4)	480	42	1.5	30	24288
				103(31.4)-113(34.4)	600	38	1.6	50	24289
				113(34.4)-123(37.5)	700	77	3.3	60	24290
				123(37.5)-133(40.5)	420	430	0.9	20	24291
				133(40.5)-143(43.6)	530	44	0.5	L10	24292
				143(43.6)-153(46.6)	580	35	0.6	L10	24293
				163(49.7)-173(52.7)	600	34		20	24294
				183(55.8)-193(58.8)	350	50		40	24298
				213(64.9)-223(67.9)	155	53		10	24299
				223(67.9)-233(71.0)	470	52		50	24300
				233(71.0)-245(74.7)	560	31		L10	24301
				273(83.2)-283(86.3)	410	60		20	24302
				283(86.3)-293(89.3)	610	90		40	24303
				304(92.7)-314(95.7)	440	26		40	24343
				334(101.8)-344(104.9)	480	38		30	24344
				364(110.9)-374(113.9)	520	58		30	24345
				394(120.1)-404(123.1)	800	61		20	24346
				426(129.8)-436(132.9)	520	130		L10	24347

APPENDIX IV

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REFERENCES

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APPENDIX V

DRILL LOGS

and

ROCK TYPE ABBREVIATIONS

ROCK TYPE ABBREVIATIONS FOR DRILL LOGS

A	andesite
AD π	andesite-dacite porphyry
B	granite breccia
Dc Bxa	dacite breccia
F	felsite
FP	felsite porphyry or feldspar porphyry
Gd	granodiorite
QM	quartz monzonite
QM Bxa	quartz monzonite breccia
RD	rhyolite-dacite
T	rhyolite tuff

COMPOSITE DRILL LOG

CORE SIZE :

SCALE :

PROJECT :

HOLE No. **RC 81-1**

CASING COLLAR ELEV.:

GROUND ELEV.:

DATE STARTED :

PAGE No. **2** OF **13**

COORDINATES :

N

E.

DATE FINISHED :

REF. TO CLAIM CORNER :

INCLINATION :

AZIMUTH :

TOTAL DEPTH :

m

LOGGED BY **T.M. ELLIOTT**

DEPTH (M)	ALTERATION					FRACTURING	MINERALS	GEOLOGY	COMMENTS :	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS			
	Q	K	Sc	Ch	Ep													Py	Si		
15	/	/						P ₁ T	Fresh chyalitic tuff.												
16	/	/						P ₁ A			16.5 m	87	1/4								
17	/	/						P ₁ A	17.0-18.0m = hairline py. v. lts common				1/4						24007C		
18	/	/						P ₁ A				100	1/2								
19	/	/						P ₁ (SI)	19.0-20.0m = medium gray silica - flooding				1/4								
20	/	/						P ₁ T	20.0-23.2m = strong zone of fracturing and shearing.				1/2						24008C		
21	/	/						P ₁ T	20.6m = minor malachite on fract.				1/4								
22	/	/						P ₁ A			22.3 m		1/2								
23	/	/						P ₁ A					1/4						24009C		
24	/	/						P ₁ A	Abundant hairline pyritic fractures.				89	1/4							
25	/	/						P ₁ A					1/2								
26	/	/						P ₁ A	P ₀ ≈ P ₁		25.3 m		1/2						24010C		
27	/	/						P ₁ A				100	1								
28	/	/						P ₁ T					1/2								
29	/	/						P ₁ A	Weak silicification along fractures		28.4 m		1/2						24011C		

COMPOSITE DRILL LOG

CORE SIZE : SCALE : PROJ : HOLE No. RC: 801
 CASING COLL. ELEV.: GROUND ELEV.: DATE STARTED : PAGE No. 6 OF 13
 COORDINATES : N. E. DATE FINISHED : REF. TO CLAIM CORNER :
 INCLINATION : AZIMUTH : TOTAL DEPTH : m LOGGED BY T.M. ELLIOTT

DEPTH (M)	ALTERATION Q, Vn Sericite Chlorite Calcite Py Sphalerite	FRACTURING	MINERALS	GEOLOGY	COMMENTS :	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS	
75															
76					76.1m = 1cm white carbonate vein 76.7-76.8m = fault gouge			100	1/10					240	26C
77					77.5m = 2-3% biotite (dissem.) 77.8m = 5mm-1.5cm white carb. vn. (calcite)				1/10						
78								78.0m	1/4					240	27C
79								100	1/10						
80									1/10						
81								81.1m	1/10					240	28C
82									100						
83									1/10						
84								84.1m	1/10					240	29C
85					85.2m = 1cm x 1cm patch of black chlorite? Crys. micac sp.				100						
86									1/10						
87					86.8m = 2mm light brown carbonate vein.			87.2m	1/10					240	30C
88									1/10						
89					89.0-89.7m = strong fracturing Cpy associated with blebs of talc maline.			98	1/10						
90									1/4	0.1					

CORE SIZE :
 CASING COLLAR ELEV :
 COORDINATES : N. E.
 INCLINATION :

SCALE :
 GROUND ELEV. :
 AZIMUTH :

