

DIAMOND DRILLING, GEOCHEMISTRY, GEOPHYSICS  
On the ZAP Claims  
Liard Mining Division  
NTS 104P/13E, 14W

January, 1981 J. R. Wilson

Vancouver, B. C.

DIAMOND DRILLING, GEOCHEMISTRY, GEOPHYSICS  
ON THE  
ZAP CLAIMS

Liard M.D

N.T.S. 104P/13E, 14W

J. Wilson  
Jan, 1981

TABLE OF CONTENTS

	<u>Page</u>
SUMMARY .....	1
LOCATION .....	1
ACCESS .....	1
BACKGROUND .....	1
DIAMOND DRILLING .....	4
ANALYSIS OF DRILL CORE .....	6
SOIL GEOCHEMISTRY .....	8
GEOPHYSICS .....	8
ROAD CONSTRUCTION .....	8
CONCLUSIONS AND RECOMMENDATIONS .....	9
Figure 079-80-1 - Property Map .....	11
<u>APPENDICES</u>	
Appendix I Drill Logs (Hole A1-A5) .....	12
Appendix II Mineralogical Report (Thornhill Lab) .....	34

SUMMARY

From mid August to early October, 1980, (477m.) 1566 feet of diamond drilling tested parts of the Zap mineral claims. Despite nearly continuous mechanical and logistical problems some encouraging results were obtained. Minor Pb-Zn-Cu mineralization and lithologies similar to the Gataga camp were intersected. Sections of core were split and analysed geochemically.

The drill results are significant in confirming this region as having stratiform Pb-Zn-Ag-Cu potential. Further drilling is recommended on the Zap claims to better explore the pyritic argillite encountered in 1980 and to test other anomalies.

A later winter program is suggested, thus avoiding access problems.

Preparatory ground work in the 1981 summer should locate, survey, and mark important sites for the following winter's program.

LOCATION - In B.C. 48 km WSW from Watson Lake, Y.T.

ACCESS - By FNM constructed road leaving the One Ace Mountain forestry road.

BACKGROUND - Refer to " Geology, Geochemistry, and Geophysics of Zap claims " by P. Burns, July 4, 1980. for property work done prior to drilling.

The claims are believed to be in the transition zone from the Selwyn Basin to the Kechika Trough (probably an extension of the Basin). For this discussion, both areas will be termed the Selwyn Basin.

In 1980 R. Carne wrote (at the CIM Annual General Meeting):

" The majority of the zinc, lead and silver resources of western Canada are contained in stratiform mineral deposits hosted by Paleozoic clastic sedimentary rocks, commonly shales, within Selwyn Basin of Yukon N.W.T. and northern British Columbia. Over half of these deposits have been discovered within the past decade and the steady rate of discovery suggests that many more will be found.

Recent studies of these deposits, and of other well-known examples, such as the McArthur \*H.Y.C.) deposit in Australia, the Meggen and Rammelsberg Mines in Germany and the Sullivan Mine in British Columbia, have led to an emerging, but still controversial, model of ore deposition. Shale-hosted or "sedimentary exhalative" (sedex) deposits form in active tectonic environments from metalliferous geothermal brines that rise to the seafloor along deep-seated fault zones and precipitate as bedded sulphide deposits, usually accompanied by barite. Although their mineralogy and zoning can be similar to that of volcanogenic massive sulphide deposits, sedex deposits have only a weak genetic link with volcanism."

Carne (1980) described the four major clusters of sedex deposits : in Yukon - Anvil, Howards Pass and MacMillan Pass in B. C. - Gataga.

Lithologic characteristics of each camp is similar.

In 1980 D MacIntyre mapped and reported on the Gataga area (at the District 6 CIM meeting). The following descriptions are from his paper. The mineral district is over 180 km. long and contains six major occurrences.

"The Cirque is the largest known deposit with reserves in excess of 30 million tonnes grading 10 per cent combined lead-zinc and 45 grams per tonne silver. The host rocks for the Cirque and other deposits in the area are Middle to Upper Devonian carbonaceous black shales, siliceous argillites, and cherts (Gunsteel 'Formation').

The mineralized interval of the Devonian succession appears to be present throughout the basin of deposition, typically consisting of thin beds of nodular barite with interbedded pyrite laminae. This interval locally thickens and grades into lens-shaped bodies of massive bedded barite which may or may not contain sphalerite and galena. Minor zinc and lead mineralization is also found in elses of laminated pyrite which are spatially associated with the bedded barite deposits."

"The K showing is currently the main exploration target on the Cirque property. The showing consists of several small outcrops and a prominent white-weathering barite kill zone exposed on the northeast-facing slope of a northwest-trending ridge. Diamond drilling in this area has intersected a massive barite bed containing diffuse bands and interstitial blebs of pyrite, sphalerite, and galena. This horizon varies from less than 5 metres on surface to greater than 35 metres down dip to the southwest. Average grades of drill intersections from this horizon are in the range of 9 to 15 per cent combined lead-zinc with 50 to 70 grams per tonne silver. The zinc/zinc + lead ratio of the 1978 drill intersections varied from 0.72 to 0.77. The overall ratio for the reserves defined in the 1979 program is 0.77.

Very fine-grained sphalerite and trace amounts of galena also occur in bands of laminated fine-grained and massive coarse-grained pyrite directly overlying the main barite horizon. Assay results from this zone are extremely variable and range from 0.5 to 8 per cent combined lead-zinc. Anomalously high background concentrations of lead and zinc also occur in rocks immediately underlying the overlying the main deposit."

At the Driftpile Creek occurrence:

"The main massive sulphide horizons are characterized by very finely laminated pyrite which locally has soft sediment deformation and graded bedding."

Barite and sphalerite are present in variable amounts.

"What appears to be the same horizon was intersected in drill holes immediately south of Driftpile Creek but here finely laminated pyrite predominates and galena and varite are present in very minor amounts."

The Anvil Camp consists of seven potential orebodies along a 25 km. belt. According to Carne (1980) the "characteristic vertical and lateral zonation seen in all deposits is, in descending order:

- (i) baritic massive sulphide
- (ii) pyritic massive sulphide
- (iii) pyritic quartzite
- (iv) ribbon banded graphitic quartzite.

Chalcopyrite - pyrrhotite stringer zone are recognizable beneath some deposits."

## DIAMOND DRILLING

D. J. Drilling completed 1566 feet (477metres) of the 2000 feet contracted in five holes. FNM was not charged for the undrilled portion. A Longyear 17 A drill was used and BQ core was recovered. All holes were vertical and were sited to test a variety of geophysical and geochemical anomalies. Drilling was entirely on the Zap 1 claim.

A D 6, rented from Grant Stewart Construction, prepared drill roads and hauled the drill and sloop.

Figure 079-80-1 shows drill hole locations and roads.

Hole #A1 drilled one of two EM-16 conductors near the base of slope that paralleled the valley axis east of our campsite. A few modestly anomalous soil samples (252 ppm Zn, 3.0 ppm Cd, 1.0 ppm Ag) and a good silt value of 10 ppm Ag had been found nearby. Drilling encountered a vertical fault zone with high water pressure. Twelve metres of overburden were found but 67 metres of casing were used in the very broken and open fault. The hole was stopped at 107 metres. Barren quartzites and siltstones were the only lithologies found. Overall core recovery was about 30%. The EM conductor is probably the fault. The high water pressure and warm water temperature encountered in the fault suggests that anomalous metal values found in nearby spring sediments could be due to migration from a considerable distance.

Hole #A2 tested a southeast trending EM-16 reverse cross-over that paralleled a normal conductor 50 metres away. This strike direction corresponds to regional fold axes as mapped by P. Burns. No soil samples at the site showed elevated values but scattered highs occur through the area ( eg. to 410 ppm Zn, 4.3 ppm Cd, 1.2 ppm Ag ).

Twenty metres of casing was used and the hole went to 151 metres encountering pyritic quartzites, siltstones and pyritic, graphitic black shales explaining the EM anomaly. Quartz/calcite veinlets are common and some carry specks of galena. A single 8 cm.

long section of core contains a veinlet with sphalerite, pyrite, galena, and chalcopyrite. Examination of the sample at Thornhill also confirmed traces of arsenopyrite and pyrrhotite ( see attached report ). Bedding here was found to be nearly vertical. Nevertheless, drilling was allowed to continue because of favourable results. Future angle holes are needed to test across the stratigraphy at the best mineralized depths or at levels suggested by rock geochemistry. Drilling should especially be directed towards the nearby EM conductor to the southwest.

Hole #A3/A4 was collared on the western "limb" of a coincident EM-16 - soils anomaly. The feature is at least 700 metres in length, strikes roughly  $100^{\circ}$ , and is slightly sinuous indicating possible faulting or folding. Soils immediately downslope were 14 ppm Pb, 1.5 ppm Ag, 470 ppm Zn, 10 ppm Cd. Although better soils values lie to the east this site was drilled because the EM response was clearest.

Drilling was difficult in this sheared graphitic shale. Casing couldn't be driven into solid rock because of the tight graphite and the BW sub broke off in hole A3. The set up was moved a few metres and hole A4 began with N casing going to 24 m. B casing going to 37 m., and coring coming to a standstill at 40 metres. During this difficult stage (12 days) considerable time was spent repairing the drill, obviously not capable of handling the conditions.

Core recovered was black, graphitic, pyritic, sheared argillite/shale. One 3 cm breccia fragment held about 50% disseminated pyrite.

The graphitic rock explains the EM response but the associated geochemical values have yet to be accounted for. This hole can be regarded as partially successful since it encountered the lithology usually associated with Selwyn Basin stratiform deposits.



