Summary Report
NAB CLAIMS
SCHAFT CREEK DEPOSIT Telegraph Creek, B. C.

Dolmage, Campbell \& Associates Ltd. July 15, 1972

Vancouver, Canada

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## Paramount Mining Ltd.

Summary Report

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## SUMMARY

Paramount Mining Ltd. owns a block of 49 mineral claims and fractions in the Liard Mining Division 40 miles south of Telegraph Creek in northwestern British Columbia. Access is by wheel aircraft to a mile-long gravel airstrip on the adjacent property owned by Liard Copper Mines Ltd. Both the Paramount and Liard Copper properties are presently under option to Helca Operating Company who are conducting drilling programs on both properties.

A major porphyry-type copper molybdenum (open pit) ore deposit has been outlined by diamond drilling on the Liard Copper property about 2000 feet south of the Paramount property. This ore deposit, termed here the Schaft Creek Orebody, has been reported to contain 240 million tons with an average grade of $0.42 \% \mathrm{Cu}$ and $0.036 \% \mathrm{MoS}_{2}$. An adjoining western orebody is centred along a steeply dipping fault breccia zone that extends northward and has been intersected by eight of the ten diamond drill holes that have been drilled inside the Paramount property.

The Schaft Creek Orebody occurs as a tabular-shaped deposit on the axis of synclinal fold in andesitic, (Permian?), tuffs and flows. The ore deposit lies about 3000 feet east of the contract of the diorite-quartz monzonite Hickman Batholith, an apophyse of the Coast Range Batholith. A regional fault belt about 4000-5000 feet in width, comprised of anastamosing major fault-shear zones, lies immediately east of the batholith contact and trends northward parallel to it. The Schaft Creek ore deposit lies within a block of intensely fractured andesites between converging major fault zones in the regional fault belt. The orebody is comprised of fine grained chalcopyrite, pyrite and molybdenite occurring as fillings and disseminations.

The Paramount Mining Ltd. claim group, north of the Schaft Creek Orebody, is on the strike of the intrusive contact, the fault zones and the andesite volcanics and contains essentially the same geological setting as that of the Schaft Creek Orebody, except for the fact that it lies on the western limb of the syncline.

Preliminary exploration of the Paramount property has produced a large area of coincident copper and molybdenum soil anomaly and I.P. anomaly in the vicinity of the ore intersections in the drilling done to date. This anomalous area overlies major converging faults in the presumably favourable andesitic volcanics and it is approximately equal in plan size to that of the Schaft Creek Orebody. Near the centre of the property a soil copper anomaly was detected around a small I.P. anomaly in an area of major faults and widespread copper showings. The north half of the property has only been geologically mapped but it encompasses a large number of copper showings in the andesite formation.

The Paramount property has room for three Schaft Creek type of orebodies. Already the diamond drill results to date indicate $90-100$ million tons of ore grading $0.337 \% \mathrm{Cu}$ and $0.047 \% \mathrm{MoS}_{2}$ at the south end of the property and this body is open to the north and east with the size potential to be a Schaft Creek type orebody.

In accordance with the existing agreement between Paramount Mining Ltd. and Hecla Mining Co. of Canada Ltd. the value to Paramount of the presently suggested ore would be about $\$ 70$ million. If this ore extends to become a Schaft Creek type orebody the value could be as much as $\$ 200$ million.

The presently warranted exploration of the Paramount property is estimated to cost approximately $\$ 500,000$.

## INTRODUCTION

At the request of the officers of Paramount Mining Ltd. the writer visited the Schaft Creek property on June 23-26, 1970, in order to form a basis for an evaluation of that portion of the property owned by Paramount Mining Ltd. In addition to his own observations the writer has had the use of the following references:

1. Nabs, Paramount Mining Ltd., W.G. Jeffrey, pp 29-30 Lode Metals in British Columbia, 1966, Department of Mines \& Petroleum Resources.
2. Final Geological Report - Nab Group - Stikine Area, Paramount Mining Ltd. by G. W. Grant - December 30, 1964.
3. Nab Group of Mineral Claims Liard Mining Division, British Columbia, Consultant's Report, by J. Sullivan - March 2,1966.
4. Geochemical Report on Nabs 1-34 Mineral Claims, Schaft Creek Area, Liard Mining Division, Adera Mining Ltd., by C.A.R. Lammie, P.Eng., July 15, 1966.
5. Report on Induced Polarization Survey of the Nabs Group, by Canadian Aero Mineral Surveys Limited, August 21 , 1966.
6. Summary Report - Schaft Creek Deposit, by D.R. Stelck and J.E. McKinney, Hecla Operating Company, November 30,1969.
7. Silver Standard Mines Ltd., News Releases, George Cross News Letter, March 11, 1969 and March 6, 1970.

Results of the examination and pertinent reference data were summarized in a report dated November 15, 1970. This present report is primarily an updating of that 1970 report, incorporating information obtained during the intervening period.

In view of the fact that only surface exploration and limited diamond drilling have been done on the Paramount property the evaluation of the property must be largely based on the relation of the property to the adjacent claims owned by Liard Copper Mines Ltd. on which a major porphyry-copper type deposit, (Schaft Creek Deposit), has been diamond drilled by Hecla Operating Company. For this reason this report discusses in detail the pertinent geological features of the Schaft Creek Deposit in relation to the general geological setting.

## LOCATION AND ACCESS: ( $57 \% 21^{\prime} \mathrm{N}, 131^{\circ} \mathrm{W}$ ) (Fig. I)

The Nabs claims of Paramount Mining Ltd. are located in Liard Mining Division in northwestern British Columbia just inside the eastern fringe of the Coast Range Mountains, east of the Alaska Panhandle. The nearest cities, although not necessarily the most accessible, are Prince Rupert and Terrace in British Columbia.

The property lies along the west slope of the north-trending ridge, 3000 feet in height, that separates Schaft Creek, on the west, from Mess Creek, on the east. Schaft Creek flows north to join Mess Creek 20 miles north of the property; Mess Creek flows another 20 miles north into the Stikine River at Telegraph Creek. The newly constructed Cassiar-Stewart Highway is 25 miles east of the property in the Iskut River Valley, (Fig. 2).

Present access to the property is by air. A good, gravel landing strip, one mile in length, has been built at the property and during the 1970 to 1972 summer seasons Harrison Airways Ltd. has been flying two trips per week by DC -3 direct from Vancouver to Schaft Creek. Flight time is approximately four hours non-stop. Trans Provincial Airlines Ltd. also fly an Otter aircraft to Schaft Creek from Terrace on a semi schedule basis.

The property can also be reached by horse trail from Telegraph Creek.

## HISTORY:

The northwestern portion of British Columbia is in the early stages of mineral exploration, the only significant mine production to date having been from Tulsequah, 100 miles northwest of Schaft Creek, and from the Portland Canal District, 100 miles to the southeast.

Significant exploration began in the Stikine area in the late 1950 's when Hudson Bay M.\& S.Co. Ltd. staked the Galore Creek deposit that was subsequently extensively drilled and proven to be an ore deposit in 1964-1966 by Kennco Ltd. This deposit lies 25 miles southwest of the Schaft Creek, over the spine of the Coast Range Mountains. During this period prospecting spread throughout the Telegraph Creek area and copper showings were discovered along the ridge between Mess Creek and Schaft Creek in 1957. The present Schaft Creek deposit of Liard Copper Mines Ltd. was originally staked at that time but the property was considerably expanded in 1959 and 1964 to essentially its present size. In 1966 Liard Copper Mines Ltd. was incorporated to include all of the companies holding various interest and claims on this deposit. In that year the property was optioned to American Smelting and Refining Company who explored the property with bulldozer trenching, induced polarization survey and 11,000 feet of diamond drilling in 24 holes. Aserco dropped their option and Hecla Operating Company then optioned the property and did 13,095 feet of diamond drilling in 9 holes in 1968, 15,501 feet in 9 holes in 1969, 32,575 feet in 26 holes in 1970 and 22,053 feet in 25 holes in 1971.

Three of the 1963 holes were on the Paramount property, five of the 1970 holes, and three of the 1971 holes for a total footage of 8,610 feet. The 1970 holes, numbers P5, P6A, and P7 were logged by the writer and his associate, Mr. R. S. Adamson.

The Nabs claims of Paramount Mining Ltd., located immediately north of the Schaft Creek deposit, (Fig. 3) were originally staked and prospected in 1964. In 1964 the southern half of the present property was investigated by a geochemical soils survey and an induced polarization survey. No further work was done until 1969 when Paramount Mining Ltd. optioned the property to Hecla Operating Company whereby the latter company agreed to expend a minimum of $\$ 37,500$ on the property in the 1969 field season and $\$ 50,000$ in 1971 etc. Under the terms of that agreement two diamond drill holes (P2, P3) were completed on the extreme southern end of the Paramount property in 1969 for a total of 2,167 feet. (Hole Pl was abandoned).