PROJ : RED CAP.
 DATE STARTED :
 DATE FINISHED :
 TOTAL DEPTH : m

HOLE No. RC 81-1
 PAGE No. 7 OF 130
 REF. TO CLAIM CORNER :
 LOGGED BY T.M. ELLIOTT

DEPTH (M)	ALTERATION				FRACTURING	MINERALS	GEOLOGY	COMMENTS :	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS	
	Q. Vn.	Sericite	Chlorite	Carbonate													Py. Vn.	Tourmaline
90										90.2	m	1/10					240	31C
91								90.0 - 91.8 m = abundant (5%) blebs of tourmaline up to 1 cm across (radiating black crystals)			100	< 1/10						
92								92.2 m & 92.4 m = malachite on shear fract.				1/4						
93								93.2 m = breccia cty. abund. tourmaline ± py in the matrix		93.3	m	1/2					240	32C
94											100	1/4						
95												< 1/10						
96								95.6 - 95.7 m = strongly tourmalinized (in matrix) chyo. brss.		96.3	m	1/10					240	33C
97								96.4 m = minor cpy in toucm. vo.				100	1/4					
98								97.9 - 98.2 = altered feldspar (now green sericite) porphyry dyke. lower contact ca 45° to the core axis.				1/2	0.1 Cu					
99								98.6 - 98.8 m = altered dyke as above.		99.4	m	1	0.2 Cu	ASSAY			240	34C
100								98.5 m = 1 X 3 cm bleb of cpy				100	1/2	0.1 Cu.				
101								99 - 100 m = Assayable copper.				100	1/2	-				
102								101.0 m = 1 X 3 cm bleb of cpy				102.4	0.05 Cu					
103								100.8 - 103.4 m = altered (sericite) and mineralized Q-feldspar porphyry dyke.				102.4	1/2	ASSAY			240	35C
104								103.0 - 103.4 m = very strong cpy				100	1	0.2 Cu				

COMPOSITE DRILL LOG

CORE SIZE :
 CASING COLLAR ELEV :
 COORDINATES :
 INCLINATION :

SCALE :
 GROUND ELEV :
 N. E.
 AZIMUTH :

PROJ RED CAP
 DATE STARTED :
 DATE FINISHED :
 TOTAL DEPTH : m

HOLE No. RC 81-
 PAGE No. 8 OF 13
 REF. TO CLAIM CORNER :
 LOGGED BY T.M. ELLIOTT

DEPTH (m)	ALTERATION					FRACTURING	MINERALS	GEOLOGY	COMMENTS :	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (m)	ASSAYS	
	Q. Vn.	Sericite	Chlorite	Carbonate	P. Vn.														
105																			
106																		240	35C
107									107.0 - 107.4 m = abnd. gouge in fault zone Thin shears continue down to 108 m.		91		1/2						
108																			
109									Rusty rhyolitic tuff.				108.5 m					240	37C
110									108.0 m = beginning of rhyolitic bxa. 1-4 cm angular frags. of rhyolite in a silica-dolomite? (fizzes when scratched)		100		1/4						
111									matrix. Matrix and frags. approx 50%-50%				111.6 m						
112									Original rhyolite frags. are also altered to carbonate (ie. they are soft; H=3)				111.9 m					240	38C
113									111.9 - 112.4 m = strong pyrite. Dk. gray colour of pyritic fragments				113.0 m						
114									113.0 m = some frags. have remnant tourmaline?		100		1/10						
115									114.2 m = end of rhyolite bxa. (carb. altered)				114.6 m					240	39C
116									Back into rhyolite which has been pervasively carbonate-altered ± clay?				116.0 m						
117													117.0 m						
118									Strong carb. ^{clay?} alt. & veining. Locally strong cracking with Q. veining				117.7 m					240	40C
119													119.0 m						
120													120.0 m						

COMPOSITE DRILL LOG

CORE SIZE :
 CASING COLLAR ELEV :
 COORDINATES :
 INCLINATION :

SCALE :
 GROUND ELEV. :
 N. E.
 AZIMUTH :

PROJ **RED CAP**
 DATE STARTED :
 DATE FINISHED :
 TOTAL DEPTH : m

HOLE No. **RC 81-1**
 PAGE No. **11** OF **13**
 REF. TO CLAIM CORNER :
 LOGGED BY **T.M. ELLIOTT**

DEPTH (M)	ALTERATION					FRACTURING	MINERALS	GEOLOGY	COMMENTS :	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS	
	Q. Vn	Sericite	Chlorite	Carb.	Py. Vn.												Taucahalite	
150									150.4m = lower contact of dyke is ca 45° to the core axis. Pyrite up to 3% for 10-15 cm from the contact with rhyolite.		100	3/4						
151									150.8 = hairline Q volts. in MoS ₂		151.2m	1/2				240	51C	
152									151-152.3m = locally strong brown K-feldspar alteration		98	3/4						
153									152.3m = minor dissemin. spx		154.3m	3/4						
154									152-154m = silicified fractures			3/4				240	52C	
155									154.4-154.5m = more brown K-spar alt. 2.			1/2						
156									155.6m = taucahalite? alteration begins. Also wk. to moderate sericitization.		100	1/4						
157									Rhyolite ctgs. 7% taucahalite?		157.3m	1/4				240	53C	
158											100	1/4						
159									Ca. 10% taucahalite?		160.4m	1/4				250	54C	
160											100	3/4						
161											163.4m	1/2						
162											100	1/2						
163									Minor MoS ₂ along hairline Q. volt.			1/2				250	55C	
164											100	1/4						

COMPOSITE DRILL LOG

CORE SIZE **OVQ**
 CASING COLLAR ELEV.:
 COORDINATES :
 INCLINATION **-160°**

SCALE **1:100**
 GROUND ELEV.:
 AZIMUTH :

PROJ **RED CAP**
 DATE STARTED :
 DATE FINISHED :
 TOTAL DEPTH : **152.4 m = 500 ft.**

HOLE No. **RC 81-81-2**
 PAGE No. **1** OF **11**
 REF. TO CLAIM CORNER :
 LOGGED BY **T.M. ELLIOTT**

DEPTH (M)	ALTERATION				FRACTURING	MINERALS	GEOLOGY	COMMENTS:	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS			
	Q. Vn.	P. Vn.	St. white	Carb. Vn.																
0																				
1																				
2																				
3																				
4										3.4m	1/4						240	62C		
5										? 1/4										
6										5.5m	1/4						240	63C		
7										6.2m = fracture bleached fractures.	100	1/10								
8										7.2m	1/4									
9											1/4						240	64C		
10											99	1/4								
11											3/4									
12										11.0m							240	65C		
13											98	1 1/2								
14										13.1m	3/4						240	66C		
15											4									

Casing = No core.