Hole #A5 was on the eastern end of the same feature tested by the previous hole. It was collared on a strong EM response and the best soils anomaly was immediatly downslope ( over 10,000 ppm Zn, 215 ppm Cd ).

Thirty metres of casing was used and the hole was stopped at 148 metres. All rock encountered was black argillite with graphite on fractures and disseminated pyrite throughout. Pyrite is also seen on fractures, as discrete blebs to 1 cm diameter, as irregular indistinct massive patches to 10 cm. and as laminated beds ( sometimes massive ) to 1 cm. thick. Rip-up soft sediment deformation is exhibited by one pyritic bed. An 8 metre section is interpreted as a possible vent with rounded clasts ot the top grading down to angular fragments and passing into solid rock. Veinlet stockworks of calcite are prominent throughout the core.

On this hole the lithology, bedded pyrite, and occasional specks of galena heighten interst and point to the necessity of more drilling.

A final drill set up was made on the road in the Zap 10 claim but freezing weather reduced the nearby water supply making drilling impossible. The site should be drilled at a future date. It consists of an EM-16 conductor and high soil values ( 1675 ppm Zn and 71 ppm Cd ).

All drilling was terminated at this point because of deteriorating water supplies, an inadequate drill for the difficult ground, small footage remaining in the contract ( and no charge for cancellation ), and a nearly impassible access road whose upgrading would be too expensive.

#### ANALYSIS OF DRILL CORE

Core from all holes except A1 was split and geochemically analysed, generally in alternate 5 foot lengths. Shorter sections were taken based on lithology or mineralogy. Sludge samples analysed were taken during drilling when core recovery was poor.

Ranges of values in drill core and in average sedimentary rocks are shown below in ppm.

<u>Hole #</u>	<u># of samples</u>	<u>Cu</u>	<u>Pb</u>	<u>Zn</u>	<u>Ag</u>	<u>Ba</u>
A2	49	4-110	2-90	4-1400	0.2-1.3	30-1090
A3/4	4	26-39	14-18	98-187	0.2-0.3	920-1350
A5	33	32-97	6-70	43-5800	0.2-7.0	490-1830

---

Average Ranges of geochemical values (Hawkes & Webb)

shale		30-150	20	50-300		300-600
Black shale		20-300	20-400	100-1000	5-50	450-700

---

According to Hawkes & Webb some of our Zn and Ba results are well above average. The high barium content in drill core is especially significant as barite is commonly associated with mineralization at the Anvil, MacPass, and Gataga camps. However it is not present at Howards Pass.

In hole A5 a chemical change occurs at about 320 feet (97.5 m ):

- a) Sb was found in all rock below this depth but not above it.
- b) Ag values change at this point. 18 samples from 30.5 to 96.0 metres had a mean of 1.9 ppm Ag (standard deviation = 1.0) and 15 samples from 97.5 to 147.8 metres had a mean Ag content of 3.6 ppm (standard deviation = 2.0)
- c) For the same top section Zinc's mean was 306 ppm (S.D. = 308) and for the bottom section Zinc's mean value was 1639 ppm (S. D. = 2045).

Although not obvious, the stronger silver values appear to correspond roughly with higher pyrite content in the form of massive pyrite patches. The lower values are in core containing a long brecciated section (based on structure this may be a vent).

No other prominent traits were recognized but a thorough manipulation of data should precede further groundwork.

### SOIL GEOCHEMISTRY

Regional soil sampling conducted by P. Burns (see earlier reference) located a concentration of good Cd and Ag values near our access road on Zap 10. I.L.Elliott reanalysed the samples for Pb and Zn. Results clarify a few multi element anomalies:

(a) two Pb zones with moderate Zn, Ag, Cd associated values are centered on line 0+00 at 8+00 S and 17+00 S (to 128 ppm Pb, 880 ppm Zn, 2.2 ppm Ag, 3.9 ppm Cd).

(b) Strong Zn and Cd results are on Line 5 E between 24+00 S and 28+00 S. (to 1675 ppm Zn, 71 ppm Cd, 3.5 ppm Ag, 13 ppm Pb).

### GEOPHYSICS

Steve Presunka was in the area to run a few lines of EM-16 on John Schussler's Donna claims. While here, he made traverses over the soils anomaly and located a nearby conductor (DDH A6-not drilled).

### ROAD CONSTRUCTION

The FNM road built in 1979 received considerable traffic during our program and consequently deteriorated quickly in the few wet sections. Culverts were installed to help drain spring waters. The road was thus improved for a week until very unusual late season rainfalls made the rest of the road nearly impassable. The cat, intended only for drill moves, had to be used to upgrade the road, adding a significant, unforeseen expense.

New roads were built to three drill sites and considerable time was spent building across unavoidable swamps. Grant Stewart construction supplied an efficient crew to do the roadwork.

K. H. Christensen and A. MacArthur started cleaning up debris from the 1979 road construction but spent more time on camp building, road repairs and construction of drill roads before leaving for personal reasons.

J. Hugi and J. Wilson later continued the clean up until snowfalls blanketed the slash, preventing safe, efficient work. Hazard abatement can be completed in 1981 while exploration ground work is in progress here.

#### CONCLUSIONS AND RECOMMENDATIONS

Diamond drilling was successful in locating minor mineralization and promising lithologies. However, mechanical and logistical problems thwarted completion of the drill contract. Several thousand feet of drilling are needed to complete primary exploration and to follow up the 1980 drill results.

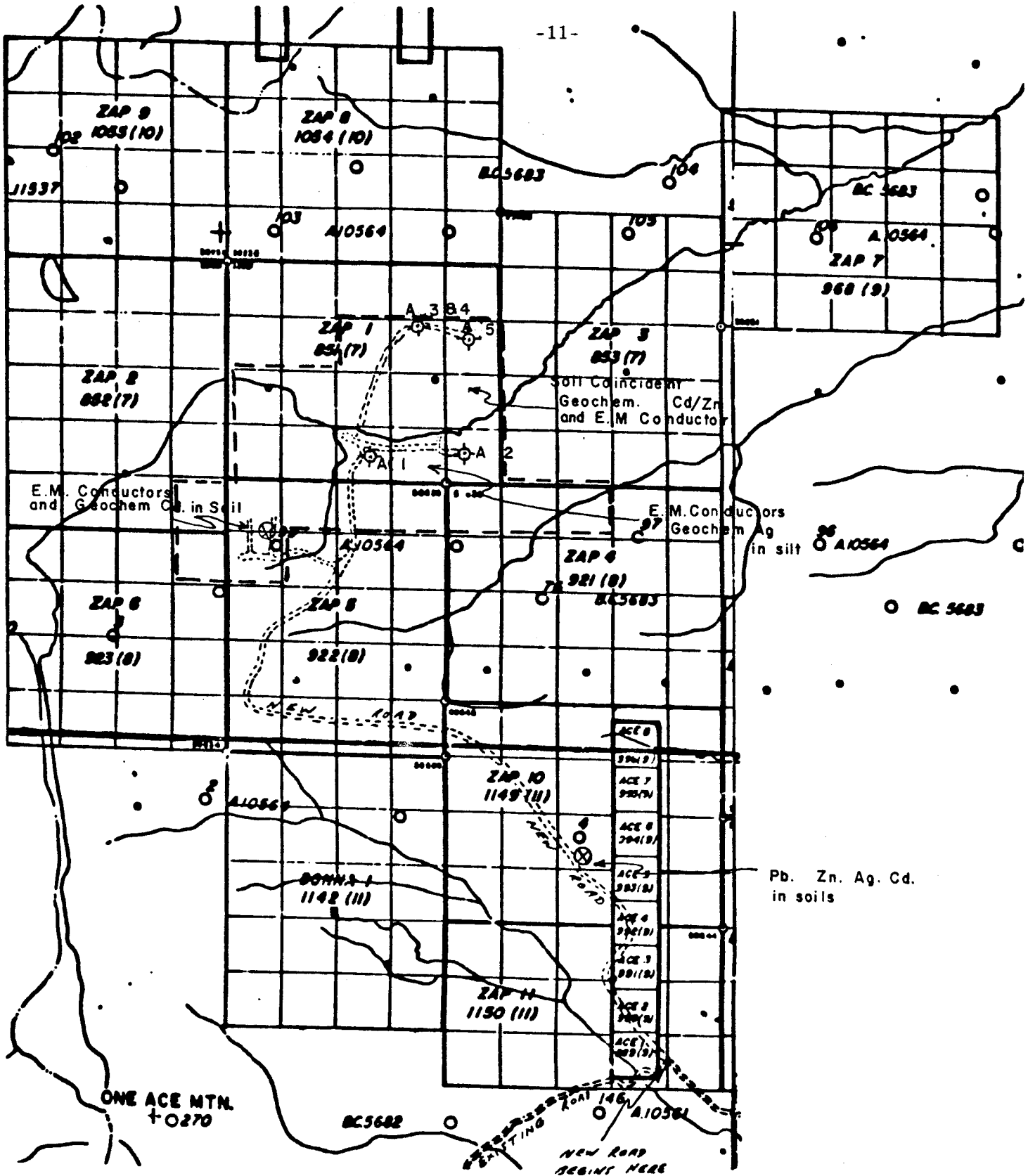
Soil sampling and EM-16 surveys provided the drill targets in 1980 and such work should be continued as fill in lines over known regional anomalies.

The following schedule is recommended:

Summer 1981 - fill in lines of soils and EM-16 in areas of good regional soils geochemical response, survey 1980 drill holes and 1980 roads with transit, survey critically placed tie lines through the grid to relate it to our topographic map and as an accurate record for future years work, mark drill sites, camp sites, water supplies, etc. for a winter drill project, and finish hazard abatement.