In 1970 Hecla conducted an induced potential survey over the southern half of the Paramount property and drilled three new holes, P5, P6A and P7 for a total of 3,756 feet on or onto the southern Paramount claims; hole P6 was abandoned. In 1971, three more holes were drilled, P8, P9 and P10 totalling 2,687 feet. Locations of all holes are shown on Figure 3.

## PROPERTY:

The Paramount property on Schaft Creek consists of the following claims and fractions that form one north northwest-trending group about 12,000 by 6,000 feet in size:

| Name |  | Record <br> Number | Tag <br> Number | In Good Standing to |
| :---: | :---: | :---: | :---: | :---: |
| GAV | 10 | 21680 | 655710 | Mar.23/84 |
|  | 11 | 21681 | 655711 | Mar.23/84 |
| NABS | 1 Fr . | 39482 | 34177 M | Aug.21/84 |
|  | 2 Fr . | 49383 | 43178 M | Aug.21/84 |
|  | 5 | 23972 | 735405 | Aug. 2/84 |
|  | 6 | 23973 | 735406 | Aug. 2/84 |
|  | 7 | 23974 | 735407 | Aug. 2/84 |
|  | 8 | 23975 | 735408 | Aug. 2/84 |
|  | 9 | 23976 | 735409 | Aug. 2/77 |
|  | 10 | 23977 | 735410 | Aug. 2/77 |
|  | 11 | 23978 | 735411 | Aug. 2/84 |
|  | 12 | 23979 | 735412 | Aug. 2/83 |
|  | 13 | 23980 | 735413 | Aug. 2/83 |
|  | 14 | 23981 | 735414 | Aug. 2/84 |
|  | 15 | 23982 | 735415 | Aug. 2/83 |
|  | 16 | 23983 | 735416 | Aug. 2/84 |
|  | 17 | 23984 | 735417 | Aug. 2/84 |
|  | 18 | 23985 | 735418 | Aug. 2/83 |
|  | 19 | 23986 | 735419 | Aug. 2/84 |
|  | 20 | 23987 | 735420 | Aug. 2/84 |
|  | 21 | 23988 | 735421 | Aug. 2/83 |
|  | 22 | 23989 | 735422 | Aug. 2/84 |
|  | 23 | 23990 | 735423 | Aug. 2/84 |
|  | 24 | 23991 | 735424 | Aug. 2/83 |
|  | 25 | 23992 | 735425 | Aug. 2/84 |
|  | 26 | 23993 | 735426 | Aug. 2/84 |
|  | 27 | 23994 | 735427 | Aug. 2/84 |
|  | 28 | 23995 | 735428 | Aug. 2/84 |
|  | 29 Fr . | 23996 | 735429 | Aug. 2/84 |


| Name |  | Record <br> Number | Tag <br> Number | In Good <br> Standing to |
| :--- | :--- | :--- | :--- | :--- |
| NABS 30 Fr. | 23997 | 735430 | Aug. 2/84 |  |
|  | 31 | 23998 | 735431 | Aug. 2/84 |
| 32 | 23999 | 735432 | Aug. 2/84 |  |
| 33 | 24000 | 735433 | Aug. 2/84 |  |
|  | 34 | 24001 | 735434 | Aug. 2/84 |
| RUM | 2 | 39485 |  |  |
|  | 5 | 39487 | 34182 | Aug.21/83 |
|  | 6 | 39488 | 34185 | Aug.21/83 |
|  | 8 | 39490 |  | Aug.21/83 |
| EMU | 8 | 39401 |  | Aug.21/83 |
|  |  |  |  |  |
| BARB | $1-4$ | $42609-12$ | $780497-500$ | Mar. 6/84 |
|  | $5-7$ | $42613-15$ | $669128-130$ | Mar. 6/84 |
|  | 8 | 42616 | 833279 | Mar. 6/84 |
|  | $9-10$ | $42617-18$ | $780423-424$ | Mar. 6/84 |

The western third of the property is underlain by forested mountain talus and valley floor of unknown overburden depths. The eastern two thirds of the property is underlain by the upper slopes of the $U$-shaped valley wall which is largely above timberline and comprised of steep rock bluffs and grass slopes and terraces. Snowfall is heavy in the area but winter temperatures are relatively mild for Canada.

## GEOLOGICAL SETTING

The Schaft Creek property of Paramount Mining Lfd. lies on the eastern edge of what is known as the Hickman Batholith. The Hickman Batholith is actually an eastern apophyse of the main Coast Range Batholith that underlies the coastal mountains of British Columbia and the Alaska Panhandle. In the Stikine area the eastern border of the Coast Range Batholith is very irregular and is characterized by numerous outlying intrusive stocks and apophyses, (Fig. 2).

Prophyry-type copper-molybdenum deposits have been discovered and explored along the eastern edge of the Coast Range Batholith from the Yukon to the Highland Valley in southern British Columbia but, with the exception of the Highland Valley deposits, access to this region is very limited and exploration is still in its early stages. All of the known copper and/or molybdenum deposits are associated with apophyses or stocks that stand out from the eastern contact of the main batholith. In the Stikine area the two proven deposits are (1) the Galore Creek copper deposit of Kennco Ltd. on the eastern edge of the main body of the batholith 25 miles southwest of the Paramount Property, and (2) the Schaft Creek copper molybdenum deposit of Liard Copper Mines Ltd. on the eastern edge of the Hickman Batholith (apophyse) adjacent to the Paramount Property.

The rock formations along Schaft and Mess Creeks that are in contact with the batholith are volcanic rocks that are tentatively classified as Permian or older in age. These volcanic formations are folded near the intrusive contact but are generally only gently dipping further to the east.

The structural trend of the formations in the Stikine area is northwestward and this is reflected by the known regional faults, all of which strike north-northwest and all of which are vertical to steeply dipping. Most of the known faults occur in the intruded rocks either along or parallel to intrusive contacts, (Fig. 2).

## PROPERTY GEOLOGY:

Intrusives: The eastern contact of the Hickman Batholith generally runs along the west side of the Schaft Creek, where the intrusive is well exposed in cliffs and bluffs; however, in the vicinity of the Schaft Creek deposit the intrusive contact bulges to the east and crops out along the lower slopes of the ridge that separates Schaft and Mess Creeks. In this area the intrusive ranges in composition from diorite to quartz
monzonite and is generally a massive, medium crystalline pinkish-white rock with local deficiencies in mafic mineral content. Most of the intrusive shown in Figure 3 of this report is actually covered with deep overburden and its presence is inferred from scattered outcrops and drill hole intersections.

North-trending dioritic dikes also cut the volcanics in this area.
Volcanics: The volcanic formations in the Schaft Creek area are folded into a relatively tight, north-trending, north-plunging syncline whose axis diverges slightly from the intrusive contact and lies about a mile east of the contact. The Schaft Creek Deposit lies in the volcanic rocks practically on this synclinal axis. The volcanic rocks are readily divisible by colour into two general formations. The uppermost rocks are comprised of purple basaltic flows, tuffs and agglomerates and the underlying rocks are grey-green flows, tuffs and agglomerates. The contact between the two formations is sharp and appears to be conformable. The Schaft Creek Deposit lies wholly within the underlying grey-green volcanics.

The grey-green volcanics are generally latite to andesite in composition and some are porphyritic, suggesting a possible hypabyssal origin. Most of the fragmentalvolcanics are crystal tuffs with common gradations to agglomerates. These volcanic rocks have been termed "andesites" in the field and this terminology will be retained in this report.

Adjacent to the intrusive the andesites are locally granitized and intricately laced by migmatitic layers and lenses. The extent of this contact zone has not been well defined but holes drilled near the intrusive have intersected these metavolcanic and migmatitic rocks so intermixed that specific identification becomes difficult.

Faults: The rocks underlying the Schaft Creek properties are traversed in a northsouth direction by lineaments that are readily discernible in aerial photos and which have been correlated with fault zone exposures in outcrops by Dr. Jeffrey of the B.C. Department of Mines. (The geology shown on Figure 3 of this report is derived largely from Dr. Jeffrey's published map of the properties). The outcrop exposures reported by Dr. Jeffrey consist of "clay-gouge and sheared and shattered iron-stained zones", which correspond to similar zones intersected by many of the diamond drill holes drilled in the western portion of the Schaft Creek Deposit. These major fault (shear)
zones dip steeply eastward and locally broaden into fault breccia zones, particularly in the vicinity of the diorite - quartz - monzonite contact zones.

As shown in Figure 3, three principal north-trending faults in the volcanic rocks curve around the bulge of the intrusive body where it crosses Schaft Creek. These three faults embrace a zone of anastamosing major and minor faults that is about 4000 feet in width and at least 4 miles in length. The east boundary fault is close to and parallel to the cxis of the syncline that passes through the centre of the saddle in the ridge. The eastern limit of the Schaft Creek Orebody lies about 500 feet east of this fault and parallel to it. The west boundary fault is parallel to and close to the intrusive contact and locally forms the confact. At least two major transverse shear faults connect the two boundary faults; one trends along the western edge of the ore deposit and has a wide breccia zone developed along it, which is richly mineralized in the deposit.