Light gray rhyolite tuff. Locally grades into lapilli tuff. Characterized by disseminated pyrite.

6.2m = fracture bleached fractures.

Alteration is only local in patches. (Q-Ser-Py)

9.6 - 10.0 m = locally pervasive green phyllic alteration.


10.5 - 10.6 m = gray feldspar porphyry dyke. Contacts ea. 20° to the core axis.

11.0 - 11.4 m = zone of strong pyritization (3-4%) in both veinlets and as blebs.

12.3 - 12.4 m = large 10cm wide vein of calcite w. pyritic selvages.

13.4 - 13.6 m = vein of SI-Py-Gnt
Aspy = MoS₂ at approx. 45° to the core axis.

COMPOSITE DRILL LOG

CORE SIZE :  SCALE : PROJ :
 CASING COLL. ELEV.: GROUND ELEV.: DATE STARTED : HOLE No. RC: 8102
 COORDINATES : N. E. DATE FINISHED : PAGE No. 6 OF 1102
 INCLINATION : AZIMUTH : TOTAL DEPTH : m REF. TO CLAIM CORNER :
 LOGGED BY T.M. ELLIOTT

DEPTH (M)	ALTERATION					MINERALS	GEOLOGY	COMMENTS :	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS			
	Q.Vn.	P.Vn.	Sticite	Calc.Vn.	Chlorite												Silicification	FRACTURING	FRAC	FRAC
75																				
76																				
77																				
78																				
79																				
80																				
81																				
82																				
83																				
84																				
85																				
86																				
87																				
88																				
89																				
90																				

75-76 m - Rhyolitic tuff composed of 1mm angular lt. gray rock fragments

79-80 m - local strong streak of py & silica fractures

80.5 - 80.7 m = silicified brown andesite dyke

82.0 - 83.9 m = locally strong buff to cream-coloured argillie alteration of rhyolitic tuff.

86-87 m = stark veining of pyrite in hairline fractures

87.8 - 88.0 m = shear zone - buff to yellowish brown coloration. Several white calcite veins in shear and adjacent to it

COMPOSITE DRILL LOG

CORE SIZE : SCALE : PROJ : HOLE No. RC 8102
 CASING COLL. ELEV. : GROUND ELEV. : DATE STARTED : PAGE No. 8 OF 1102
 COORDINATES : N E. DATE FINISHED : REF. TO CLAIM CORNER :
 INCLINATION : AZIMUTH : TOTAL DEPTH : m LOGGED BY T.M. ELLIOTT

DEPTH (M)	ALTERATION Qz, Kfs, Sericite, Calc. Mn, Chlorite, Silicification	FRACTURING	MINERALS	GEOLOGY	COMMENTS :	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS	
105					105-106m = abundant disseminated pyrrhotite		105.1m	100 1/2							
106							106.1m	3/4						24	102 C
107							107.1m	100 1							
108					Rhyolitic tuff breccia		108.5m	1/2						24	103 C
109					109-110m = dull green and dull brown alteration along a myriad of fractures that have been reheated		109.5m	1							
110					110.7m = some hard taconite? veins		110.7m	95							
111							111.6m	1							
112							112.6m	3/4						24	104 C
113							113.6m	100							
114							114.6m	1							
115							115.6m	3/4						24	105 C
116					116m = abund. dissem. po ± py		116.7m	100							
117					117m - fracturing and alteration has diminished		117.7m	3/4						24	106 C
118					118.9m = Breccia composed of altered OMQ. Mn ± ? fragments up to 4cm across Bx a dk gray py and po-rich siliceous		118.9m	1						24	107 C
120							120.9m	90 1							

CORE SIZE :
 CASING COLL. ELEV. :
 COORDINATES :
 INCLINATION :


SCALE :
 GROUND ELEV. :
 N E
 AZIMUTH :

PROJ. :
 DATE :
 DATE FINISHED :
 TOTAL DEPTH : m


HOLE No. RC 81-³
 PAGE No. 9 OF 11
 REF. TO CLAIM CORNER :
 LOGGED BY T. M. F. L. IOTT


DEPTH (M)	ALTERATION					FRACTURING	MINERALS	GEOLOGY	COMMENTS :	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS			
	Qz	Py	Sch	Chl	Sil																
120									cement. Frags. are rounded.												
121									10M Bxa												
122									121.6 - 122.0m = strong chloritization giving rock a mottled texture.		121.6m	3/4					24	108	C		
123									122 - 123m = strong po-chl in matrix 123 - 124m = po-chl in matrix continues		123.7m	2									
124																					
125									Still abundant chlorite in matrix		125.6m	3/4									
126																	24	109	C		
127									10M Bxa 127 - 129m = epidote mixed with chl and po in the matrix.		127.7m	1/2									
128																					
129									129 - 130m = epidote along some fractures 1mm Qz in rounded and altered fragment		130.8m	1/4									
130																					
131									131 - 132m = 1-2% epidote in bsa matrix			1/2					24	111	C		
132																					
133									10M Bxa			3/4									
134									Fragments up to 10cm across 134.6m = epidote in matrix with chlorite.		134.9m	1/2					24	112	C		

COMPOSITE DRILL LOG

CORE SIZE : 
 CASING COLLAR ELEV :
 COORDINATES : N. E.
 INCLINATION :

SCALE :
 GROUND ELEV :
 AZIMUTH :