Late winter 81/82 (March) - Diamond drilling. Winter work will be more expensive than normal B.C. drill costs but, for this area, is the most economical. The main problem overcome is that of road building. In late winter most swamps and

and creeks are frozen and can be traveled by cat, avoiding forests, thus reducing problems with B.C.F.S. and costs to us. Also, the frozen road needs less maintenance. Finally drill sites can easily be made in swamps and waterways.



ALBERT CREEK SILVER

N.T.S. NO.:104-P-13 : FIG. NO.:079-80-1

SCALE: 1:50 000

⊕ Drill holes 1980

⊗ Future drill targets

--- F.N.M. Road 1979-1980

APPENDIX I

Drill Logs for Holes #A1, A2, A3/A4 and A5

ABBREVIATIONS

H.	hardness	veins and stockworks spacing
no.	none	(eg.) 1-3/cm = 1 to 3 per cm.
mod.	moderate	(eg.) 1-3 cm = 1 to 3 cm between
v.	very	veins.
m.g.	medium grained	
f.g.	fine grained	
vfg	very fine grained	
med.	medium'	
qzte	quartzite	
dia.	diameter	
bed	bedding	
occ.	occasional	
tr.	trace	
diss.	disseminated	
fr.	fracture	
inter.	interstitial	
cm.	centimetre	
mm.	millimetres	
m.	metres	
spl.	sample of core	
qz.	quartz	
calc.	calcite	
py.	pyrite	
gr.	graphite	
cpy	chalcopryrite	
gal	galena	
sph	sphalerite	
?	questionable	
*	dominant (angle, mineral, etc).	



NORTH 10+75  
 West  
 EAST 8+00  
 ELEV. 1016 metres  
 BEARING Vertical  
 DIP -90° collar. Acid test not possible due to high water pressure.

STARTED 26 Aug. 1980  
 COMPLETED 31 Aug. 1980  
 LENGTH 107.3 m (352 ft)

# FALCONBRIDGE DIAMOND DRILL RECORD

PROPERTY

ALBERT CREEK

PURPOSE For geologic  
data, testing EM-16  
conductor, and testing  
near good silt sample.

LOGGED BY J. Wilson

HOLE No. A 1  
 CLAIM Zap 1  
 SECTION \_\_\_\_\_  
 OFFSET \_\_\_\_\_  
 PLOTTED \_\_\_\_\_

FOOTAGE	DESCRIPTION	SAMPLE	FOOTAGE	C.L.				
D. J. Drilling. 17 A Drill B Q Core								
0 - 12.2 m	Overburden							
12.2 - 46.9 m	F.g. - m.g. light grey, porous qzte vugs to 1 cm. dia. Mod. to very hard but occ. soft, buff weathering patches (spl. 16.4 m) Bed: 50° @ 14.3 m, 64° @ 30.5 m. Concretion ? spl: 20.7 m. Sand layer spl: 26.5 - 26.8 m.							-15-
46.9 - 53.3 m	As above plus soft, buff, finely laminated siltstone spl: 47.2 m. Bed - contorted, slumped ? 20° - 80°.							
53.3 - 74.1	Mod H, f.g., light grey qzte vugs to 5 mm. Bed: 50° @ 71.9 m.							
74.1 - 107.3 m	Mod. H, f.g., med. to dark grey impure quartzite, vugs to 5 mm. brown coatings on fractures. Bed: 45° @ 89.9 m							
End of hole								

FOOTAGE	DESCRIPTION	SAMPLE	FOOTAGE	C.L.				
	Spl: 76.8, 84.1, 86.9, 89.9, 92.0, 107.3.							
	Veins							
27.2 - 30.5	Rare 45° to 60° 3 mm qz veinlets							
76.2 - 107.3	Occasional 5° 1mm qz veinlets.							
	Fractures							
12.2 - 12.95	1 cm spacing @ 20°, 70°							
12.95 - 13.9	5 cm @ 20°							
13.9 - 14.0	1 cm @ 35°							
14.0 - 15.5	5 cm @ 40°							-14-
15.5 - 26.5	3 cm @ 5°, 40°							
26.5 - 32.0	5 cm @ 5°, 20°							
32.0 - 32.6	2 cm @ 3°							
45.7 - 70.1	2-5 cm @ 3°, 35°							
70.1 - 72.5	5-2 cm @ 20°, 30°							
72.5 - 83.8	2-5 cm @ 3°, 25°, 40°							
83.8 - 107.3	7-2 cm @ 5°, 25°, 45°							
	mn stain @ 88.4 m							
	HCL test							
12.2 - 18.0	mod, interstitial and vugs							
18.0 - 24.0	Weak, interstitial							

FOOTAGE	DESCRIPTION	SAMPLE	FOOTAGE	C.L.				
24.0 - 26.0	weak, frs.							
26.0 - 28.0	weak, frs, & vugs							
28.0 - 35.0	weak, inter & vugs & frs							
35.0 - 46.0	mod, inter							
46.0 - 48.0	none in siltstone							
48.0 - 51.0	mod, inter							
51.0 - 52.0	mod, inter & vugs, none in siltstone							
52.0 - 53.0	mod, inter							
53.0 - 60.0	none							
60.0 - 68.0	mod, vugs							
68.0 - 70.0	mod, inter & vugs							
70.0 - 72.0	weak inter, strong vugs							
72.0 - 74.0	mod, inter							
74.0 - 75.0	weak inter							
75.0 - 77.0	strong, vugs							
77.0 - 78.0	weak, inter, mod, frs.							
78.0 - 79.0	strong vugs							
79.0 - 81.0	weak, inter							
81.0 - 88.0	mod, frs							
88.0 - 89.0	weak inter, mod vugs							

FOOTAGE	DESCRIPTION			SAMPLE	FOOTAGE	C.L.				
89.0 - 101.0	weak inter & frs									
101.0 - 103.0	weak inter, mod frs.									
103.0 - 104.0	weak inter, weak frs									
104.0 - 106.7	weak inter, mod vugs									
Core Recovery (%) and casing										
0 - 12.2	casing	45.7 - 48.8	25%		85.3 - 88.4	62%				
12.2 - 15.5	90%	48.8 - 51.8	16%		88.4 - 91.4	20%				
		51.8 - 54.9	28%		91.4 - 94.5	16%				
12.2 - 15.2	casing	45.7 - 67.1	casing		94.5 - 97.5	15%				
15.5 - 18.6	40%	67.0 - 70.1	20%		97.5 - 100.6	33%				
18.6 - 23.5	7%	70.1 - 73.1	43%		100.6 - 103.6	57%				
23.5 - 26.5	6%	73.1 - 76.2	48%		103.6 - 107.3	12%				
26.6 - 32.6	sand 25%	76.2 - 79.2	60%							
		79.2 - 82.3	34%							
15.2 - 45.7	casing	82.3 - 85.5	30%							

-16-

NORTH 10+25  
 West 1+00  
~~EAST~~  
 ELEV. 1018 m.  
 BEARING vertical  
 DIP -90° @ collar Acid test @ 150.9 m is -88°

STARTED 1 Sept. 1980  
 COMPLETED 5 Sept. 1980  
 LENGTH 150.9 m (495 ft)

# FALCONBRIDGE

## DIAMOND DRILL RECORD

PROPERTY  
ALBERT CREEK

PURPOSE Geologic data, HOLE No. A 2  
testing EM-16 conductor CLAIM Zap 1  
 SECTION \_\_\_\_\_  
 LOGGED BY J. Wilson OFFSET \_\_\_\_\_  
 PLOTTED \_\_\_\_\_

FOOTAGE	D. J. Drilling	DESCRIPTION	17 A Drill	BQ core	SAMPLE	FOOTAGE	C.U.	Pb	Zn	Ag	Ba
0 - 19 m		casing									
19.7 - 26.8		Mod. H, f.g., light grey quartzite. Occ. trace diss				21.3 - 22.9	10	6	16	0.2	90
		py(?) and py in veinlets. (26 m 5° veinlets with py,				24.4 - 25.9	7	8	12	0.2	40
26.8 - 53.9		Hard, f.g. med to dark grey impure qzte. Buff, iron				27.4 - 28.9	6	8	16	0.2	140
		stained zones: 27.1 - 27.3, 33.5-34.2				30.5 - 32.0	8	8	10	0.2	80
		Bedding 10° @ 27.7 m 5° @ 33.2 m									
		Trace diss. py & possibly cpy throughout				33.5 - 35.0	10	20	20	0.2	40
		28.3 - 32.6 0° to 5° 2 mm calc vein. Tr py & gr ?									
		34.1 - 34.7 5° to 15° 1-4 mm py vein.				36.6 - 38.1	6	6	8	0.2	40
		37.2 - 37.3 altered, sheared? with 1-4 mm pyrite vein									
		as above.				39.6 - 41.1	6	6	8	0.2	20
		Py. increasing gradually downhole									
		40.5 py in 2 mm vugs.				42.7 - 44.2	6	10	8	0.2	40
		41.1 80°, irregular, 2 mm black (Fe ?) vein.									
		41.8 - 45.7 1% py in 15° & 80° 1 mm py. veins and blebs				45.7 - 47.2	6	7	6	0.2	40
		to 1 cm dia, and on rim of calc/qz vugs, and diss.									
		45.7 - 49.1 0.5% py. as above but mainly diss. 1%				48.8 - 50.3	6	6	8	0.2	20
		black mineral in veinlets.									
		51.8 - 53.6 occ. suture-like fractures with trace				51.8 - 53.3	8	10	15	0.2	50
		black mineral & py.									
53.8 - 55.3		Soft, v.f.g., black mudstone as silt/clay or core with									
		0° to 15° fractures. 1% (?) v.f.g. diss. py.;				54.9 - 55.3	33	24	36	0.9	740
		possibly graphitic. Bedding 5° @ 54.6 m.									
55.3 - 57.0		Hard, v.f.g., light grey, quartzite with trace v.f.g.									
		diss. py.									
		55.5 - 57.0 brecciated & poorly cemented, with calc.									
		cavities. Top 5 cm is buff.									