The andesitic volcanic rocks in the area of the ore deposit between the major boundary faul ts, are intensely fractured, sheared and finely veined by calcite, quartz, epidote, chlorite, talc and rather minor potash feldspar. Pervasive brownish fine crystalline potash feldspathization of the volcanic rocks is common in the vicinity of the fault zones. Pervasive rock alterations in the area of the orebodies are sericitization, carbonatization and chloritization.

The drilled Schaft Creek Orebody occurs in a block of intensely fractured andesitic volcanic rocks in the vicinity of converging major fault-shear zones.

## ECONOMIC GEOLOGY

One orebody, the Schaft Creek Deposit, has been proven on the Liard Copper Mines property by surface diamond drilling. In view of the fact that this deposit is the only orebody thus far discovered in the area it comprises a criteria in the search for other orebodies, hence it is described in some detail below:

## (A) SCHAFT CREEK OREBODY

The Schaft Creek deposit is formed by pervasive mineralization of fractured andesitic rocks by chalcopyrite, bornite, molybdenite and pyrite. The sulphide minerals are generally fine grained and occur both as fracture fillings and as discrete disseminations in the wall rocks.

Mineralogy: The writer examined two polished sections made from a high grade section of DDH 37, 905 ft . This examination revealed the principal sulphides to be chalcopyrite and pyrite, with local high concentrations of molybdenite. Also identified was a probable titanium oxide, (perovskite), and accessory quantities of tetrahedrite and sphalerite.

Two distinct grain-size populations are present. (1) coarse irregular shaped masses of chalcopyrite and crystals of pyrite, all ranging in size from 0.5 to 1.0 cm ., and (2) fine-grained chalcopyrite plus molybdenite in grains ranging from a few microns to 1 mm in size. Chalcopyrite constitutes about 80 percent of the fine sulphides and is evidently earlier than the molybdenite whose emplacement appears to have been controlled by late microfractures. The tetrahedrite is present in the polisections in very minor amounts nevertheless it suggests that in a large tonnage operation the concentrate may contain economic biproduct quantities of silver.

Form: The general form of the main orebody is a flat-lying tabular pod about 4500 feet in length, 3000 feet in width and 700 feet in vertical thickness. Its flat-lying configuration is of interest in that it corresponds generally with the location of the synclinal axis. The writer feels that the folded form of the andesites may have exerted some control on the orebody but that this control would be of a secondary nature in that it possibly influenced the shape of the orebody but not its emplacement. The primary
control of this orebody is probably more tectonic than stratigraphic in that it is located in a block of rock that has been relatively intensely fractured due to its position in an area of converging major faults within or close to the overall regional fault zone. This type of orebody control is characteristic of most of the porphyrytype copper and/or molybdenum deposits in the North and South American cordillera. They all occur firstly along or near a major geological plumbing system along which the metal-bearing solutions have been able to pass through the crustal rocks. Such plumbing systems generally consist of steep regional faults and trend parallel to batholith contacts or through a series of batholithic outliers. In most cases they are probably related to a primordial crustal weak zone along which the intrusive bodies were emplaced earlier. Deposition of the sulphide minerals along these plumbing systems is a function of temperature-pressure insofar as the general depths of the deposition is concemed but concentrated deposition, to form orebodies, is more commonly a function of reaction rates in the flow system. In this concept, which has been well demonstrated, metal-bearing acid hydrothermal solutions may travel long distances at high temperatures through igneous rocks without losing their reactive capacity if the fractures through which they flow, (the primary plumbing system), have widths greater than 1 mm . If these solutions encounter a brecciated or sheeted zone in which the solution contact surface area is much greater than that in the open fractures of the primary plumbing system, the solution is likely to react rapidly with its environment, causing intense alteration of wall rocks and/or deposition of ore minerals in the locus of the ramifying fracture-breccia system. This would seem to fit the situation of the Schaft Creek orebody wherein the primary plumbing system is the zone of anastomosing faults that follows the east contact of the Hickman Batholith, and the locus of deposition is the block of brecciated, intensely fractured rock between two converging major faults, (Fig.3).

An important part of the Schaft Creek deposit is what is known as the Western Zone. This is a richly mineralized steeply dipping zone of fault breccia and mylonite that has been formed along the northwesterly trending transverse fault that curves northwestward from Wolverine Creek and enters the Paramount property at about drill hole P6, (Fig. 3).

Tonnage and grade: The published figures for the Schaft Creek deposit, (Mar. 1970), are:

0.42\% Cu $0.80 \% \mathrm{Cu}$

Main orebody Breccia zone - (?)

240,000,000 tons @ 10,000,000 tons @ (?)

At Canadian prices of $50 \mathrm{c} / \mathrm{lb}$. copper and $\$ 1.70 / \mathrm{lb}$. molybdenum, and using a possible $90 \%$ recovery for both minerals the gross recoverable value of the above ore is $\$ 4.88 /$ ton. The large amount of reserves available will ensure long life for a major operation, i.e., about 15 years @ 50,000 +/day.

In addition to the above orebody, the potential on the Paramount and Liard Copper properties for the occurence of additional orebodies can be considered to be geologically good.

## (B) PROPERTY OF PARAMOUNT MINING LTD.

The Paramount claims cover essentially the same geological environment as that in which the Schaft Creek orebody is situated to the south, (Fig. 3). The Paramount property includes some of the diorite-quartz-monzonite intrusive along its western edge and extends eastward well into the purple volcanic formation near the crest of the ridge. The economic possibilities for the property would appear to fall into three general categories: (1) the occurrence of relatively small orebodies which could be mined by openpit or underground methods and shipped to the Schaft Creek Orebody mill, (2) the occurrence of other deposits of the same general size and grade as the Schaft Creek Orebody, and (3) the extension of the Schaft Creek Orebody onto Paramount Ground. The available data which indicates that one or more of these eventualities is a good possibility are discussed below:

## GEOLOGY:

The presumably favourable andesitic host rock formation and the contact of the diorite-quartz monozite intrusive body are included in the Paramount property for a strike length of nearly 3 miles. The only obvious geological difference between
this geological setting and that to the south, in which the Schaft Creek orebody occurs, is that the Paramount claims lie along the western limb of the synclinal fold in the volcanics whereas the ore zone to the south more or less straddles the axis or keel of this syncline.

The western portion of the major fault belt is included for the full length of the property and becomes complexly branched near the centre of the property with two major faults converging at the south end, (Fig. 3). Thus, if it is accepted that this zone of major faulting represents the primary plumbing system for metal-bearing solutions then any loci of converging faults or brecciation along the system on the Paramount property would comprise potential ore zones.

Malachite stain is visible on the bluffs in many places in the andesitic rocks on the Paramount property. Near DDH P2 bulldozer stripping has exposed bedrock consisting of breccia type ore that is heavily impregnated with pyrite and chalcopyrite. This particular outcrop would appear to be the projection from the south of the Western Breccia ore zone. The Hecla geologists have mapped the chalcopyrite and malachite showings on both the Paramount and the Liard Copper properties and these are shown on Fig. 3. It is of interest that the largest concentration of such mapped copper exposures occurs along the fault zone - intrusive contact on the Paramount property, particularly in the northern half.

Thus, on the Paramount property there is a juxtaposition of the following geological features that are analogous to those in the vicinity of the Schaft Creek Orebody and are therefore considered to be encouraging for the possible occurrence of other orebodies.

1. Faulted and brecciated andesitic rocks.
2. Intrusive contact.
3. Major faulting.
4. Relatively abundant copper showings.

## GEOPHYSICS:

The southem half of the Paramount property was covered by an I.P. survey in 1966 and a large anomaly was detected at the south end of the property and a relatively small one near the centre of the property, (Fig. 4). the large anomaly is some $4000 \times 2000$ feet in plan size and is open at the southem boundary of the claims. It is underlain by andesitic rocks and covers the areas of converging major faults. The smaller I.P. anomaly is only a few hundred feet in diameter but occurs in the centre of an extensive copper soil anomaly.

## GEOCHEMISTRY:

The southern half of the Paramount property was also explored in 1966 by a geochemical soils survey for copper and molybdenum, the general results of which are shown in Figure 4.

The copper survey detected three anomalous areas of soil:

1. An irregular shaped anomaly at the south end of the property, somewhat downhill from the I.P. anomaly. Although irregular in shape this anomaly contains relatively high copper values.
2. A large area near the centre of the property. This anomaly contains moderate copper values, is centred on the small I.P. anomaly, and is open to the north.
3. A small, weak, irregular anomaly that is open off the property to the east.

The molybdenum survey detected one weak, ( $3-10 \mathrm{ppm}$ ), anomaly at the southern end of the property, within the large I.P. anomaly.

The two areas of most interest outlined by the soils survey are the large copper anomaly at the centre of the property and the coincident copper and molybdenum anomalies at the south end of the property. The central anomaly is of interest because it includes the western faults, a cluster of copper showings and the small I.P. anomaly. The southem anomaly is of great interest because it is coincident with the large I.P. anomaly as well as the convergence of the major faults. Both the southem and central anomalous areas are comparable in size to the plan area of the Schaft Creek orebody.