PROJ : 
 DATE STARTED :
 DATE FINISHED :
 TOTAL DEPTH : m

HOLE No. RC 81 
 PAGE No. 10 OF 11
 REF. TO CLAIM CORNER :
 LOGGED BY T.M. ELLIOTT

DEPTH (M)	ALTERATION					FRACTURING	MINERALS	GEOLOGY	COMMENTS :	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS			
	Q	Py	St	Ch	Ep																
135									M												
136									Bra			100		1 1/2							
137									137.0m = 3mm Q-Py-Mo in rhyolite fragment. (5cm across & rounded)		137.2m			1 1/2				24	113	C	
138									Po & Py in blebs up to 3cm x 1cm in matrix.					1							
139												100		1							
140									Chlorite content of rock now ca. 10% as alt. of biotite? in Q.M. and in matrix to fragments		140.2m			1 1/2							
141									Epidote in matrix and as hairline veinlets.					1 1/2							
142									141-142m - py >> po in matrix					100							
143									142m = minor MoS ₂ in hairline fract. in Q.M. fragment		143.3m			1							
144									143.9m - breccia is now a bre. of rhyolite fragments.					1							
145												100		1/2							
146														1/2							
147									Sulphide content has diminished to 1/2%		146.3m			1/2							
148												98		1/2							
149														1/2							
150											149.4m			1/2							

CORE SIZE : HQ

SCALE :

PROJ :

HOLE No. : 8

CASING COLL. ELEV.: 1375.5m

GROUND ELEV.: 1375m

DATE STARTED : 2/8/81

PAGE No. 6 OF

COORDINATES : 1010

N. -1080 E.

DATE FINISHED : 15/8/81

REF. TO CLAIM CORNER :

INCLINATION : -60°

AZIMUTH : 020°

TOTAL DEPTH : m

LOGGED BY B : HEMINGWAY

DEPTH (M)	ALTERATION					FRACTURING	MINERALS	GEOLOGY	COMMENTS:	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS				
	Calc. Vein	Fe Vein	K-Sp	Epithermal	Chlorite													DESCRIPTIVE GEOLOGY				
85									H - Homophane veins of alb G.O.													
86									- Homophane - Qtz Manginitic. Matrix much more fine grained than below Open Fractures. Silic veins irregular, blocky. Fracture density 4/10mm ² . Qtz vein slightly offsetting K-spn Venettes by about 1cm. Qtz vein Mo K-Spn Not mineralized. Also occurs at 84.7m Qtz vein offsetting K spn at an acute angle by 1cm. Qtz vein is Mo mineralized.		847-86.0 m	40%										
87									Qtz vein slightly offsetting K-spn Venettes by about 1cm. Qtz vein Mo K-Spn Not mineralized. Also occurs at 84.7m Qtz vein offsetting K spn at an acute angle by 1cm. Qtz vein is Mo mineralized.		86.0-87.5 m	95%										
88									occurs at 84.7m Qtz vein offsetting K spn at an acute angle by 1cm. Qtz vein is Mo mineralized.		87.5-88.7 m	75%										
89									K-Spn Not mineralized. K spn Venettes. 5cm wide. Appearencing at 87m. Mo on fracture planes at 88m. Again RRP < 5%. Fractures are generally mineralized.		88.7-89.6 m	100%										
90									K-Spn vein generally barren. At 88.1 some fractures show slickensides. Fair Py dissemination occur along fracture planes. Epidote on fracture surfaces. Qtz veins at 90m contain Mo, some Py on fracture surfaces with Py associated. Phenocrysts of matrix much smaller now and < 15%.		89.6-90.5 m	65%										
91									aligned to chlorite at 90m. Fracture density at 92m is 4/30mm ² . Mo spotty on surfaces. Venettes appear tight with well rich Mo on K-Spn venettes at 91.5m.		90.5-91.4 m	50%										
92									- Spotty at 92m Mo associated with Po. Rich severely. Also from 92.5 to 93m. Many Fractures contain spotty Mo, Py, Cu. Sulfide around many fractures and venettes starting at 95m. Much K-Spn occurring throughout. At 96.2m, rth contains many amorphous to coarse, filled with calcite. Py, Epid chlorite, K-Spn, Mo, Cu. Density 6/10mm ² .		91.4-92.9 m	50%										
93									Bluish green and venettes at 93.7m. At 93.2m K-Spn is 100% more regular and regular. Mo, py, mag, po occur with the. The sulfide around including a joint present.		92.9-93.3 m	50%										
94									At 94.0 Py occupies v. Eg. - almost aphanitic. Primary Qtz filling. Sulfide around venette at 93.8. Qtz vein at 40° contain Mo, py. 70 Qtz vein. Mo mineralized.		93.3-93.7 m	60%										
95											93.7-94.5 m	70%										
96											94.5-96.6 m	70%										
97											96.6-97.5 m	100%										
98											97.5-98.3 m	100%										
99											98.3-99.0 m	90%										
100											99.0-100.3 m	90%										

COMPOSITE, DRILL LOG

CORE SIZE : HQ
 CASING COL. ELEV. : 1375.5 m
 COORDINATES : 1010 N. -1080 E
 INCLINATION : -60°

SCALE :
 GROUND ELEV. : 1375 m
 AZIMUTH : 020°

PRO. :
 DATE STARTED : 2/8/81
 DATE FINISHED : 15/8/81
 TOTAL DEPTH : m

HOLE No. : 8
 PAGE No. 11 OF
 REF. TO CLAIM CORNER :
 LOGGED BY B. : HEMINGWAY

DEPTH (m)	ALTERATION					FRACTURING	MINERALS	GEOLOGY	COMMENTS: B Δ - Breccia fragments of various granitic rocks	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (m)	ASSAYS			
	Clay	P. Vein	P. Spm	Epither	Chlorite													SiO ₂			
160	C																				
161	C																				
162	C																				
163	C																				
164	C																				
165	C																				
166	C																				
167	C																				
168	C																				
169	C																				
170	C																				
171	C																				
172	C																				
173	C																				
174	C																				
175	C																				