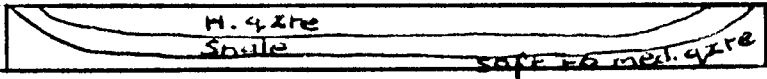
FOOTAGE	DESCRIPTION	SAMPLE	FOOTAGE	C.U.	Pb	Zn	Ag	Ba
57.0 - 60.3	Very hard, f.g., banded light to dark grey quartzite. Usually 2 to 5 mm parallel bands @ 0°, indistinct contacts. Bedding 0° @ 58.5 5° @ 59.7.		57.9-59.4	3	22	128	0.4	30
	Note: (2) 3 cm green volcanic pebbles @ 57.6 m. Caved? or in place? 60.0 - 60.3 py, galena, brown sph., + cpy? in 1-2 mm. veinlet.							
60.3 - 61.9	V. hard, f.g., dark grey, impure qzte with minor light grey streaks. Rare diss, f.g. py.		60.9-61.9	1	2	5	0.2	100
61.9 - 62.6	Light breccia zone. Pale f.g. qzte fragments in calcite rich matrix. Rare, irregular, dark, pyritic veinlets. Bed: 5° @ 62.2		61.9-64.0	4	22	24	0.2	50
62.6 - 67.5	Hard v.f.g. light grey qzte with occ. dark wisps. Patches of mod. H, f.g. med. grey qzte. Traces f.g. py. throughout. 67.2 - 67.5: 1% py-diss. and veinlets and one v.f.g. galena in veinlet.		64.0-65.5	2	10	28	0.2	220
			67.1-67.5	3	10	21	0.2	60
67.5 - 72.4	Mod. to v.h. , f.g., med grey. impure qzte with occ. 1-2 mm black wisps often pyritic. Traces diss f.g. py. cubes throughout and 1 mm py. on frs. 68.4 - 70.6 Brecciated qzte with qz/calc. stockworks 1% py on fractures. Up to 2% black mineral. Lower contact is 35°, 3 cm calc/qz		67.5-68.6	8	16	14	0.2	60
			70.1-71.6	6	4	7	0.2	20
72.4 - 73.8	V.H, vfg, med. grey qzte with 3 - 5% vfg fr. & diss. py.cubes.		73.1-73.8	3	14	4	0.2	340
73.8 - 74.7	Soft to mod. H, f.g, med. grey, impure qzte. Bed: 0° @ 74.1 3 - 5% vfg diss py.		73.8-74.7	15	20	10	0.3	1090

FOOTAGE	DESCRIPTION	SAMPLE	FOOTAGE	C.U.	PE	Zn	Ag	Ba
74.7 - 75.4	Soft to mod. H, m.g, med. grey, impure qzte. Possible breccia. 1 - 3% py-diss and on frs.							
75.4 - 82.3	Hard, f.g, light to med. grey, impure qzte. 79.9 - 80.5 - Breccia. Bed: 10° @ 76.2. 0.5% py in 1 mm irregular frs with black mineral.		76.2-77.7	7	10	20	0.2	80
	Lower contact is 3 cm, 30° qz vein in qzte breccia.		79.2-80.8	6	6	8	0.2	30
82.3 - 82.9	Soft, vfg, med. grey siltstone 3 - 5% diss. vfg py 82.3 m: 5 mm py vein.		82.3-82.9	25	12	24	0.2	70
82.9 - 83.8	V.H, vfg, light grey, qzte with 0°, 1 - 3 mm. wisps of black mineral & pyrite. Dark grey 1 cm band with 2% diss. py. Bed: 10° @ 83.5, 13° @ 83.8		82.9-83.8	35	14	17	0.3	260
83.8 - 85.0	Soft, vfg, med. grey siltstone with 1 - 3% diss. fg. py. occ. veinlets of pale green waxy mineral (clay?).							
85.0 - 86.6	Hard, vfg, light grey impure qzte with zones of med. grey wisps carrying 3% diss. py.  eg. (Dark wisps in light rock - 3% py. throughout light grey - trace py).  Lower contact is 4 cm. brecciated, calc & qz filled at 25°.		85.3-86.6	18	6	11	0.2	390
86.6 - 87.2	Soft, med. grey siltstone. .5% vfg. diss. py.							
87.2 - 90.7	Hard, fg to mg, med. grey qzte .5%-10% diss. vfg py		88.4-88.7	13	17	12	0.2	730

FOOTAGE	DESCRIPTION	SAMPLE	FOOTAGE	C.LI	Fe	Zn	Ag	Ba
	Strongest py to 88.7 2-4% py in frs & diss. to 90.7		88.7-89.9	9	14	25	0.2	480
90.7 - 91.4	Soft, med grey, siltstone. 3% f.g. diss. py.							
91.4 - 93.9	Mod. to hard, f.g. to m.g, impure qzte 1-2% diss. f.g. py. 1% py in 5° to 45° frs. with 1% black mineral		91.4-93.0	9	6	11	0.2	310
93.9 - 95.2	0° bedding contact between (a) v. hard f.g pale grey qzte with 1-2% diss. py and 1% py in frs. (b) soft, graphite ? black shale with white layer (barite ? or calc-qz ? or both ?). 5% diss. py. in shale. Bedding 5° @ 94.2							
95.2 - 97.7	V.hard, f.g, pale grey qzte with patchy trace diss. py & tr. py on frs.		97.5-99.1	10	12	11	0.2	560 <sup>1</sup> / <sub>0</sub>
97.7 - 98.8	Soft, fg-m.g, med. grey, impure qzte. 5% diss. f.g py.							
98.8 - 101.6	V. hard, f.g., light grey. qzte. Occ. dark grey patches. Tr. to 0.5% f.g. diss. py. often cubes. Some on frs.		100.6-102.1	10	16	28	0.2	170
101.6 - 102.1	Hard, f.g. to m.g., pale grey qzte with rare diss.py laminated with mod. H to soft, black, graphitic ? shale with 1% py vein on contact. 3 mm to 2 cm white barite ? or qz-calc ? vein ? in shale . Bed 5° @ 102.1							
102.1 - 103.5	Hard, f.g. to m.g pale grey qzte with occ. dark streaks. Patches of tr. to 1% diss vfg py usually with dark zones.							
103.5 - 103.9	Soft, med. grey siltstone. 3% diss. py. Bed 5° @ 103.6		103.6-103.9	110	28	31	1.3	300



FOOTAGE	DESCRIPTION	SAMPLE	FOOTAGE	C.U.	Pb	Zn	Ag	Ba
103.9 - 105.6	m.g., pale grey qzte 104.4 - 104.7: brecciated 105.0 - 105.6: altered friable. Trace to 0.5% diss. fg. py.		103.9-105.2	23	20	72	0.2	140
105.6 - 107.2	Mod. H, f.g. med. grey, impure qzte. Dense, fine (1 mm) black pyritic stockworks. Total py. 5 to 10%. Lower contact @ 36°.		106.7-107.2	12	16	10	0.3	810
107.2 - 108.8	Mod H, m.g., pale grey, pzte with 1 cm band of shale and white qz ? @ 20°. Tr. diss. f.g. cube py. 107.9 108.2: as 105.6 - 107.2							
108.8 - 110.5	V. hard, f.g. pale grey qzte. Tr. to 0.5% diss. py. 108.8 - 109.7: brecciated, qz-calc filled 110.4: greenish waxy veinlet (clay ?)							-21-
110.5 - 110.9	Mod. H. m.g., med. grey, impure qzte.		110.5-111.2	10	8	16	0.2	360
110.9 - 112.6	Brecciated as 108.8 - 110.5							
112.6 - 113.8	Hard, m.g., med. grey qzte. Irregular 1-2 mm black and pyritic veinlets @ 0° total py. 0.5%.		112.8-113.8	28	20	580	0.2	360
113.8 - 114.3	Mod. H, f.g., dark grey, impure qzte 2% DISS. PY. Bed: 0° @ 114.3.							
114.3 - 118.4	Hard to mod. H, m.g., light grey qzte Vague, laminated appearance to 115.2 (dark wavy 1-3 mm bands and occ. qz veins @ 0 to 10°. 116.4 - 117.3: 30° sheet, soft, friable. 118.1 - 118.4: sheared Tr. diss. py. throughout. 115.2 : 80° 1 to 6 mm py. veining. 115.3: Tr. galena on fr.		115.8-117.3	14	16	94	0.2	190

FOOTAGE	DESCRIPTION	SAMPLE	FOOTAGE	C.U.	Pb	Zn	Ag	Ba
118.4 - 120.2	V.H. f.g., med. grey, qzte		118.9-120.2	6	12	20	0.5	620
	118.4 - 118.9 shear 1% diss. f.g. py. Tr. py. on frs. Lower contact @ 12°		<del>121.9-123.4</del>	<del>7</del>	<del>13</del>	<del>16</del>	<del>0.2</del>	<del>80</del>
120.2 - 120.4	Soft, friable, black shale 2% diss. py.							
120.4 - 123.4	Mod. H to soft, m.g., med. grey, impure qzte. Irregular 1 mm py. stockworks @ 0°, 5°, 80°. 120.4 - 121.0: 0.5% diss. py.		121.9-123.4	7	13	16	0.2	80
123.4 - 127.6	Soft to mod. H, f.g. med. grey qzte. (125.3 - 126.5: ½ to 1cm black shale with 2-5% diss. py. Contact zone is 2 mm py. vein @ 0° to 5°).  eg. 		125.3-126.5	11	66	167	0.8	750
123.4								
	127.6							
	5% diss. f.g. py. as cubes in qzte. some concentration in blackish zones. Note: vague, near boxworks of black throughout. Bed: 5° @ 126.2							
127.6-137.2	V.H., fg, light grey qzte. 0.5% diss vfg py cubes & 0.5% py on frs 131.7-137.2 broken, sheared? 131.7-132.0 whitish, bleached, brecciated.		128.0-129.5	4	12	530	0.4	360
			131.1-132.6	6	19	36	0.2	430
			134.1-135.6	6	16	18	0.2	50
137.2-149.0	usually Hard, f.g light grey qzte. Brecciated, veined, fractured. Some parallel 0° to 5° white and light grey laminations ( 1-3 mm ). Trace diss. vfg py.		137.2-138.7	13	42	1400	0.2	700
			140.2-141.7	11	9	11	0.2	80
149.0 - 150.9	Soft to mod. H, f.g., pale grey qzte trace to 0.5% diss. py. 149.0 - 149.2 med. H, fg, med. grey qzte with 1% diss. py. Bed @ 5°		143.3-144.8	9	20	16	0.2	140
			146.3-147.8	9	12	8	0.2	60
			149.3-150.9	3	6	12	0.2	150
End of hole.								