## DIAMOND DRILLING: (Figs. 5, 6, 7, 8)

In 1969 two holes, P2 and P3 were drilled at the extreme south end of the Paramount property. Both holes were drilled to the west at -45 deg. across the northern projection of the Western Breccia Ore Zone and both traverse the complex contact zone of interfingered intrusive and metavolcanic layers, with granitization and migmatization locally much in evidence.

The western hole, P2, penetrates the contact of the main intrusive body and only about 40 percent of its length is in (meta) volcanic rocks. Copper values are negligible in P2; most of the hole assaying $.05-.06 \% \mathrm{Cu}$ with local highs of 0.4 ( 70 feet ) and $0.3(40 \mathrm{feet}) \% \mathrm{Cu}$. The bottom 150 feet of hole, in the intrusive proper, is essentially barren.

The eastern hole, P3, intersected the northem projection of the West Breccia Ore Zone at a depth of about 150 feet; beyond that depth the hole was largely in intrusive layers, all of which dip steeply eastward. The first 380 feet of hole P3 is well mineralized, averaging $0.43 \% \mathrm{Cu}$ and $0.077 \% \mathrm{MoS}_{2}$, approximately the grade of the main Schaft Creek orebody to the south. From the fault breccia zone to the end of the hole at 1200 feet metal values are submarginal, averaging about $0.10 \% \mathrm{Cu}$ and $0.005 \% \mathrm{MoS}_{2}$.

In 1970 three additional holes P4, P5, P6A, were drilled on the P2-P3 section, (Fig. 5). (Hole P6 was also drilled but abandoned after a short distance). Of these, P5 was collared at P3 and drilled at $\mathbf{- 7 5}$ deg. westward through the Breccia Zone to a depth of 1137 feet. Hole P6A was collared at the Paramount boundary, 400 feet east of P5, and drilled at -75 degrees west. Hole P4 was collared 400 feet
east of P6A on Liard ground and drilled west at -45 degrees but was abandoned at 447 feet in caving rock and overburden. (The depth of overburden on this section is roughly 100 feet.) In addition, a fourth hole, P7, was collared 700 feet north of P5 and drilled parallel to it to a depth of 1567 feet (Fig. 6). The locations of these holes are shown on Figs. 3 and 4 . of this report and geological and assay sections through them are presented in Figs. 5 and 6. Geological logs of holes P5, P6A and P7 are appended to the report.

In 1971 a further three holes, (P8, P9, P10), were drilled in the southeast corner of the Paramount property, south of all previous holes. P8 was collared 50 feet east and 275 feet south of P3 - P5 and drilled vertically through the Breccia Zone. P9 was collared approximately 275 feet west and 275 feet south of P3 - P5 where it also drilled vertically through the Breccia Zone. Locations of these holes are shown on Figs. 3 and 4 and geological and assay sections are presented in Figs. 7 and 8. Geological logs are not available but sample and assay details are appended.

## RESULTS:

Three holes drilled in 1970 intersected very extensive lengths of coppermolybdenum mineralization, P5 and P7 in very low grade and P6A in ore grade. The complete assays for the 1970 holes are recorded at the end of each of the drill logs appended to this report.

All of the holes drilled in 1971 intersected considerable ore grade mineralization and P8 bottomed in good grade material ( $0.63 \% \mathrm{Cu}$ equivalent). P9 and Pl0 were still weakly mineralized at hole bottoms.

As shown in Figs. 5, 6, 7 and 8 the sulphide mineralization intersected on the Paramount claims is concentrated in the fractured, altered volcanic rocks within and east of the regional fault zone termed the West Breccia Ore Zone on the Liard Copper claims to the south. The richest concentration is within the West Breccia Zone and in the most sheared and fractured and altered rock east of it. There is insignificant mineralization west of the zone, within the more competent, less fractured (unfaulted)intrusive and metavolcanic rocks toward the main intrusive contact. Within and east of the West Breccia Zone the volcanic rocks are altered by hydrothermal bleaching that has been accompanied by the replacement of the
mafic minerals by sericite, carbonate, some K-feldspar and chlorite. Chalcopyrite, pyrite and bomite occur in the more fractured rock as fine grained disseminations and as scattered fracture fillings; in contrast, the molybdenite is generally confined to smears on fracture planes or in quartz veinlets. Recent experience with the Adanac molybdenite deposit near Atlin indicates that molybdenite occurring primarily on fractures suffers a loss in NQ core drilling in excess of $20 \%$. The Paramount drill holes have experienced considerable core loss in the most fractured-sheared (and best mineralized) rock and it is most likely that the drill core copper-molybdenum assays are all lower than the true grade.

GRADE:
The average grades of the better mineralized portions of the drill holes are:

| Hole | Interval | Length | Average Assay |
| :---: | :---: | :---: | :---: |
| P3 | - | 3801 | $0.32 \% \mathrm{Cu}, 0.077 \% \mathrm{MoS}_{2}$ |
| P5 | 120'-900' | $780^{\prime}$ | $0.37 \% \mathrm{Cu}, 0.032 \% \mathrm{MoS}_{2}$ |
| P6A | 190'-1152 | 962' | $0.37 \% \mathrm{Cu}, 0.052 \% \mathrm{MoS}_{2}$ |
| P7 | 210'-830' | $620^{\prime}$ | 0.30\%Cu, $0.040 \% \mathrm{MoS}_{2}$ |
| P8 | 120'-780' | $660^{\prime}$ | $0.33 \% \mathrm{Cu}, 0.043 \% \mathrm{MoS}_{2}$ |
|  | 980'-1161 | 1811 | 0.30\%Cu, $0.089 \% \mathrm{MoS}_{2}$ |
| P9 | 12'-780' | 768' | 0.36\%Cu, $0.047 \% \mathrm{MoS}_{2}$ |
| P10 | 53'-594' | 541 | $0.20 \% \mathrm{Cu}, 0.047 \% \mathrm{MoS}_{2}$ |

The weighted average of these intersections is $0.337 \% \mathrm{Cu}$ and $0.047 \%$ $\mathrm{MoS}_{2}(0.028 \% \mathrm{Mo})$. If it is assumed that the actual grades will be $5 \%$ higher for copper and $20 \%$ higher for molybdenum and that recoveries for both will be $90 \%$ the recoverable grade would be $0.319 \% \mathrm{Cu}$ and $0.051 \% \mathrm{MoS}_{2}$.

Using metal prices of $\$ 0.50 / \mathrm{lb}$. for copper and $\$ 1.70 / \mathrm{lb}$. for molybdenum the current gross recoverable value of the Paramount ore is $\$ 4.23 /$ ton. Current operating costs for open-pit operations in excess of 15,000 tons per day in southern British Columbia are about $\$ 2.00 /$ ton, therefore the net operating profit for the Paramount material would probably be at least $\$ 2.20 /$ ton.

TONNAGE:
For purposes of preliminary assessment it is of interest to estimate the tonnage of the above mineralized rock that has been indicated by the several diamond drill holes. Using the cross-sections and plan, the average area to a depth of 1200 feet of the mineralized zone on Paramount ground is approximately $1,000 \mathrm{sq}$. ft. or about 90,000 tons per vertical foot. Allowing for some waste sections the tonnage encompassed to a depth of 1200 feet is in the range of $90-$ 100 million tons.

The mineralized zone is open to the north and to the east (the Paramount boundary diverges to the east going north, Fig.3).

## CONCLUSIONS

The geological setting of the Nabs claims of Paramount Mining Ltd. is essentially the same as that to the south in the vicinity of the Schaft Creek Orebody except for the fact that the axis of the synclinal fold of the volcanic formations passes through the orebody but does not cross the Paramount property. The other geological features to which the orebody is, or may be, related are all repeated on the Paramount property in the same general dimensions and character as at the orebody. These include:

1. Ostensibly favourable andesitic host rocks.
2. Nearby intrusive contact.
3. Major zone of converging faults.
4. Extensive close fracturing and hydrothermal alteration of andesites.
5. Many surface exposures of copper mineralization in the andesites.

In addition to these general geological features are the encouraging results of the exploration that has been done on the Paramount property, namely:

1. Some 90-100 million tons of "ore" mineralized rock indicated by 8 diamond drill holes drilled to date, and open to the east and north.
2. The juxtaposition of strong copper and molybdenum geochemical anomalies with a good I.P. anomaly at and north of P3 over an area measuring $2000 \times$ 4000 feet, approximately the plan size of the Schaft Creek orebody.
3. The occurrence of an extensive copper soil anomaly near the centre of the Paramount property, in an area of major faults, extensive copper showings and a weak I.P. anomaly.
4. An area $4000 \times 6000$ feet representing the unexplored north end of the property.

## EVALUATION:

At this time, when the ultimate value of the Schaft Creek orebody has still to be determined, it is impossible to assign a meaningful monetary value on the Nabs Property of Paramount Mining Ltd.; however, a preliminary assessment can be made of the possible potential of the Paramount property.