DESCRIPTIVE GEOLOGY

Light green epidote-plagioclase porphyry felicit
 - Contact granite dyke - diffused fracturing anastomosing
 but tight, better contact blechy and irregular
 (Granite 20% K-Spm, 5% surface, 50% plagioclase
 20-25% Qtz. No disseminated and within fractures
 also contain chlorite, pyrite - Mo in Qtz scatters at
 412° - 20° cross-cut appears later and NO mineral Mo
 - 167 m. K-Spm Granite Contact diffused but noticeable
 has an coarse grained as at 161 F.D. 4/10mm²
 - Chlorite at 167.2
 Contact assumed at Chert margin feldspar mottled green
 - Rock below contact shows slight fracture, very
 fine grained with blocks of feldspar - granite
 G.D. felsite - Angular fragments, some show
 contacts with each other, K-Spm + Qtz being
 every 1 meter - Granite breccia patches over
 sandy matrix - possibly reworked with K-Spm - Qtz.
 visible - banded and anastomosing F.D. 1/20mm².
 Mo disseminated and in scatters at 163.4 m. Possible
 brecciated, that has been intruded and broken by
 K-Spm and Plagioclase Porphyry Felicit
 at 166 m. K-spm Selva around fractures - anastomosing.
 Qtz veins at preferred angle 40-50° contain Ca, Mo
 by the 167 m. At 167.2 m K-Spm Dyke - intrudes
 has million of G.D. Contact zone - irregular Pyrite
 contact of sections <1% Blended zone at 171 m. old
 R.R. low section is roughly 75%
 Qtz scatters at right angles to core axis contain Mo.
 Selva over some fractures at 174.8

DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (m)	ASSAYS
160.6 - 163.4 m.	95%						
163.4 - 165.5 m.	100%			TS16			
165.5 - 166.7 m.	85%						
166.7 - 167.0 m.	100%						
167.0 - 169.2 m.	100%						
169.2 - 171.3 m.	100%			TS17			
171.3 - 174.0 m.	90%						
174.0 - 175.0 m.	90%						

CORE SIZE : 100 SCALE : PROJ : HOLE No. : 8
 CASING COLL. ELEV : 1375.5 m. GROUND ELEV. : 1375 m. DATE : 2/8/81 PAGE No. 14 OF 1
 COORDINATES : 1010 N 1040 E. DATE FINISHED : 12/8/81 REF. TO CLAIM CORNER :
 INCLINATION : -60° AZIMUTH : 020° TOTAL DEPTH : m LOGGED BY B : HEMINGWAY

DEPTH (M)	ALTERATION					FRACTURING	MINERALS	GEOLOGY	COMMENTS :	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS			
	Calc.	F. V. Calc.	K-Span	Fe-Sulphide	Chloride																
205									Qtz, Ironoxide or Plagioc - Fe-bearing porphyry felsite as before. (containing and massive)		204.8 - 207.5 m.	100%									
206								QM													
207									- usually scintils contain chlorite scales and pyrite. Fractures occur as breaks and cont. relatively little material		207.5 - 208.5 m.	46%									
208									- At 207.5 depth of Calcite, pyrite - Occasional magnetite scintils		208.5 - 211.5 m.	95%									
209									- Epidote patches in matrix		211.5 - 213.1 m.	100%									
210											213.1 - 216.4 m.	95%									
211								QM													
212											216.4 - 218.2 m.	100%									
213									- Qtz, Plagioc and matrix 3m scale - possible dilution realized		218.2 - 219.2 m.	100%									
214																					
215									- Chlorite fractures and scintils occur every 0.3m												
216									Alteration confined to surface of the scintils. Some matrix contain chlorite, calcite, pyrite												
217								QM	RDN show out section in 90°												

COMPOSITE DRILL LOG

CORE SIZE : \varnothing 102
 CASING COLLAR ELEV : 1375m
 COORDINATES : 1.10 N. E.
 INCLINATION : -60°

SCALE : 1 : 100
 GROUND ELEV : 1375m
 AZIMUTH : 197°

PRO. : RED CAP
 DATE STARTED : 15/12/81
 DATE FINISHED : 19/8/81
 TOTAL DEPTH : 506' \approx 154m

HOLE No. 81-4 : \varnothing 81-4
 PAGE No. 1 OF 10
 REF. TO CLAIM CORNER :
 LOGGED BY : F. HEMMINGWAY

DEPTH (M)	ALTERATION					MINERALS	GEOLOGY	COMMENTS :	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS
	Calcite	Pyrite	K-Sp	Chlorite	Silica												
10																	
11								- Anomalous fracture (right with pyrite)									
12								- Broken (Bkn) weak ss.									
13								- Qtz vein, horizontal COARSE TALUS									
14								- Bkn weak Gneiss									
15								- Anomalous fracture, CASING FID: 4/20mm									
16								Light grey, mafic pheno-cryst altered to albite, 15% Mafic content, 50% clay, Qtz < 40%. Grain size < 2mm		15.2-17.1m	20%						
17								Qtz veins, one discontinuous, due to slippage on fracture plane. FID 6/20mm ² . ALL fractures are partly open with gravel or grit fillings. Core recovery moderate.		17.1-18.9m	65%						
18								RQD 40%. MANY AREAS ARE completely shattered to a rusty grit and gravel. Pyrite occurs as disseminations (sparse) and in some tight fractures.		18.9m-20.1m	95%						
19								It appears to be a Gneiss covered from 14.9m to 18.3m		20.1m-21.6m	71%						
20								Qtz veins at 18.6m anomalous and discontinuous. 3mm wide also 18.6m Qtz veins 50% - fractures show effect by lens, veined slightly pink - 1.7, 1.8, 1.9m		21.6m-22.5m	55%						
21								Pyrite occurs with albite in altered mafic pheno-cryst. Bkn core at 20.1-21.9; 22.3-23.0; 23.3-24.0m.		22.5m-23.3m	55%						
22								- Trace of Mo in Fracture - Qtz filled < 1mm at 19.1m at 20.1m Qtz vein blebbing - 15cm wide		23.3m-24.2m	45%						
23								- Core in silica section shaly bkn with rust and fragments filling fracture plane.		24.2m-25.1m	45%						