Veins, stockworks, breccia filling

<u>From</u>	<u>To</u>	<u>Spacing</u>	<u>Width (mm)</u>	<u>Mineral Filling</u>	<u>Angle (°)</u>
19.7	m-29.0 m	1-2/cm	0.5-1	qz*, calc	10,17,55,85
29.0	- 31.4	1-10 cm	0.5-1	calc*,qz	0, 5,40
31.4	- 33.2	5-20 mm	0.5-1	calc	5,80
33.2	- 34.7	1-4 /cm	0.5-2	calc,py,qz	5,20,35,75
34.7	- 38.7	1-2 cm	0.5-2	qz*,calc	10,40,80
38.7	- 39.6	breccia		qz	
39.6	- 50.3	1-3 /cm	0.5-5	calc,qz,py, black mineral	40,80
50.3	- 51.5	1-3 cm	0.5-1	calc,qz	5,20,40,80
51.5	- 53.0	1-3 /cm	0.5-1	calc,qz	5,20,40,80
53.0	- 53.9	2-10/cm	0.5-2	calc,qz	5,20,40,80
53.9	- 55.8	1-2 cm	0.5-1	calc,qz	5,20,40,80
55.8	- 57.3	breccia and 1-2 /cm	0.5-2	calc,qz	15
57.3	- 63.4	1-3 cm	0.5-2	qz*,calc,sph,gal,py	10,50,80
63.4	- 67.4	3-20 cm	0.5-1	qz*,calc	5,40
67.4	- 72.5	1-2 /cm	0.5-2	calc	5,20,60,80
72.5			3 cm vein	calc	33
72.5	- 74.8	5 cm	0.5-3	calc*,qz	20,45,80
74.8	- 80.5	1-3 /cm	0.5-4	calc,qz,py, black mineral	5,25,80
80.5	- 84.6	1-3 cm	0.5-3	calc*,qz	10,20,80
83.8			2 cm vein	qz, calc	15
82.3			3 cm vein	qz,calc,py	35
84.6	- 86.7	1-2 /cm	0.5-4	calc*,qz	30,40
86.7			2 cm vein	qz,calc	20
86.7	- 88.7	5-20 cm	0.5-1	calc,qz	20,80
88.7	- 89.1	1 cm	0.5-3	calc	15,25
89.1	- 89.6	10-20 cm	0.5	calc	15,20
89.6	- 93.0	1-3 cm	0.5-2	calc	15,20
93.0	- 93.3	10-20 cm	0.5	calc*,qz,py	10
93.6	- 94.5	2-5 cm	0.5-2	qz*,calc	5,25
94.5	- 97.5	1-2 /cm	0.5-3	calc*,qz	5,30,50
94.8			1 cm vein	calc,qz	20
97.5	- 98.7	20 cm	0.5	calc,qz	4
98.7	- 101.2	1 /cm	0.5-2	calc	15,50
101.2	- 104.2	1-5 cm	0.5-2	calc*,qz	15,30
101.8			1 cm vein	qz,calc	5
104.2	- 104.5	3 cm	0.5-1	calc	
104.5	- 108.8	2-10 cm	0.5-1	calc	5,20
108.8	- 109.7	primary ? ( sedimentary ) breccia zone with: 5-10 cm	0.5-1	calc,qz	15,30
109.3			1 cm vein	qz*,calc	15

Veins, stockworks, breccia filling

<u>From</u>	<u>To</u>	<u>Spacing</u>	<u>Width (mm)</u>	<u>Mineral Filling</u>	<u>Angle (°)</u>
109.7m	- 110.3m	3-10 cm	0.5-1	calc	40
110.3	- 110.6	1-3 cm	0.5-1	calc,qz,waxy grain mineral	5,40
110.6	- 112.9	2-4 cm	0.5-2	calc	10,30,70
112.9	- 113.6	1 cm	0.5-2	qz*,calc	10,30,70
113.6	- 116.7	5-20 cm	0.5-2	qz*,calc	30,75
115.1			1 cm vein	py	75
116.7	- 118.9	1-4 /cm	0.5-5	calc*,qz	15,35,70
118.9	- 120.4	2-5 cm	0.5-2	calc*,qz	15,35,70
120.4	- 122.7	1-2 /cm	0.5-3	calc,qz,py	5,45
112.7	- 122.9	2-4 /cm	0.5-3	calc*,qz	30
122.9	- 127.7	10-20 cm	0.5-1	calc,qz	5,30
127.7	- 134.7	1 /cm	0.5-2	qz*,calc	5,25,35,55
134.7	- 140.2	breccia zone, gouge		calc	
140.2	- 149.0	1-2 /cm	0.5-2	calc	15,20,70
149.0	- 150.8	1-3 cm	0.5-2	calc,qz*	20,35,50

Fractures

<u>From</u>	<u>To</u>	<u>Spacing (cm)</u>	<u>Angle</u>
19.7m	- 21.9m	1-5	25,60*
21.9	- 25.9	10	70
25.9	- 33.2	15-30	10,25,55
33.2	- 34.7	1-10	10,80*
34.7	- 52.6	10-25	70
52.6	- 53.0	1-5	5,80*
53.0	- 53.8	5-15	25,60
53.8	- 55.3	clay, graphitic, pyritic gouge	
55.3	- 57.4	10-20	65,80
57.4	- 58.7	1-5 & pebbles	5,45,80
58.7	- 63.4	5-7	50,70
63.4	- 68.3	1-5	5,40,70
68.3	- 71.3	7-20	55
71.3	- 75.0	5-15	40
75.0	- 75.3	clay, graphitic, pyritic, gouge 30	
75.3	- 78.2	10	10,40,65
78.2	- 80.1	1-5	35,75
80.1	- 80.2	crumbly - clay, graphite,py	

Fractures

<u>From</u>	<u>To</u>	<u>Spacing</u>	<u>Angle</u>
80.2m	- 84.7m	1-5	5,35*,75*
84.7	- 90.5	10-15	30,50
90.5	- 98.4	5-10	0,75
98.4	- 99.1	1-5	5,25
99.1	- 99.7	5	65
99.7	- 100	0.5-3	50
100.0	- 100.9	1-5	55
100.9	- 103.0	5-15	10,65
103.0	- 105.8	5	5,35
105.8	- 110.0	10-20	5,25
110.0	- 116.4	7-15	5,30
116.4	- 121.0	1-5	5,35,45
121.0	- 122.8	5-15	5,35
122.8	- 126.5	1-5	0,5,25
126.5	- 128.9	5-20	10,40,80
128.9	- 131.7	1-5	25,80
131.7	- 137.2	0.5-1 & pebbles	
139.9	- 150.9	5-15	5,40,60

Acid Test

<u>From</u>	<u>To</u>		
19.7m	- 27.4m	moderate	interstitial and veinlets
27.4	- 33.5	strong	interstitial
33.5	- 38.1	moderate	interstitial
38.1	- 51.8	strong	interstitial
51.8	- 56.4	v. strong	interstitial
56.4	- 57.8	strong	interstitial
57.8	- 67.0	weak frs	
67.0	- 71.6	strong	interstitial
71.6	- 74.7	weak frs	
74.7	- 86.9	strong	interstitial
86.9	- 95.1	moderate	interstitial
95.1	- 101.2	weak	frs
101.2	- 101.8	strong	
101.8	- 103.0	weak	frs

Acid Test

<u>From</u>	<u>To</u>		
103.0m	- 105.5m	moderate	frs
105.5	- 107.1	weak	frs
107.1	- 108.8	v. strong	none in shales
108.8	- 123.3	usually v. strong in white matrix of fragmental rock none in dark grey zones of qzte.	
123.3	- 131.7	weak	frs
131.7	- 131.8	v. strong	
131.8	- 138.4	moderate	interstitial
138.4	- 150.6	strong	interstitial in shear ?
150.6	- 150.9	moderate	interstitial

Core Recovery

<u>From</u>	<u>To</u>	<u>%</u>
18.3m	- 21.3m	55
21.3	- 24.4	95
24.4	- 33.5	100
33.5	- 36.6	95
36.6	- 45.7	100
45.7	- 48.8	95
48.8	- 54.9	100
54.9	- 57.9	95
57.9	- 128.0	100
128.0	- 131.0	60
131.0	- 134.1	55
134.1	- 137.2	45
137.2	- 140.2	60
140.2	- 143.3	95
143.3	- 146.3	20
146.3	- 149.3	20
149.3	- 150.9	100

NORTH 19 + 90  
 West 4 + 75  
 ELEV. 1077 metres  
 BEARING Vertical  
 DIP -90° @ collar

STARTED 5, Sept 1980

COMPLETED 17, Sept 1980

LENGTH 40.2 metres (132 feet)

# FALCONBRIDGE

## DIAMOND DRILL RECORD

PROPERTY

ALBERT CREEK

D. J. Drilling, 17A Drill, B.Q. core

PURPOSE Geologic data,

EM-16 conductor, soil

sample anomaly.