It is estimated that the operating profit for the 90-100 million tons of possible ore; presently indicated on the Paramount property could be as much as $\$ 220$ million. If this ore continues to the north it is conceivable that the equivalent of another Schaft Creek orebody will be developed on the Paramount property. Spatially there is room on the Paramount property for three Schaft Creek type of orebodies; geologically, geophysically and geochemically there is good evidence for one such possible orebody at the south end of the property, some evidence for a possible one in the centre of the property and unknown potential at the north end. The drill hole evidence to date suggests that there is an orebody of major size, ( 100 million tons), at the south end of the property.

The contract between Hecla Mining Co. of Canada and Paramount Mining Ltd. stipulates that Paramount will be paid $30 \%$ of the net proceeds for any ore mineral from the Paramount claims. If a Schaft Creek orebody occurs on the Paramount property the eventual value to Paramount Mining Ltd. would be about $\$ 200$ million.

## RECOMMENDATIONS:

If the Paramount property is considered as an entity separate from the Liard Copper property it would warrant the following work, based on existing data:

1. Diamond drilling to extend and sample the 15,000 feet ore in the present drill holes.
2. Diamond drilling to determine relevance of
5,000 feet geochemical and geophysical anomalies.
3. Geochemical and I.P. surveys of north end of property. (Acknowledging the downhill migration of soil anomalies in such steep terrain, they may still be useful).
4. Reconnaissance exploratory drilling of remainder of property.

The total cost of such a program is roughly estimated to be about $\$ 500,000$ and would be near the minimum required to test the whole property for possible orebodies of the size of the existing Schaft Creek Orebody to the south. If such a program discovered new ore additional drilling would of course be necessary to fully evaluate the property.

Respectfully submitted, DOLMAGE CAMPBELL \& ASSOCIATES LTD.











## APPENDIX

## Diamond drill core logs and assays for DDH P5, P6A and P7.

Diamond drill assays for DDH P8,
P9, P10




DH P6A
Page
description

White to pink med. crystalline feldspathic rock.
Core badly broken, sheared parallel to hole, local smears of chalcopyrite and moly on fractures.
Abundant cherty tan-coloured sections ( $K$-spar probably).
187-210 shear zone gauge locally (very poor core recovery). $258-365$ shear zone, broken core, section of gouge. $50 \%$ core recovery, fair amount of hematite.

From 365-426 mineralization increases, generally heavily fractured with chalcopyrite and moly common on fracture planes $10^{\circ}-45^{\circ}$ to core.
From 420-440 rock becomes holo-crystalline pinkish, syenitic, mineralization decreases, moly on flat fractures, very fine chalcopyrite on steep fractures. From 440 intense moly on all fractures. No chalcopyrite.

Uniform, pale grey, locally pink, massive, medium crystalline rock that is hard, blocky fractured, locally cherty. Occasional heavy chlorite shears at very low angle to hole. Rock is either "granitized" volcanic or hybrid monzonite.

- 515 to 522 black soft lamprophyre dyke at $50^{\circ}$ to hole axis.
- Mineralization : (470-580) Abundant moly on shear fractures and slip planes, generally $45-10^{\circ}$ to hole.
Very fine grained chalcopyrite disseminated throughout rock.
(580-636) Minor moly but relatively abundant very fine grained chalcopyrite.

| Interval | Length | \%Cu | \%MoS |  |
| :---: | :---: | :---: | :---: | :---: |
| $140-190$ | 50 | 0.03 | 0.012 | \%Cu Equivalent |
| $190-370$ | 180 | 0.14 | 0.006 | 0.07 |
| $370-440$ | 70 | 0.20 | 0.030 | 0.16 |
| $440-530$ | 90 | 0.19 | 0.045 | 0.29 |
| $530-710$ | 180 | 0.34 | 0.076 | 0.33 |
| $710-780$ | 70 | 0.66 | 0.112 | 0.57 |
| $780-1070$ | 290 | 0.55 | 0.063 | 1.00 |
| $1070-1152$ | 82 | 0.39 | 0.040 | 0.74 |

Diamond Drill Lc
PARAMOUNT SCHAFT CREEK
DH DHP7
Page 1

|  | AGE | ROCK TYPE | DESCRIPTION | CORE LOSS |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: |
| FROM | 10 |  |  | FROM | ro | LOST |
| 0 | 203 | OVERBURDEN |  |  |  |  |
| 203 | 310 | META-VOLCANIC | - Medium crystalline, hard, massive pale gray-green with pinkish (monzonitic ?) |  |  |  |
|  |  |  | sections. |  |  |  |
|  |  |  | Core very badly broken, recovery poort to moderate. |  |  |  |
|  |  |  | (231-234) Fault - fractured rock and gouge. |  |  |  |
|  |  |  | Mineralization: Finely disseminated chalcopyrite and pyrite throughout. |  |  |  |
|  |  |  | Molybdenite smeared on fracture planes and in quartz veinlets. |  |  |  |
|  |  |  |  |  |  |  |
| 310- | 637 | ALTERED VOLCAN | - Pale green-gray, hard, massive, locally extensively pink tinted (granitized ?). |  |  |  |
|  |  |  | Intensely fractured but core recovery better than preceding section. Rock is |  |  |  |
|  |  |  | probably "bleached" by replacement of mafics by sericite and carbonate (?), |  |  |  |
|  |  |  | appears to be locally granitized. |  |  |  |
|  |  |  | Mineralization: Finely disseminated chalcopyrite but more molybdenite on |  |  |  |
|  |  |  | fractures than in preceding core. |  |  |  |
|  |  |  | Basalt dikes - green-black, hard, massive, aphanitic. (367-384, 402-412, |  |  |  |
| , |  |  | 426-434). |  |  |  |
|  |  |  | Fault - (480) - Gouge (6'). |  |  |  |
|  |  |  |  |  |  |  |
| 637 | 674 | FAULT ZONE | - Major structure, includes 15 ft . of gouge and intense brecciation. Sheared, |  |  |  |
|  |  |  | bleached volcanics. |  |  |  |
|  |  |  | Mineralization: Disseminated chalcopyrite very common in brecciated sections. |  |  |  |
|  |  |  |  |  |  |  |
| 674 | 755 | BRECCIA ZONE | - Intensely brecciated with variety of rock types in the fragments (altered |  |  |  |
|  |  |  | volcanics and meta (granitized) volcanics), "bleaching" of fragments may have |  |  |  |
|  |  |  | been pre-brecciation. |  |  |  |
|  |  |  |  |  |  |  |
| 755 | 777 | META-VOLCANICS | - Fine to medium crystalline, hard, massive, grayish-pink with "granitized" appea |  |  |  |
|  |  |  | very pronounced locally, i.e., (755-763) could be monzonite. |  |  |  |
|  |  |  | (763-767) - Basalt dike. |  |  |  |
|  |  |  | Mineralization: Very finely disseminated chalcopyrite and moly on fractures. |  |  |  |
|  |  |  | Core broken on fractures has probably resulted in considerable moly loss. |  |  |  |
|  |  |  |  |  |  |  |
|  |  |  |  |  |  |  |
|  |  |  |  |  |  |  |
|  |  | . |  |  |  |  |
|  |  |  |  |  |  |  |
|  |  |  |  |  |  |  |



| Start: | July 11, 1971 | Coordinates: | 27501.35N |
| :---: | :---: | :---: | :---: |
| Finish: | July 20, 1971 |  | 7966.46E |
| Core Size: | NQ | Elevation: | 3,110.7' |
| Core Recovery: | $98 \%$ | Inclination: | $90^{\circ}$ |
| Sample: | Split Core | Bearing: | N $0^{\circ} 0^{\prime} \mathrm{E}$ |
| Directional Sur | ey: See last page | Length: | 1.161' |