COMPOSITE DRILL LOG

CORE SIZE : NQ
 CASING COLLAR ELEV: 1375.5
 COORDINATES : 104 N E
 INCLINATION : -10"

SCALE :
 GROUND ELEV: 1375
 AZIMUTH : 197°

PROJ :
 DATE STARTED : 15/8/81
 DATE FINISHED : 19/8/81
 TOTAL DEPTH : m

HOLE No. 81-4:
 PAGE No. 2 OF 10
 REF. TO CLAIM CORNER :
 LOGGED BY HEDDINGWAY

DEPTH (M)	ALTERATION					FRACTURING	MINERALS	GEOLOGY	COMMENTS :	AVG. CORE RECY/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS			
	Calc	Pyrite	K-Sp	Epidote	Chlorite													Slite			
25									Granodiorite as before.												
26								Gd			24.7-26.2m	15%									
27																					
28									- Epidote disseminations with chlorite around vesicles		26.2-28.7m	95%									
29									RDP for section < 5%												
30									Almost all fracture faces extremely rusty and weathered.		28.7m-30.8m	85%									
31									No quartz veins present as well as <u>no</u> chlorite occurring as inclusions and disseminations		30.8m-32.3m	20%									
32									- Anastomosing fractures at 33.4m F.D. 6/20mm ²		32.3-33.8m	50%									
33									- Magnetite within pyrite vesicles		33.8m-35.4m	80%									
34									- Fracturing branching F.D. 3/20mm ²		35.4-36.6m	95%									
35									- Bkn		36.6m-38.4m	75%									
36									- Qtz veins broken (Bkn) with a little K-Sp - Spars Mo associated												
37									Rud. as fine grained and some chlorite clined.												
38									At 37m Py vesicles showing some propylite.												
39									At 37.4m sericite occurs with chlorite?												
40									irregular Q over at 38m.												
41									- Bkn												

T520

COMPOSITE DRILL LOG

CORE SIZE : SCALE : PROJ : HOLE No. 81-4
 CASING COLLAR ELEV : GROUND ELEV : DATE STARTED : PAGE No. 3 OF 10
 COORDINATES : N. E. DATE FINISHED : REF. TO CLAIM CORNER :
 INCLINATION : AZIMUTH : TOTAL DEPTH : m LOGGED BY B : HEMINGWAY

DEPTH (M)	ALTERATION					FRACTURING	MINERALS	GEOLOGY	COMMENTS: B D - Broken rock fragments; A - Intense silicification and alteration	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS			
	Calc	Pyrite	K-Spar	Epidote	Chlorite													ASSAYS	ASSAYS	ASSAYS	ASSAYS
40									Granodiorite as before.												
41									- Bkm, rusty fractures & weathering		40.2-41.1m	30%									
42									- Malachite & Covellite appear sparsely disseminated in matrix, Mo on fracture surface		41.1m-43.0m	85%									
43									- at 41.1 QTZ vein dense		41.1m-41.1m	85%									
44									At 41.7 Granodiorite completely altered to chlorite quartz epoxide, calcite. Relict texture broken somewhat distinguishable. Fractures completely healed with quartz. Many Qtz inclusions show Mo. K-Spar alteration confined		41.1m-46.9m	95%									
45									sparsely to patchy areas (K-Spar in patches among epoxide)		46.9m-47.9m	95%									
46									chlorite confined to phreatic K-Spar associated with veins (selving) Mo on sides of well rock next QTZ and veins. Offset vein at 43.0m with Mo (about 4cm of relict epoxide)		47.9m-48.9m	80%									
47									Mo veinlet about 2cm thick		48.9m-49.9m	50%									
48									43.5m vein Mo 30° 43.58 AT 43.5m		49.9m-50.6m	95%									
49									- Very little fracturing between 41.1 and 43.9m		50.6m-52.1m	80%									
50									No weathering features from 40.2 - 44.4m RQP Part 40.2m to 5%		52.1m-55.5m	100%									
51									Section is extreme phyllic alteration, relict texture almost indistinguishable from Granodiorite from 45.0m to 48.5m Aplastic granodiorite from 45.0m to 47.5m F.D at 47m 4/20m												
52									At 49.5m dense Qtz filling, Calcite Alteration at approx 49.0m												
53									Rock from 44.8m is getting more & more granular, granular to alter to a slight green chlorite color												
54									Unit could be a possible chlorite - phyllic sequence												
55									rather than a granodiorite of high grade (Microgranodiorite)												
56									and the contact could be the altered phyllic zone												
57									Basalt fracture from 44.0m - Still in weather Selving at 54.0m												

TS21

COMPOSITE DRILL LOG

CORE SIZE : \bigcirc : NCR
 CASING COLLAR ELEV :
 COORDINATES : N E
 INCLINATION : -60°

SCALE :
 GROUND ELEV :
 AZIMUTH : 197

PROJ : Red Cap
 DATE STARTED : 16/8/81
 DATE FINISHED : 19/8/81
 TOTAL DEPTH : m

HOLE No. 81-4 : \bigcirc
 PAGE No. 4 OF 10
 REF. TO CLAIM CORNER :
 LOGGED BY B : H F MINGUWAY

DEPTH (M)	ALTERATION					FRACTURING	MINERALS	GEOLOGY	COMMENTS: BA - Broken breccia - rock fragments.	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS			
	Calcite	Pyrite	A-S	Epithermal	Chlorite																
55																					
56																					
57																					
58																					
59																					
60																					
61																					
62																					
63																					
64																					
65																					
66																					
67																					
68																					