LOGGED BY J. Wilson

HOLE No. A 3 & A 4

CLAIM Zap 1

SECTION \_\_\_\_\_

OFFSET \_\_\_\_\_

PLOTTED \_\_\_\_\_

FOOTAGE	DESCRIPTION	SAMPLE	FOOTAGE	C. LI	PE	Zn	Ag	Ba
0 - 21.3 m	N casing to 24 m. B casing to 36.6 m							
21.3 - 40.2	Black Graphitic, sheared shale. Probably near vertical fault. Pyrite throughout as 0.5% (?) diss. v.f.g. One 3 cm. breccia fragment seen near top of hole with 50% pyrite in dark f.g. mass. Very angular clasts to 2 cm. are fine laminated mudstone. Black laminae alternating with light grey. Stockwork veinlets across fragments, not matrix. No veining. Fractures, Structures.							-27-
21.3 - 22.9	1 cm., gravel/chips		21.3 - 24.4	26	14	90	0.2	920
22.9 - 23.5	1-3 cm @ 5°, 45°							
23.5 - 24.4	3 mm. to 1 cm. flakes, occasional 3 cm. block.							
24.4 - 27.4	2 cm. pebbles.		24.4 - 30.5	38	16	143	0.2	1160
27.4 - 29.6	3 mm. to 1 cm. flakes							
29.6 - 34.7	0.5 to 1 cm. chips @ 5° to 20°		30.5 - 36.6	38	16	173	0.3	1350
34.7 - 40.2	5 to 20 cm. segments, some chips @ 5° to 20°		36.6 - 40.2	39	18	187	0.2	1190
End of Hole	Dark to medium grey breccia fragments in black matrix throughout hole.							
	HCL test							
	Strong to moderate reaction throughout Interstitial and in veinlets.							

FOOTAGE	DESCRIPTION	SAMPLE	FOOTAGE	C.L.				
	Core Recovery							
21.34 - 24.4	40%							
24.4 - 27.4	15%							
27.4 - 30.5	15%							
30.5 - 33.5	45%							
33.5 - 36.6	90%							
36.6 - 39.6	40%							
39.6 - 40.2	25%							

-28-

SLUDGE SAMPLES

Footage (m)	Pb	Zn	Ag	Ba
30.5	20	220	0.2	1490
36.6	23	260	0.3	1740
40.2	16	136	0.2	1180



NORTH 20 +80EAST 2 + 00ELEV. 1077 metresBEARING VerticalDIP -90° @ collar, 87° @ 148 MSTARTED 16, Sept 1980COMPLETED 22, Sept 1980LENGTH 148.4 m. (487 feet)

## FALCONBRIDGE

## DIAMOND DRILL RECORD

PROPERTY

ALBERT CREEK

PURPOSE Geologic data,EM-16 conductor, soilsample anomalyLOGGED BY J. WilsonHOLE No. A 5CLAIM Zap 1

SECTION \_\_\_\_\_

OFFSET \_\_\_\_\_

PLOTTED \_\_\_\_\_

D.J. Drilling 17A Drill, B.Q. core

FOOTAGE	DESCRIPTION	SAMPLE	FOOTAGE	C.U.	Pt	Zn	Ag	Ba
0 - 30.5	Casing							
30.5 - 45.7	Very soft, crumbly black shale to 38.7 and broken chips of black argillite to 45.7.		30.5 - 32.0		16	230	1.4	1280
			33.5 - 35.0		22	1400	3.4	1830
45.7 - 48.8	Mod. to v. Hard black, fragmented argillite. Rounded clasts (0.5 to 1 cm. dia). Clast supported. 20% v.v.fg diss. py. in clasts. Matrix is black with diss py. occ. 5mm. py bleb. and py. on edges of calc veins. 45.8: brucite crystals in vertical vein (?) Total py = 5% (?) Graphite on frs.		45.7 - 47.2	52	14	43	2.3	1210
48.8 - 53.9	As above but angular brecciated fragments to 3 cm. diameter. Gradation of roundness indicates possible vent/collapse type structure.		48.8 - 50.3	46	12	110	1.4	800
			51.8 - 53.3	71	10	124	3.0	1210
	51.8, 49.4 m. 25° banding (bed?)							
	50.6 m. grey specks gal.? graph.?							
	50.3 - 53.9 - strong black material and diss & bleb py. + 10%							
53.9 - 81.4	As above but usually massive. Higher py (10-15%) diss., layers & veins in black argillite.		54.9 - 56.4		16	570	0.9	1020
			57.9 - 59.4	66	14	630	1.0	1100
	57.9: 1mm. 60° py. layer		60.9 - 62.5		14	144	0.2	1250
	59.4: 1mm. 40° py. layer		64.0 - 65.5		14	138	1.0	1150
	60.6: 5mm. 55° py layer							
	66.4: 3mm. 40° py layer		70.1 - 71.6		8	174	4.2	940

v-11.

73.1 - 74.7

76.2 - 77.7

6 166 2.8 780  
HOLE No. A5  
6 110 0.8 1270

FOOTAGE	DESCRIPTION	SAMPLE	FOOTAGE	C.U.	Pb	Zn	Ag	Ba
81.4 - 91.1	As above but 1% (?) v.v.f.g. diss. py.		79.2 - 80.8		17	126	1.0	1810
91.1 - 114.6	As above but py as bands, veins, blebs, diss., and irregular patches total 5-10%		82.3 - 83.8		8	210	2.1	770
	92.3: 1-2 mm. 50° py. layers and calc. veinlet		85.2 - 86.9	30	15	420	1.8	1110
	w. specks of grey mineral (galena ?)							
	107.6 - 107.9: 1-3 cm. irregular massive py. patch		88.4 - 89.9	40	25	350	2.4	920
	111.9: 3 mm. 35° py. band.							
	112.2 - 113.1: 1 mm. calc. vein with occ. grey specks.		91.4 - 93.0	33	22	320	1.6	720
	114.0: 1 cm. 30° py layer and grey specks in 1 mm. calc vein.		94.5 - 96.0	32	9	230	2.4	740
114.6 - 130.1	As above (massive black, moderate to hard argillite). Trace to 1% diss. vfg py. graphite on fractures and some strongly graphitic zones (116.1 - 116.7, 121.0 - 121.3, 122.2 - 122.5)		97.5 - 99.1	44	10	310	4.0	840
			100.6 - 102.1		10	260	5.7	620
			103.6 - 105.2		30	1300	7.0	1100-30
130.1 - 134.1	As above but 3% py. (vfg. diss, veins, 5 mm. blebs).		115.8 - 117.3	43	8	340	3.8	720
134.1 - 136.5	As above but stockworks prominent. Trace diss. py.		118.9 - 120.4		6	250	3.4	570
136.5 - 146.3	Upper contact is 5 mm. py. layer @ 40°. Rock is as above but trace to 1% total py. as patches of 5% (?) diss. vfg py., blebs, layers		121.9 - 123.4		10	370	6.0	710
	142.3: 1 cm. py layer @ 55°.		125.0 - 126.5		7	320	3.5	720
			128.0 - 129.5		25	1200	5.6	1120
→ 146.3 - 148.4	As above but trace py.		131.1 - 132.6		26	370	0.3	810
End of hole.			134.1 - 135.6		8	80	0.2	490
	<u>Veins, stockworks, breccia fillings</u>		137.2 - 138.7		48	5500	4.2	910
			138.7 - 140.2	97	70	5800	4.0	800
30.5 - 148.4	All calcite filling except at 96.8 & 146.0 : 2-3 mm. calcite		140.2 - 141.7		44	5000	4.4	740
30.5 - 45.7	Broken rock.		143.3 - 144.8		22	3200	2.4	760
			146.3 - 147.8		12	290	0.2	1040

FOOTAGE	DESCRIPTION	SAMPLE	FOOTAGE	C.L.				
45.7 - 54.5	Brecciated with 1 mm. stock works 1 to 5 /cm spacing (1 cm vein 45.9 m. 45°) (5 mm vein 47.8 m. 5°) (1 cm vein 50.3 m. 10°)							
54.5 - 70.1	5 - 10 cm spaced 1 - 2 mm vein @ 45°, 55°							
70.1 - 80.5	Breccia zone and 1/cm veins @ 60°							
80.5 - 82.0	5 - 10 cm. spaced 1 mm. @ 50°							
82.0 - 89.6	2 - 5 /cm spaced stockworks 85.3 - 89.6: very broken							
89.6 - 106.7	2 / cm spaced stockworks 0.5 - 1 mm.							
106.7 - 112.6	2 - 5 /cm spaced stockworks 0.5 - 1 mm. occasional 3 mm. vein @ 15°, 60°.							
112.6 - 122.5	1 - 3/cm spaced stockworks 0.5 - 1 mm.							
122.5 - 133.8	1 / cm spaced stockworks 0.5 - 2 mm. (123.4 1 cm 25° vein) (132.6 4 mm 20° vein)							
133.8 - 133.9	3 mm irregular calcite veins.							
133.9 - 136.2	1 - 5/cm spaced stockworks and breccia filling 0.5 - 2 mm. 134.2 2 cm qz, calc, breccia filling @ 55°, 80°.							
136.2 - 137.0	1 / cm spaced 1 mm @ 25°.							
137.0 - 146.3	2 - 3 / cm spaced 0.5 - 1 mm stockworks.							
146.3 - 148.1	Stockworks and breccia filling 0.5 - 3 mm.							
148.1 - 148.4	1 - 2 cm spacing @ 20°, 70°.							
			SLUDGE SAMPLES					
			Footage (m)	Cu	Pb	Zn	Ag	Ba
			21.3		10	600	0.4	2040
			44.8		12	510	1-2	1480
			82.3		20	690	15.	1270
			85.3		18	890	4.2	1060
			88.4		16	480	2-5	1240
			91.4		24	940	6.6	1070
			94.5		20	570	5.8	1150