| Core <br> Sample <br> Number | Hole Interval (feet) |  | Core Recovery (percent) | Copper <br> (percent) | Molybdenite (percent) | \% Copper Equivalent $\left(8 \mathrm{Cu}+3 \times \mathrm{HOS}_{2}\right.$ |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: | :---: |
| 66426 | 120 - | 130 | 77 | 0.28 | 0.003 | 0.29 |  |
| 66427 | 130 - | 140 | 99 | 0.20 | 0.007 | 0.22 |  |
| 66428 | 140 - | 150 | 111 | 0.30 | 0.021 | 0.36 |  |
| 66429 | 150 - | 160 . | 94 | 0.15 | 0.015 | 0.20 |  |
| 66430 | $160-$ | 170 | 89 | 0.29 | 0.018 | 0.34 |  |
| 66431 | 170 - | 180 | 92 | 0.29 | 0.053 | 0.45 |  |
| 66432 | 180 - | 190 | 100 | 0.24 | 0.031 | 0.33 |  |
| 66433 | 190 - | 200 | 97 | 0.17 | 0.030 | 0.26 |  |
| 66434 | 200 - | 210 | 93 | 0.29 | 0.025 | 0.36 |  |
| 66435 | 210 - | 220 | 100 | 0.09 | 0.054 | 0.25 |  |
| 66436 | 220 - | 230 | 104 | 0.11 | 0.050 | 0.26 |  |
| 66437 | $230-$ | 240 | 106 | 0.09 | 0.023 | 0.16 |  |
| 66438 | 240 - | 250 | 84 | 0.22 | 0.020 | 0.28 |  |
| 66439 | $250-$ | 260 | 93 | 0.28 | 0.012 | 0.32 |  |
| 66440 | 260 - | 270 | 102 | 0.32 | 0.025 | 0.40 |  |
| 66441 | 270 - | 280 | 98 | 0.25 | 0.038 | 0.36 |  |
| 66442 | 280 - | 290 | 103 | 0.05 | 0.018 | 0.10 |  |
| 66443 | 290 - | 300 | 97 | 0.13 | 0.020 | 0.19 |  |
| 66444 | $300-$ | 310 | 99 | 0.13 | 0.006 | 0.15 |  |
| 66445 | 310 - | 320 | 99 | 0.11 | 0.011 | 0.14 |  |
| 66446 | 320 - | 330 | 98 | 0.33 | 0.037 | 0.44 | - |
| 66447 | $330-$ | 340 | 99 | 0.27 | 0.023 | 0.34 |  |
| 66448 | 340 - | 350 | 99 | 0.32 | 0.042 | 0.45 |  |
| 66449 | $350-$ | 360 | 100 | 0.52 | 0.046 | 0.66 |  |
| 66450 | 360 - | 370 | 100 | 0.40 | 0.050 | 0.55 |  |
| 66451 | 370 - | 380 | 100 | 0.47 | 0.068 | 0.67 |  |
| 66452 | 380 - | 390 | 99 | 0.48 | 0.075 | 0.70 |  |
| 66453 | 390 - | 400 | 98 | 0.34 | 0.036 | 0.45 |  |
| 66454 | 400 - | 410 | 100 | 0.37 | 0.096 | 0.66 |  |
| 66455 | 410 - | 420 | 100 | 0.47 | 0.077 | 0.70 |  |
| 66456 | 420 - | 430 | 99 | 0.29 | 0.120 | 0.65 |  |
| 66457 | 430 - | 440 | 100 | 0.22 | 0.120 | 0.58 |  |
| 66458 | 440 - | 450 | 97 | 0.19 | 0.086 | 0.45 |  |
| 66459 | $450-$ | 460 | 97 | 0.24 | $\cdots$ | 0.40 |  |
| 66460 | 460 - | 470 | 101 | 0.25 | 0.040 | 0.37 |  |
| 66461 | 470 - | 480 | 98 | 0.52 | 0.002 | 0.53 |  |
| 66452 | 480 - | 490 | 102 | 0.89 | 0.002 | 0.90 |  |
| 66463 | 490 - | 500 | 96 | 0.20 | 0.002 | 0.21 |  |
| 66464 | $500-$ | 510 | 105 | 0.41 | 0.002 | 0.42 |  |
| 65465 | $510-$ | 520 | 95 | 0.81 | 0.013 | 0.85 |  |
| 66466 | 520 - | 530 | 95 | 0.16 | 0.026 | 0.24 |  |
| 66467 | $530-$ | 540 | 101 | 0.26 | 0.069 | 0.47 |  |
| 66468 | 540 - | 550 | 95 | 0.33 | 0.048 | 0.47 |  |
| 66469 | $550-$ | 560 | 95 | 0.32 | 0.096 | 0.61 |  |
| 65470 | 560 - | 570 | 97 | 0.18 | 0.035 | 0.28 |  |
| $664 \% 1$ | 570 - | 580 | 98 | 0.21 | 0.029 | 0.30 |  |
| 66472 | $580-$ | 590 | 102 | 0.23 | 0.025 | 2. 30 |  |
| 66473 | $590-$ | 600 | 101 | 0.19 | 0.028 | 0.27 |  |
| 66474 | 600 - | 610 | 101 | 0.19 | 0.023 | 0.26 |  |
| 65475 | $610-$ | 620 | 97 | 0.08 | 0.088 | 0.34 |  |
| 66476 | 620 - | 630 | 99 | 0.29 | 0.092 | 0.57 |  |
| 65477 | $630-$ | 640 | 102 | 0.30 | 0.031 | 0.39 |  |
| 66478 | 640 - | 650 | 101 | 0.36 | 0.050 | 0.51 |  |
| 66479 | $650-$ | 660 | 101 | 0.42 | 0.071 | 0.65 |  |
| 66480 | $660-$ | 670 | - 100 | 0.42 | 0.063 | 0.62 |  |

DIAMOND DRILL HOLE P-8
(continued)

| Core <br> Sample <br> Number | Hole Interval (feet) | Core Recovery (percent). | Copper (percent) | Molybdenite (percent) | \% Copper Equivalent. $\left(\% \mathrm{Cu}+3 \times \% \mathrm{MOS}_{2}\right)$ |
| :---: | :---: | :---: | :---: | :---: | :---: |
| 66481 | 670-680 | 100 | 0.54 | 0.044 | 0.67 |
| 66482 | 680-690 | 100 | 0.55 | 0.063 | 0.74 |
| 66483 | 690-700 | 99 | 0.50 | 0.054 | 0.66 |
| 66484 | 700-710 | 100 | 0.19 | 0.025 | 0.26 |
| 66485 | $710-720$ | 101 | 0.57 | 0.036 | 0.68 |
| 66486 | $720-730$ | 97 | 0.64 | 0.038 | 0.75 |
| 66487 | 730-740 | 95 | 0.50 | 0.160 | 0.98 |
| 66488 | 740-750 | 86 | 0.33 | 0.031 | 0.42 |
| 66489 | 750-760 | 95 | 0.72 | 0.050 | 0.87 |
| 66490 | $760-770$ | 98 | 0.74 | 0.073 | 0.96 |
| 66491 | $770-780$ | 98 | 0.40 | 0.027 | 0.48 |
| 66492 | $780-790$ | 97 | 0.14 | 0.010 | 0.17 |
| 66493 | 790-800 | 102 | 0.19 | 0.010 | 0.22 |
| 66494 | 800-810 | 97 | 0.14 | 0.011 | 0.17 |
| 66495 | 810-820 | 98 | 0.08 | 0.012 | 0.12 |
| 66496 | $820-830$ | 97 | 0.06 | 0.012 | 0.10 |
| 66497 | $830-840$ | 98 | 0.03 | 0.013 | 0.07 |
| 66498 | $840-850$ | 96 | 0.03 | 0.007 | 0.05 |
| 66499 | $850-860$ | 100 | 0.04 | 0.004 | 0.05 |
| 66500 | $860-870$ | 96 | 0.06 | 0.005 | 0.08 |
| 80002 | $870-880$ | 91 | 0.06 | 0.014 | 0.10 |
| 80003 | $880-890$ | 104 | 0.03 | 0.005 | 0.04 |
| 80004 | 890-900 | 106 | 0.02 | 0.003 | 0.03 |
| 80005 | 900-910 | 102 | 0.01 | 0.005 | 0.02 |
| 80006 | 910-920 | 100 | 0.01 | 0.001 | 0.01 |
| 80007 | $920-930$ | 96 | 0.02 | 0.002 | 0.03 |
| 80008 | $930-940$ | 100 | 0.03 | 0.002 | 0.04 |
| 80009 | 940-950 | 98 | 0.03 | 0.002 | 0.04 |
| 80010 | $950-960$ | 95 | 0.06 | 0.003 | 0.07 |
| 80011 | $960-970$ | 95 | 0.06 | 0.002 | 0.07 |
| 80012 | 970 - 980 | 104 | 0.04 | 0.002 | 0.05 |
| 80013 | $980-990$ | 103 | 0.09 | 0.058 | 0.26 |
| 80014 | 990-1000 | 101 | 0.78 | 0.160 | 1.26 |
| 80015 | 1000-1010 | 98 | 0.21 | 0.047 | 0.35 |
| 80016 | 1010-1020 | 62 | 0.07 | 0.009 | 0.10 |
| 80017 | 1020-1030 | 98 | 0.34 | 0.050 | 0.49 |
| 80018 | 1030-1040 | 93 | 0.52 | 0.084 | 0.77 |
| 80019 | 1040-1050 | 93 | 0.40 | 0.120 | 0.76 |
| 80020 | 1050-1060 | 102 | 0.31 | 0.049 | 0.46 |
| 80021 | 1060-1070 | 104 | 0.30 | 0.140 | 0.72 |
| 80022 | 1070-1080 | 99 | 0.45 | 0.110 | 0.78 |
| 80023 | 1080-1090 | 101 | 0.26 | 0.069 | 0.47 |
| 80024 | 1090-1100 | 97 | 0.28 | 0.091 | 0.55 |
| 80025 | 1100-1110 | 101 | 0.32 | 0.120 | 0.68 |
| 80026 | 1110-1120 | 102 | 0.29 | 0.067 | 0.49 |
| 80027 | 1120-1130 | 102 | 0.12 | 0.090 | 0.39 |
| 80028 | 1130-1140 | 102 | 0.11 | 0.085 | $\because 0.37$ |
| 80029 | 1140-1150 | 100 | 0.24 | 0.130 | 0.63 |
| 80030 | 1150-1161 | 102 | 0.23 | 0.130 | 0.62 |