Microgranulite? - Quartz feldspar porphyry - grey-green
 - Fractures show selvage clay + pyrite filled
 Soil in weathering effects along fracture planes
 - Green clays and some fracture planes altered slightly
 to chlorite. Trace Mo on quartz veins - discontinuous
 Diffuse - Pyrite on fracture surfaces - Mo on veins
 - very coarse. F.O at 56.0m 3120mm
 very weak chlorite alteration
 Rock texture is shattered and resealed by fractures
 Fracture surfaces appear to be filled with sandy pyrite and
 clay. Qtz veins on the average are not more than 1cm
 thick
 - Epidote in veins - diffuse
 Bleached zone - rock aphanitic and by many pyrite
 and clay filled veins or fractures. Pyrite visible on some
 fractures
 - White veins of 60.5m show slickensides at 70° to
 the fracture face or roughly 80° to C/A
 - Drill holes 60.0 - 60.5m
 At 62.7m - Pyrite veins lower selvage of chlorite.
 Generally < 1mm thick.
 Rocks generally have a texture of mottled
 greens. It appears that fragments have deformed
 and rounded to sub-rounded edges. At 64.9m - fracture
 filled with 2mm of clay
 Many pyrite filled fractures have a ring feature due to
 weathering. At 65.8m pyrite veins show selvage of
 quartz. Possible Qtz veins with pyrite dissemination in
 veins. RRP after 60m weathering to 80% up to 71.7m
 Selvage of Qtz and on veins at 69.0m - 1cm thick.

DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS
55.5m - 57.6m	95%						
57.6m - 59.7m	100%						
59.7m - 61.9m	95%						
61.9m - 64.9m	0						
64.9m - 67.4m	55%						
67.4m - 70.4m	0						

COMPOSITE DRILL LOG

CORE SIZE: 11 ALI: 11
 SCALE: 1:100 PRO: 11 : Red Camp
 CASING COLLAR ELEV: 1275.2m GROUND ELEV: 1275m DATE STARTED: 26/8/81 HOLE No. 81-6 : 11
 COORDINATES: 1010 N - 1110 E DATE FINISHED: 30/8/81 PAGE No. 4 OF 11
 INCLINATION: -54° AZIMUTH: 000° TOTAL DEPTH: 167.6 m 600' REF. TO CLAIM CORNER:
 LOGGED BY: B. HEMMINGWAY

DEPTH (m)	ALTERATION				FRACTURING	MINERALS	GEOLOGY	COMMENTS:	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (m)	ASSAYS			
	Clay	Sulf	Ep	Ch																
60								Feldspar Perphyry Garnetiferous as before.												
61								Biotite with pyrite - disseminated												
62								Section 60-63.4m - Dual Strike Preferred Shearing Direction: about 50-70° & 20-40° (2 sets) ROD 01												
63								K-Spar Veinlet up to 3mm wide												
64								Sparse Epidote on fracture planes 5/20mm ² Fk												
65								Biotite partially altered to chlorite, pyrite.												
66								Qtz veins with K-Spar veinlets. Mo veins sparsely in K-Spar veinlets. Calcite veins in microveinlets w/epi K-Spar.												
67								- At 63.2m K-Spar veins plus - out matrix (looking)												
68								Some magnetite also occurs with K-Spar at 63.2.												
69								- 4cm wide Qtz veinlet (irregular at 35°) at 64.6												
70								- Gangue at 64.6. - Fault Zone - from 64.0m to 66.1												
71								Mo veins as disseminations and in Qtz veinlets (4cm wide)												
72								- Breking along weak fracture planes.												
73								- Epidote alteration confined to fractures. Some as disseminations. Chlorite completely altered Biotite to 71.0m												
74								- Chlorite alteration of Biotite not complete, occurs as disseminations from 71.0m												
75								- Fracture planes and breks occurs at 40-60° and 10-20° TS 38 - Fresh unaltered rock from 71.0-71.15												
76								- broad Mo on fracture surface at 70.1m (35°) 1cm wide with pyrite.												
77								Complete alteration of plagioclase to epidote (wavy at 70.1m to 71.0m). At 71.0-71.15 Fresh G.D. contains fine dissemination (sparsely) of Qtz and Biotite.												
78								Mo on Qtz veinlets 1cm wide at 73.5m (wall rock)												
79								- Calcite occurs on fracture planes - show slickensides												