-51-

Fractures

<u>From</u>	<u>To</u>	<u>Fracture spacing (cm.)</u>	<u>Angle (°)</u>
30.5	- 45.7	1-5	
45.7	- 47.2	5-10	45,55
47.2	- 54.2	10-30	0,45*,15
54.2	- 56.4	2-5	35,55
56.4	- 59.1	5-10	0,55,70
59.1	- 60.0	1-5	55,70
60.0	- 60.6	10	20,70
60.6	- 61.6	1	0,70
61.6	- 62.5	5-10	25,55
62.5	- 65.2	1-5	5,30
65.2	- 84.1	5-15	30,65*
84.1	- 88.1	1-3	irregular
88.1	- 91.4	0.5*-5	crumbly*,5,30
91.4	- 95.7	1*-5	0,80
95.7	- 98.4	2-10*	5,30
98.4	- 99.1	1-5	0,80
99.1	- 105.5	5-15	5,60*
105.5	- 106.7	2-5	15
106.7	- 110.9	10-20	15,30
110.9	- 111.5	1-5	5
111.5	- 114.6	5-15	5,20
114.6	- 117.3	1-5	5,30
117.3	- 127.8	2-10	25,50
127.8	- 133.8	3-5	25
133.8	- 138.4	10-15	30,45,60
138.4	- 139.3	1-5	45,60
139.3	- 139.9	10-20	25
139.9	- 141.1	1-3	45,60
141.1	- 142.3	10.15	25
142.3	- 143.3	5-10	5,30
143.3	- 145.4	1-5	0,45
145.4	- 148.1	10-15	60
148.1	- 148.4	1-5	25

HCL Test

30.5	- 65.2	strong	interstitial
65.2	- 133.2	none	
133.2	- 136.2	strong	interstitial
136.2	- 146.6	none	
146.6	- 148.2	strong	

Core Recovery (%)

<u>From</u>	<u>To</u>	<u>%</u>
30.5	- 33.5	30
33.5	- 36.6	16
36.6	- 42.7	2
42.7	- 45.7	10
45.7	- 54.9	100
54.9	- 57.9	80
57.9	- 61.0	100
61.0	- 64.0	80
64.1	- 67.1	45
67.1	- 70.1	50
70.1	- 73.1	60
73.1	- 76.2	40
76.2	- 79.2	100
79.2	- 82.3	100
82.3	- 85.3	60

<u>From</u>	<u>To</u>	<u>%</u>
85.3	- 88.4	30
88.4	- ,94.5	75
94.5	- 97.5	80
97.5	- 100.6	75
100.6	- 115.8	100
115.8	- 121.9	95
121.9	- 125.0	55
125.0	- 131.1	65
131.1	- 134.1	90
134.1	- 137.2	100
137.2	- 140.2	80
140.2	- 143.2	90
143.2	- 146.3	70
146.3	- 148.4	90

Sheet 5

Hole A5

APPENDIX II

Thornhill Mineralogical Report (JO#2754)

by R. Buchan

Albert Creek Project

A. Drill core A2, 197.0' - 197.2'

A description of PTS 5893, cut from the mineralized drill core, is given on the following page. The host rock is an equigranular, fine grained quartzite. It is cut by veinlets of carbonate-sphalerite accompanied by lesser pyrite, arsenopyrite, chalcopyrite and galena.

B. "Gunnar's sphalerite"

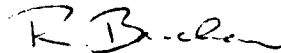
The brown mineral in a carbonate matrix now consists entirely of goethite with traces of hematite. However, relict textures in the polished section (PS7202) and the presence of zinc (Table I, sample B), strongly indicate that the goethite has completely replaced sphalerite.

C. Overburden cuttings, hold A2 (10', 50' and 100' depth).

At first examination, there appeared to be no obvious heavy minerals in the three samples of drill cuttings. The samples from 10' and 50' were both treated by heavy liquid separation (S.G. - 2.9), giving sink fractions of about 20% the original volume. Both heavy fractions appeared similar and a portion of the sample from 10' was crushed and X-rayed. It consists of feldspar, clinopyroxene, amphibole and possibly epidote with no indications of any common heavy minerals.

The sample from A2 at 100' was also X-rayed. It consists of quartz, mica, chlorite, feldspar and possibly calcite and dolomite. Again there are no indications of a heavy mineral constituent. The presence of barium in the qualitative spectrographic analysis, Table I, might indicate the presence of minor amounts of barite.

In summary, no abnormal amounts of heavy minerals were detected in any of the samples which would account for the difficulty in raising the cuttings during drilling.



RB:sls  
Attach.

B. Buchan

FALCONBRIDGE METALLURGICAL LABORATORIES  
QUALITATIVE SPECTROGRAPHIC ANALYSIS

DISTRIBUTION: \_\_\_\_\_ REPORT No. \_\_\_\_\_

ANALYTICAL METHOD: \_\_\_\_\_

REQUESTED BY: \_\_\_\_\_ DATE: \_\_\_\_\_

RECEIVED FROM: \_\_\_\_\_ CHARGE: JO#2754

SAMPLE No.: 80-705 No. of SAMPLES: 1

SAMPLE DESCRIPTION: Mineralized Sediment

DRILL CORE (DDH #A2)

A2, 197.0' - 197.2'

10	- 100%	Si
3	- 30%	Zn
1	- 10%	Fe
0.3	- 3%	Mg, Ca
0.1	- 1%	
0.03	- 0.3%	Al, Ni, Cu, Ti
0.01	- 0.1%	Mn, Cr, Co
0.003	- 0.03%	Pb
0.001	- 0.01%	
0.0003	- 0.003%	
0.0001	- 0.001%	Ag
< 0.0003%		
I		K, Sr, Cd, Zr
S		

I = Interference prevents positive identification.

S = Strong spectral lines, unable to estimate amount.

Unless specified above, the following were not detected at the approx. ppm  
lower limits of 0.5 Cu, Ag; 1 Mn; 5 Mg, Cr; 10 Ba, Be, Bi, Ca, Co, Ni, V;  
25 Ge, Fe, Pb, Mo, Si, Sr, Sn, Ti, Zr, Tl, Pd; 50 Al, Sb, B, Cd, Ga, In, Li, Zn;  
100 As, Au, Na; 200 Rh, Re, Ir, Pt, Ru, Sc; 300 Te, Os; 1000 K, U, Th; 2000 P.



FALCONBRIDGE METALLURGICAL LABORATORIES  
QUALITATIVE SPECTROGRAPHIC ANALYSIS

DISTRIBUTION: \_\_\_\_\_ REPORT No. \_\_\_\_\_

ANALYTICAL METHOD: \_\_\_\_\_

REQUESTED BY: \_\_\_\_\_ DATE: \_\_\_\_\_

RECEIVED FROM: \_\_\_\_\_ CHARGE: 83-302

SAMPLE No.: 80-705 No. of SAMPLES: 3

SAMPLE DESCRIPTION: Rock Samples

*Cuttings from DDH A2*

A2 cuttings at 100'

"B"

"SC-1"

Scraping of sphal? in  
calcite

Black S after Py(?)

10	- 100%	Si		
3	- 30%	Fe, Ca, Al	Fe	Fe
1	- 10%			Si
0.3	- 3%	Mg, Ti, Na	Si	
0.1	- 1%		Mg, Ca	Mg, Al
0.03	- 0.3%		Zn	
0.01	- 0.1%	Mn, Ni, V, Ba	Al, Pb	Ni, Cu, Ti
0.003	- 0.03%	Cr	Ni	Ca, Co
0.001	- 0.01%	Cu, Zn	Mn, Cu, Au	Mn
0.0003	- 0.003%	Pb, Co, Mo		Mo
0.0001	- 0.001%		V, Mo	V
< 0.0003%			Ag	Ag
I		B, K, Sr, Be, Zr	K, Sr, Zr	K, Sr, Zr
S				

I = Interference prevents positive identification.

S = Strong spectral lines, unable to estimate amount.

Unless specified above, the following were not detected at the approx. ppm lower limits of 0.5 Cu, Ag; 1 Mn; 5 Mg, Cr; 10 Ba, Be, Bi, Ca, Co, Ni, V; 25 Ge, Fe, Pb, Mo, Si, Sr, Sn, Ti, Zr, Tl, Pd; 50 Al, Sb, B, Cd, Ga, In, Li, Zn; 100 As, Au, Na; 200 Rh, Re, Ir, Pt, Ru, Sc; 300 Te, Os; 1000 K, U, Th; 2000 P.

Location Albert Creek Project, B.C.