DIRECTIONAI SURVEY DATA
Hecla - (acid bottle \& transit)
Sperry-Sun - (multi-shot gyro)

| Location | Inclination | Bearing |
| :---: | :---: | :---: |
| Collar | $90^{\circ}$ | N $O^{\circ} O^{\prime \prime} \mathrm{E}$ |
| 300' | $86^{\circ}$ | NA |
| $600^{\prime}$ | $86^{\circ}$ | NA |
| 900' | $87^{\circ}$ | NA |
| 1160' | $86^{\circ}$ | NA |

Iocation Inclination Bearing

## DIAMOND DRILL HOLE P-9

| July 20, 1971 |  |  |  | Coordinates: | 27499.65N |
| :---: | :---: | :---: | :---: | :---: | :---: |
| Finish: July 27, 1971 |  |  |  |  | 7660.08E |
| Core Size: N |  |  |  | Elevation: | 3,034.8' |
| Core Recovery: 9 | ery: 95\% |  |  | Inclination: | $90^{\circ}$ |
| Sample: Sp |  | Core |  | Bearing: | NO $0^{\circ}{ }^{\prime \prime} \mathrm{E}$ |
| Directional Survey: See last page |  |  |  | Length: | 932 ' |
| Core | Hole | Core |  |  | \% Copper |
| Number | (feet) | (percent) | (percent) | (percent) | $\left(8 \mathrm{Cu}+3 \times 8 \mathrm{mos}_{2}\right)$ |
| 80031 | 12-20 | 45 | 0.32 | 0.014 | 0.36 |
| 80032 | 20-30 | 3 | 0.25 | 0.017 | 0.30 |
| 80033 | 30-40 | 34 | 0.48 | 0.046 | 0.62 |
| 80034 | 10-50 | 100 | 0.54 | 0.098 | 0.80 |
| 80035 | $50-60$ | 90 | 0.45 | 0.120 | 0.81 |
| 80036 | 60-70 | 91 | 0.18 | 0.012 | 0.22 |
| 80037 | $70-80$. | 94 | 0.42 | 0.027 | 0.50 |
| 80038 | 80-90 | 95 | 0.52 | 0.058 | 0.69 |
| 80039 | 90-100 | 97 | 0.52 | 0.040 | 0.64 |
| 80040 | 100-110 | 100 | 0.47 | 0.049 | 0.62 |
| 80041 | 110-120 | 63 | 0.38 | 0.120 | 0.74 |
| 80042 | 120-130 | 93 | 0.48 | 0.063 | 0.67 |
| 80043 | 130-140 | 98 | 0.40 | 0.063 | 0.59 |
| 80044 | 140-150 | 90 | 0.12 | 0.034 | 0.22 |
| 80045 | 150-160 | 88 | 0.22 | 0.033 | 0.32 |
| 80046 | 160-170 | 85 | 0.22 | 0.041 | 0.34 |
| 80047 | 170-180 | 86 | 0.19 | 0.027 | 0.27 |
| 80048 | 180-190 | 94 | 0.10 | 0.014 | 0.14 |
| 80049 | 190-200 | 96 | 0.07 | 0.014 | 0.11 |
| 80050 | 200-210 | 56 | 0.16 | 0.018 | 0.21 |
| 80051 | 210-220 | 88 | 0.09 | 0.005 | 0.10 |
| 80052 | 220-230 | 95 | 0.01 | $<0.001$ | 0.01 |
| 80053 | 230-240 | 95 | 0.27 | 0.020 | 0.33 |
| 80054 | 240-250 | 99 | 0.18 | 0.006 | 0.20 |
| 80055 | 250-260 | 102 | 0.11 | 0.026 | 0.19 |
| 80056 | 260-270 | 99 | 0.15 | 0.019 | 0.21 |
| 80057 | 270-280 | 102 | 0.37 | 0.065 | 0.56 |
| 80058 | 280-290 | 101 | 0.37 | 0.069 | 0.58 |
| 80059 | 290-300 | 95 | 0.48 | 0.048 | 0.62 |
| 80060 | 300-310 | 49 | 0.62 | 0.150 | 1.07 |
| 80061 | 310-320 | 85 | 0.45 | 0.140 | 0.87 |
| 80062 | 320-330 | 102 | 0.34 | 0.084 | 0.59 |
| 80063 | $330-340$ | 97 | 0.96 | 0.088 | 1.22 |
| 80064 | 340-350 | 96 | 0.52 | 0.077 | 0.75 |
| 80065 | 350-360 | 100 | 0.28 | 0.016 | 0.33 |
| 80066 | 360-370 | 101 | 0.31 | 0.026 | 0.39 |
| 80067 | 370-380 | 101 | 0.47 | 0.013 | 0.51 |
| 80068 | 380-390 | 98 | $0.30{ }^{\text {- }}$ | 0.007 | - 0.3 .32 |
| 80069 | 390-400 | 100 | 0.36 | $0.012 \cdot$ | 0.40 |
| 80070 | 400-410 | 103 | 0.18 | 0.015 | 0.22 |
| 80071 | 410-420 | 106 | 0.42 | 0.037 | 0.53 |
| 80072 | 420-430 | 101 | 0.32 | 0.014 | 0.36 |
| 80073 | 430-440 | 102 | 0.52 | 0.013 | -0.56 |
| 80074 | 440-450 | 96 | 0.37 | 0.035 | 0.48 |
| 80075 | 450-460 | 101 | 0.48 | 0.023 | 0.55 |
| 80076 | 460-470 | 100 | 0.36 | 0.020 | 0.42 |
| 80077 | 470-480 | 100 | 0.35 | 0.016 | 0.40 |
| 80078 | 480-490 | 103 | 0.36 | 0.015 | 0.40 |
| 80079 | 490-500 | 106 | 0.41 | 0.033 | 0.51 |
| 20080 | 500-510 | 98 | 0.27 | 0.024 | 0.34 |
| 80081 | 510-520 | 106 | 0.21 | 0.019 | 0.27 |
| 80082 | 520-530 | 101 | 0.19 | 0.008 | 0.21 |
| 80083 | 530-540 | 99 | 0.24 | 0.038 | 0.35 |
| 80024 | 540-550 | 100 | 0.23 | 0.027 | 0.31 |
| 80085 | 550-560 | 101 | 0.18 | 0.003 , | 0.19 |

DIAMOND DRILL HOLE P-9
(continued)

| Core <br> Sample <br> Number |  | Core Recovery (percent) | Copper (percent) | Molybdenite <br> (percent) | $\begin{gathered} \text { \% Copper } \\ \text { Equivalent } \\ \left(\% \mathrm{Cu}+3 \% \mathrm{MOS}_{2}\right) \end{gathered}$ |
| :---: | :---: | :---: | :---: | :---: | :---: |
| 80085 | 560-570 | 94 | 0.13 | 0.007 | 0.15 |
| 80087 | 570-580 | 94 | 0.36 | 0.010 | 0.39 |
| 80088 | 580-590 | 113 | 0.18 | 0.007 | 0.20 |
| 80089 | 590-600 | 93 | 0.29 | 0.061 | 0.47 |
| 80090 | 600-610 | 99 | 0.14 | 0.019 | 0.20 |
| 80091 | 610-620 | 101 | 0.26 | 0.021 | 0.32 |
| 80092 | 620-630 | 100 | 0.34 | 0.048 | 0.48 |
| 80093 | 630-640 | 100 | . 0.24 | 0.073 | 0.46 |
| 80094 | 640-650 | 102 | 0.24 | 0.169 | 0.75 |
| 80095 | 650-660 | 102 | 0.15 | 0.050 | 0.30 |
| 80096 | 660-670 | 102 | 0.11 | 0.151 | 0.56 |
| 80097 | 670-680 | 101 | 0.26 | 0.121 | 0.62 |
| 80098 | 680-690 | 100 | 1.68 | 0.065 | 1.88 |
| 80099 | 690-700 | 101 | 1.08 | 0.023 | 1.15 |
| 80100 | 700-710 | 102 | 1.36 | 0.040 | 1.48 |
| 80101 | 710-720 | 98 | 0.64 | 0.019 | 0.70 |
| 80102 | 720-730 | 100 | 0.32 | 0.056 | 0.49 |
| 80103 | 730-740 | 102 | 0.50 | 0.046 | 0.64 |
| 80104 | 740-750 | 99 | 0.47 | 0.042 | 0.60 |
| 80105 | 750-760 | 100 | 0.47 | 0.015 | 0.52 |
| 8010 ¢́ | 760-770 | 104 | 0.16 | 0.011 | 0.19 |
| 80107 | 770-780 | 98 | 0.24 | 0.052 | 0.40 |
| 80108 | 780-790 | 101 | 0.16 | 0.010 | 0.19 |
| 80109 | $790-800$ | 104 | 0.12 | 0.010 | 0.15 |
| 80110 | $800-810$ | 99 | 0.24 | 0.018 | 0.29 |
| 80111 | 810-820 | 97 | 0.05 | $<0.001$ | 0.05 |
| 80112 | $820-830$ | 101 | 0.19 | 0.006 | 0.21 |
| 80113 | $830-840$ | 101 | 0.37 | 0.016 | 0.42 |
| 80114 | $840-850$ | 100 | 0.05 | 0.002 | 0.06 |
| 80115 | $850-860$ | 102 | 0.10 | 0.003 | 0.11 |
| 80116 | 860-870 | 101 | 0.08 | 0.003 | 0.09 |
| 80117 | $870-880$ | 104 | 0.30 | 0.008 | 0.32 |
| 80118 | $880-890$ | 100 | 0.12 | 0.002 | 0.13 |
| 80119 | 890-900 | 101 | 0.07 | 0.003 | 0.08 |
| 80120 | 900-910 | 100 | 0.17 | 0.004 | 0.18 |
| 80121 | 910-920 | 100 | 0.03 | 0.002 | 0.04 |
| 80122 | 920-932 | 101 | 0.04 | 0.002 | 0.05 |