TS 38

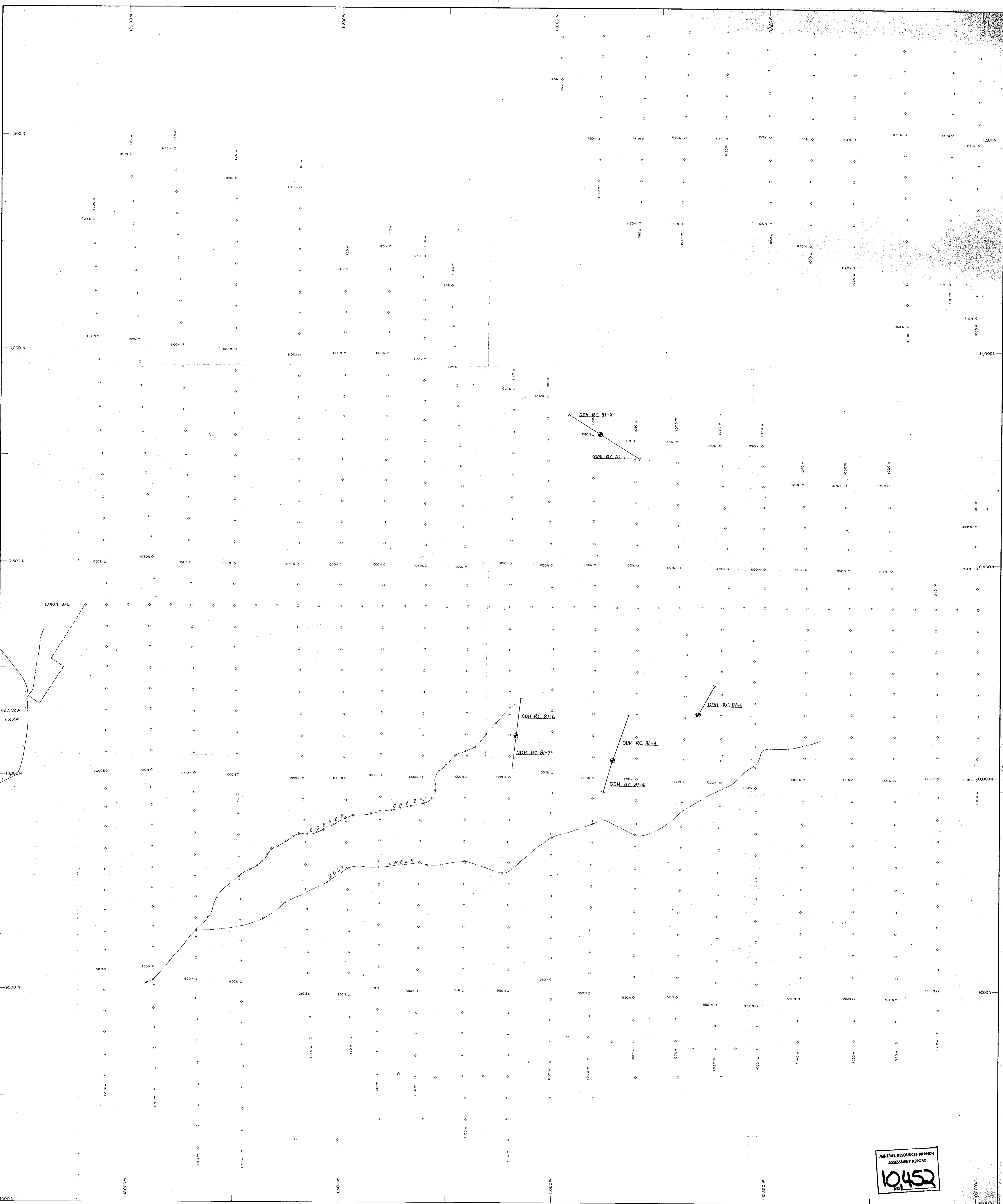
CORE SIZE: \bigcirc NQ
 CASING COLLAR ELEV.:
 COORDINATES: N E
 INCLINATION: -55°

SCALE: 1:100
 GROUND ELEV.:
 AZIMUTH: 188

PROJECT: REOCAP
 DATE STARTED: Sep 11/81
 DATE FINISHED: 4/81
 TOTAL DEPTH: 132.3 m

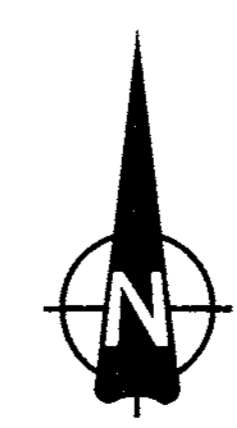
HOLE No.: 81-7
 PAGE No. 7 OF 9
 REF. TO CLAIM CORNER:
 LOGGED BY: P. HEMINGWAY

DEPTH (M)	ALTERATION					FRACTURING	MINERALS	GEOLOGY	COMMENTS: With return	AVG. CORE REC'Y/HOLE	DRILLING INTERVAL	% CORE RECOVERED	% SULPHIDES	ESTIMATED	SAMPLE No.	% SAMPLE RECOVERED	SAMPLE INTERVAL (M)	ASSAYS			
	Calc	Pyrite	Qtz	Chal	Other																
101								GD	Abund. brownish to Actinolite - chlorite, epidote Feldite. 11. The core fracture planes at 101. Rock extremely blk up RQD < 15% to 105.5		100.3										
102									Feldite, altered to clay from 100.2 to 100.9 m, Qtz, some epidote, chlorite - massive?		90.1										
103									- massive epidote and quartz at 102.5		101.5										
104									- Actinolite in patches and veins with chlorite, pyrite. (Pyrite cubes > 1mm size < 2mm)		95%										
105									- Qtz veins > 10mm wide with many hollow veins - Pyrite in associated with chlorite and epidote		102.6										
106									- Qtz veins < 2mm thick. F11 3/40mm ² RQD > 50%		80.1										
107									- Fracturing roughly between 30 and 50°.		104.2										
108									- Qtz veins 3. - wide at 40°. Large of pyrite grains		85										
109											105.5										
110									- Actinolite - No. 11. Qtz veins at 40°		75										
111									- K. Sp. fluorine with broken Qtz veins, with Qtz.		106.1										
112								GD	- Pyrite patches, Conductivity control at 112.8m		107.2										
113								AD	- Actinolite Epidote - chlorite altered A. D. veins with		80										
114									- Wall striae to 1mm - massive K. Sp. fluorine, 15mm wide at 114.5		109.2										
115									- Some chlorite		95										



MINERAL RESOURCES BRANCH
ASSESSMENT REPORT
10452
NCL

SYMBOLS
○ Sample location



OMNI RESOURCES INC
SOIL SURVEY GRID AND
DRILL HOLE LOCATION MAP
FIGURE 3