Lab. No. 80-705

Sample Description DRILL CORE (DDH A2)  
DDH A2, 197' - 197.2'

PTS No. 5893

MINERALS	Est. % by Vol.	Grain Size (m.m.)	
		Max.	Avg.
Quartz	86	-	0.12
Tourmaline	Tr	-	0.08
Zircon	Tr	-	0.05
Carbonate	5	-	0.25
Sphalerite	5	-	0.25
Galena	<1	-	0.02
Chalcopyrite	1	-	<0.01
Arsenopyrite	Tr	-	0.08
Pyrite	<1	-	0.12
Pyrrhotite	Tr	-	0.01

## DESCRIPTION

The host rock consists of equigranular partly rounded quartz with scattered grains of tourmaline and zircon. It is transected by trains of minute gas and/or fluid inclusions and with occasional sulphide-carbonate veins. Sulphides consist mainly of dark brown sphalerite containing abundant micro-inclusions of exsolved chalcopyrite, pyrrhotite and galena. Subhedral arsenopyrite and pyrite occur as blocky grains associated with the sphalerite veinlets.



# BONDAR-CLEGG & COMPANY LTD.

130 PEMBERTON AVENUE, NORTH VANCOUVER, B.C.

(604) 985-0681

TLX: 04-352667

## Geochemical Lab Report

FROM: Falconbridge Nickel Mines

REPORT NUMBER: 21 - 89

PROJECT: 079 WILSON Albert Creek

DATE: February 2, 1981

SAMPLE NUMBERS		Si	Fe	Mn	Al	Ca				
From	to									
21.3	22.9	10	6	16	0.2	90				
24.4	25.9	7	8	12	0.2	40				
27.4	28.9	6	8	16	0.2	140				
30.5	32.0	8	8	10	0.2	80				
33.5	35.0	10	20	20	0.2	40				
36.6	38.1	6	6	8	0.2	40				
39.6	41.1	6	6	8	0.2	20				
42.7	44.2	6	10	8	0.2	40				
45.7	47.2	6	7	6	0.2	40				
48.8	50.3	6	6	8	0.2	20				
51.8	53.3	8	10	15	0.2	50				
54.9	55.3	33	24	36	0.9	740				
57.9	59.4	3	22	128	0.4	30				
60.9	61.9	1	2	5	0.2	100				
61.9	64.0	4	22	24	0.2	50				
64.0	65.5	2	20	28	0.2	220				
67.1	67.5	3	10	21	0.2	60				
67.5	68.6	8	16	14	0.2	50				
70.1	71.6	6	4	7	0.2	20				
73.1	73.8	3	14	4	0.2	340				
73.8	74.7	15	26	10	0.3	1090				
76.2	77.7	7	10	20	0.2	30				
79.2	80.8	6	6	8	0.2	30				
82.3	82.9	25	12	24	0.2	70				
82.9	83.8	35	14	17	0.2	260				
85.3	86.6	18	6	11	0.2	390				
88.4	88.7	13	17	12	0.2	750				
88.7	89.9	9	14	25	0.2	480				
<del>91.4</del> 91.4	93.0	9	6	11	0.2	310				
97.5	99.1	10	12	11	0.2	560				
100.6	102.1	10	26	29	0.2	170				
103.6	103.9	110	23	31	1.3	300				
103.9	105.2	23	20	72	0.2	140				
106.7	107.2	12	15	10	0.3	310				
110.5	111.2	10	3	16	0.2	350				
112.8	113.8	28	30	530	0.2	360				
115.8	117.3	14	16	34	0.2	190				
118.9	120.2	6	12	20	0.5	620				
121.9	123.4	7	13	16	0.2	30				
125.3	126.5	11	66	167	0.8	750				

FOR METHOD, EXTRACTION AND FRACTION USED - SEE ATTACHED

# Geochemical Lab Report

REPORT NUMBER: 21 - 83

PAGE: 2

From	SAMPLE NUMBERS	To	Fe	Mn	Zn	Pb	Cd	Hg
128.0	2-430	129.5	4	12	330	0.4	360	
131.1	430	132.6	6	19	34	0.2	430	
134.1	440	135.6	6	16	18	0.2	50	
137.2	450	138.7	13	42	1400	0.2	700	
140.2	460	141.7	11	9	11	0.2	30	
143.3	470	144.8	9	20	26	0.2	140	
146.3	430	147.8	9	12	8	0.2	60	
149.3	450	150.9	3	6	22	0.2	150	
21.3	4	24.4	26	14	98	0.2	920	
24.4	80	30.5	38	16	143	0.2	1160	
30.5	100	36.6	20	16	179	0.3	1350	
36.6	120	40.2	39	18	187	0.2	1190	
30.5	5-100	32.0	-	18	250	1.4	1230	
33.5	110	35.0	-	22	1400	3.4	1830	
45.7	150	47.2	32	14	43	2.3	1210	
48.8	160	50.3	48	12	110	1.4	800	
51.8	170	53.3	71	10	124	3.0	1210	
54.9	180	58.4	-	16	370	0.9	1020	
57.9	190	59.4	86	14	630	1.0	1100	
60.9	200	62.5	-	-14	144	0.2	1250	
64.0	210	63.5	-	-14	138	1.0	1150	
<del>64.0</del> 20.1	230	<del>71.6</del> 71.6	-	-8	174	4.2	940	
73.1	240	74.7	-	-6	166	2.8	700	
76.2	250	77.7	-	-6	110	0.8	1270	
79.2	260	80.8	-	-17	126	1.0	1810	
82.3	270	83.8	-	-8	210	2.1	770	
85.3	230	86.9	30	15	420	1.8	1110	
88.4	290	89.9	48	25	350	2.4	920	
91.4	300	93.0	33	22	320	1.6	720	
94.5	310	96.0	32	9	230	2.4	740	
97.5	320	99.1	44	10	310	4.0	840	
100.6	330	102.1	-	-10	260	5.7	620	
103.6	340	105.2	-	-30	1300	7.0	1100	
115.8	380	117.3	43	8	340	3.8	720	
118.9	390	120.4	-	-6	250	3.4	570	
121.9	400	123.4	-	-10	370	6.0	710	
125.0	410	126.5	-	-7	320	3.5	720	
128.0	420	129.5	-	-25	1200	5.4	1120	
131.1	430	132.6	-	-26	370	0.5	810	
134.1	440	135.6	-	-8	0	0.2	490	
137.2	450	138.7	-	-48	3500	4.2	910	
138.7	455		97	70	3000	4.0	800	
140.2	460	141.7	-	-44	3000	4.4	740	
143.3	470	144.8	-	-22	3200	2.3	760	
146.3	480	147.8	-	-12	290	0.2	1040	

# Geochemical Lab Report

REPORT NUMBER: 21 - 95

Sledge

PAGE: 3

*Feet* *metres*

SAMPLE NUMBERS			Cu ppm	Pb ppm	Zn ppm	As ppm	Mn ppm				
30.5	A3	100	.	20	220	0.2	1490				
36.6	AA	120	-	23	260	0.3	1740				
40.2		132	.	16	136	0.2	1150				
21.3	A3	70	.	10	600	0.4	2040				
44.8		147	.	12	310	1.2	1480				
82.3		270	.	20	690	13.	1270				
85.3		250	.	18	690	4.2	1060				
88.4		190	.	16	430	2.3	1240				
91.4		300	.	24	940	6.6	1070				
94.5		310	.	20	370	3.8	1150				



# BONDAR-CLEGG & COMPANY LTD.

130 PEMBERTON AVENUE, NORTH VANCOUVER, B.C.

(604) 985-0681

TLX: 04-352667

## Geochemical Lab Report

FROM: Falconbridge Nickel Mines

REPORT NUMBER: 21 - 209

PROJECT: 079

DATE: February 20, 1981

SAMPLE NUMBERS	Pb ppm	Ba ppm							
2 - 70	-	90							
80	-	40							
90	-	140							
100	-	80							
110	-	40							
120	-	40							
130	-	20							
140	-	40							
150	-	40							
160	-	20							
170	-	50							
180	-	740							
190	-	30							
200	-	100							
203	-	50							
210	-	220							
220	-	60							
221	-	60							
230	-	20							
240	-	340							
242	-	1090							
250	-	80							
260	-	30							
270	-	70							
272	-	260							
280	-	390							
290	-	780							
291	-	480							
300	-	310							
340	-	300							
341	-	140							
350	-	810							
362	-	360							
370	-	360							
380	-	190							
390	-	620							
400	-	80							
411	-	750							
420	-	360							
430	-	430							

FOR METHOD, EXTRACTION AND FRACTION USED - SEE ATTACHED

# BONDAR-CLEGG & COMPANY LTD.

## Geochemical Lab Report

REPORT NUMBER: 21 - 209

PAGE: 2

SAMPLE NUMBERS	Pb ppm	Ba ppm						
2 - 440	-	50						
450	-	700						
460	-	80						
470	-	140						
480	-	60						
490	-	150						
4 - 70	-	920						
80	-	1160						
100	-	1350						
120	-	1190						
5 - 100	16	1280						
110	22	1830						
150	-	1210						
160	-	800						
170	-	1210						
180	16	1020						
190	-	1100						
200	14	1250						
210	14	1150						
230	8	940						
240	6	780						
250	6	1270						
260	17	1810						
270	3	770						
280	-	1110						
290	-	920						
300	-	720						
310	-	740						
320	-	840						
330	10	620						
340	30	1100						
380	-	720						
390	5	570						
400	10	710						
410	7	720						
420	25	1120						
430	26	810						
440	8	490						
450	48	910						
455	-	800						
460	44	740						
470	42	760						
480	12	1040						
A3 - 100	-	1490						
A4 - 120	-	1740						

# BONDAR-CLEGG & COMPANY LTD.

## Geochemical Lab Report

REPORT NUMBER: 21 - 209

PAGE: 3

SAMPLE NUMBERS	Pb ppm	Ba ppm							
A4 - 132	-	1180							
A5 - 70	-	2040							
147	-	1480							
270	-	1270							
280	-	1060							
290	-	1240							
300	-	1070							
310	-	1150							
 <b>NOTE: 5-330 to 5-480 contain Sb</b>									