DIRECTIONAL SURVEY DATA

| Hecla - (acid bottle \& transit) |  |  | Sperry-Sun - (multi-shot gyro) |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: |
| Location | Inclination | Bearing | Iocation | Inclination | Bearing |
| Collar | $90^{\circ}$ | NO ${ }^{\circ}{ }^{\prime \prime} \mathrm{E}$ |  |  |  |
| $300 \cdot$ | $89^{\circ}$ | . NA |  |  |  |
| $60{ }^{\prime}$ | $90^{\circ}$ | - NA | . |  |  |
| 900' | $90^{\circ}$ | NA |  |  |  |

## DIAMOND DRILL HOLE P-10

| Start: | July 27, 1971 | Coordinates: | 26995.74 N |
| :--- | :--- | :--- | ---: |
| Einish: | July 30, 1971 |  | 7761.23 Z |
| Core Size: | NQ | Elevation: | $2,998.0^{\prime}$ |
| Core Recovery: | $99 \%$ | Inclination: | $90^{\circ}$ |
| Sample: | Split Core | Bearing: | $N^{\circ} 0^{\prime} \mathrm{E}$ |
| Directional Survey: | See last page | Length: | $594^{\prime}$ |


| Core <br> Sample <br> Number | Hole <br> Interval <br> (feet) | CoreRecovery(percent) | Copper (percent) | Molybdenite (percent) | \% Copper Equivalent |  |
| :---: | :---: | :---: | :---: | :---: | :---: | :---: |
|  |  |  |  |  |  |  |
|  |  |  |  |  | $(\% \mathrm{Cu}+3 \times$ | ${ }_{8} \mathrm{MOS}_{2} \mathrm{~J}$ |
| 80123 | 53-60 | 99 | 0.21 | 0.050 | 0.36 |  |
| 80124 | 60-70 | 98 | 0.21 | 0.061 | 0.39 |  |
| 80125 | $70-80$ | 100 | 0.40 | 0.073 | 0.621 |  |
| 80126 | 80-90 | 100 | . 0.33 | 0.044 | 0.46 |  |
| 80127 | 90-100 | 100 | 0.41 | 0.050 | 0.561 |  |
| 80128 | 100-110 | 100 | 0.44 | 0.025 | 0.52 |  |
| 80129 | 110-120 | 99 | 0.65 | 0.052 | 0.81! |  |
| 80130 | 120-130 | 97 | 0.38 | 0.046 | 0.52; |  |
| 80131 | 130-140 | 101 | 0.31 | 0.025 | 0.38 |  |
| 80132 | 140-150 | 96 | 0.37 | 0.031 | 0.46 |  |
| 80133 | 150-160 | 95 | 0.52 | 0.028 | 0.601 |  |
| 80134 | 160-170 | 99 | 0.45 | 0.073 | 0.671 |  |
| 80135 | 170-180 | 85 | 0.22 | $0.023{ }^{\circ}$ | 0.29 |  |
| 80136 | 180-190 | 88 | 0.06 | $0.010^{\circ}$ | 0.09 |  |
| 80137 | 190-200 | 103 | 0.07 | 0.012 | 0.11 |  |
| 80138 | 200-210 | 98 | 0.05 | 0.007 | 0.07 |  |
| 80139 | 210-220 | 101 | $<0.01$ | < 0.001 | $<0.01$ |  |
| 80140 | 220-230 | 100 | 0.04 | 0.005 | 0.06 |  |
| 80141 | 230-240 | 101 | 0.10 | 0.038 | 0.21 |  |
| 80142 | 240-250 | 89 | 0.24 | 0.536 | 1.85 |  |
| 80143 | 250-260 | 93 | 0.08 | 0.030 | 0.17 |  |
| 80144 | 260-270 | 102 | 0.05 | 0.113 | 0.39 |  |
| 80145 | 270-280 | 100 | 0.03 | 0.330 | 1.02 |  |
| 80146 | 280-290 | 102 | 0.07 | 0.217 | 0.72 |  |
| 80147 | 290-300 | 101 | 0.09 | 0.005 | 0.10 |  |
| 80148 | 300-310 | 96 | 0.11 | 0.063 | 0.30 |  |
| 80149 | 310-320 | 96 | 0.05 | 0.098 | 0.34 |  |
| 80150 | 320-330 | 97 | 0.04 | 0.002 | 0.05 |  |
| 80151 | $330-340$ | 97 | 0.06 | 0.002 | 0.07 |  |
| 80152 | 340-350 | 101 | 0.07 | 0.010 | 0.10 |  |
| 80153 | 350-360 | 100 | 0.07 | 0.016 | 0.12 |  |
| 80154 | 360-370 | 99 | 0.04 | 0.006 | 0.06 |  |
| 80155 | 370-380 | 100 | 0.09 | 0.019 | 0.15 |  |
| 80156 | 380-390 | 100 | 0.06 | 0.125 | 0.44 |  |
| 80157 | 390-400 | 102 | 0.25 | 0.022 | 0.32 |  |
| 80158 | 400-410 | 98 | 0.20 | 0.059 | 0.38 |  |
| 80159 | $410-420$ | 100 | 0.10 | - 0.010 | 0.13 |  |
| 80160 | 420-430 | 99 | 0.09 | $0.025=$ | $\cdots 0.16$ | $\cdots$ |
| 80161 | 430-440 | 101 | 0.34 | 0.025 | 0.42 |  |
| 30162 | 440-450 | 100 | 0.15 | 0.003 | 0.16 |  |
| 80163 | 450-460 | 101 | 0.26 | 0.009 | 0.29 |  |
| 80164 | 460-470 | 103 | 0.31 | - 0.010 | 0.34 |  |
| 80165 | 470-480 | 101 | 0.34 | 0.005 | 0.36 |  |
| 80166 | 480-490 | 100 | 0.27 | 0.006 | 0.29 |  |
| 80167 | $490-500$ | 102 | 0.26 | 0.020 | 0.32 |  |
| 80168 | 500-510 | 101 | 0.48 | 0.038 | 0.59 |  |
| 80169 | 510-520 | 98 | 0.32 | 0.033 | 0.42 |  |
| 80170 | 520-530 | 101 | 0.12 | 0.008 | 0.14 |  |
| 80171 | 530-540 | 97 | 0.23 | 0.002 | 0.24 |  |
| 80172 | $540-550$ | 102 | 0.17 | 0.007 | 0.19 |  |
| 80173 | 550-560 | 100 | 0.15 | 0.009 | 0.18 |  |
| 80174 | 560-570 | 102 | 0.24 | 0.012 | 0.28 |  |
| 80175 | 570-580 | 103 | 0.19 | 0.006 | 0.21 |  |
| 80176 | 580-594 | 101 | 0.08 | 0.004 | 0.09 |  |

DIAMOND DRILL HOLE P-10
(continued)

DIRECTIONAL SURVEY DATA

| Hecla - (acia bottle \& transit) |  |  | Sperry-Sun - (multi-shot gyro) |  |  |
| :---: | :---: | :---: | :---: | :---: | :---: |
| Iocation | Inclination | Bearing | Location | Inclination | Eearing |
| Collar | $90^{\circ}$ | NOOO'E | Collar | $90^{\circ}$ | N $0^{\circ} 0^{\prime \prime} \mathrm{E}$ |
| 3001 | $90^{\circ}$ | NA | 1001 | $88^{\circ} 45^{\prime}$ | S12034'W |
| $590^{\prime}$ | $89^{\circ}$ | NA | 2001 | $88^{\circ} 45^{\prime}$ | $529^{\circ} 00^{\prime} \mathrm{W}$ |
|  |  |  | $300 \cdot$ | $88^{\circ} 15^{\prime}$ | S15002' ${ }^{\circ}$ |
|  |  |  | $400{ }^{\prime}$ | $87^{\circ} 40^{\prime}$ | S29057'w |
|  |  |  | $500^{\prime}$ | $87^{\circ} 35^{\prime}$ | S27034'\% |
|  |  |  | $550{ }^{\circ}$ | $87^{\circ} 30^{\prime}$ | S30 ${ }^{\circ} 05^{\prime \prime}$ \% |

