

REPORT ON THE PROPERTY OF

JONSMITH GOLD MINES LIMITED

PARKIN TOWNSHIP - DISTRICT OF SUDBURY.

W. P. CORKING JUNE 1947

# REPORT ON THE PROPERTY OF JONSMITH GOLD MINES LIMITED PARKIN TOWNSHIP - DISTRICT OF SUDBURY.

#### INTRODUCTION:

well in nickel, copper, platinum, gold and silver, the Directors of this Company recently undertook a short preliminary program of diamond drilling and surface examination on their property in the Sudbury District. The primary aim was to corroborate the grab-sampling and to provide data with which to estimate the extent and nature of further work. The following report deals with the results of this program and with recommendations regarding the next stages of development.

#### LOCATION AND PHYSICAL ASPECTS:

The Jonsmith property consists of 24 mining claims covering portions of Lots 3, 4, 5, 6 and 7, in Concessions 2 and 3, Parkin Township, District of Sudbury, Ontario. (See accompanying Claim Plan - Figure "D".)

Three more have recently been staked. It lies some eight miles north-northeast of Capreol, and 22 miles north-northeast of Subbury. Access is gained by 6.2 miles of

improved road from Capreol in the direction of Milnet followed by approximately 5 miles of tote road to the present camp. Small aircraft can land in 20 minutes from Sudbury on a small lake one half mile south of the property. The C.N.R. transcontinental line passes along the Vermilien river  $3\frac{1}{6}$  miles to the east of the boundary. Good timber and water are plentiful on the ground.

## RELATIONSHIP OF JONSMITH PROPERTY TO GENERAL SUDBURY GEDLOGY

The Sudbury basin possesses economic significance too well-known for description here. It consists of a boat-shaped mass of igneous rock lying above a steeply tilted series of sediments and volcanics, and overlain by a series of lightly disturbed, almost horizontal sediments. The igneous basin itself is about 1 miles thick, 36 miles long and 17 miles broad; it takes the firm of a gentle syncline so that only the rim is exposed on surface. The outer shell of the basin consists of norite - a basic differentiate, and the inner shell of micropegmatite - an acid differentiate. Because of its association with many massive ore deposits, the norite is often referred to as thenickel eruptive. The ores of nickel, copper, platinum, gold, etc. are found either in bays around the margin

of the norite, or in "offsets" -- dike-like tongues which protude outward from the basin. On surface only two of the offsets (Foy and Copper Cliff) are actually continuous with the norite of the main body. Although variable from place to place, the average composition of the rock in the offsets approaches that of the norite and it is noteworthy that all of the offsets embrace numerous bodies of breccia.

Representatives of the offset deposits are Worthington, Frood, Stobio, Vermilion, Copper Cliff, Victoria mines, and of the basin deposits, Creighton, Gertrude, Murray and Falconbridge.

Certain low temperature lead-zine deposits occur in the sediments of the interior of the basin.

The south boundary of the Jonsmith property lies 2½ miles from the northeast corner of the nickel eruptive, at which point the Whistle ore deposit occurs. In this vicinity also, Ontario Nickel Mines Ltd., recently reported development of 800,000 tons of ore averaging 1.30% nickel, and 0.50% copper covering a strike length of 800 feet to a depth of 600 feet.

Traversing Parkin Township in a northwest-southeast direction, there are several dike-like bodies of

basic intrusive rock which are generally regarded as basic offsets from the Sudbury basin and these have been mapped as such.

Bodies of heavy sulphide mineralization similar in composition and habit to the Sudbury ores and containing combined values of ore grade in nickel, copper, platinum and gold occur within basic dike-rocks on the Jonemith claims.

#### LOCAL GEOLOGY

extends for approximately 2 3/4 miles along the northwestsoutheast strike of a series of sedimentary rocks. Underlying the ediments and forming the bedrock of the southwest
third of the property is a band of greenstone (very old
volcanics and intrusives). The contact of a large granite
stock with the greenstone parallels the long axis of the
claim block about 1 mile to the southeast. Several tengues
of basic rock of the offset type include the sodiments
parallel to their strike. About 65% of the area is covered
by glacial overburden, and It is interesting to note that
large sand and gravel plains occur in the granite country to
the cast. At Milnet, some 3% miles east, an attempt is
currently being made to develop placer gold reserves for
dredge operation.

The gold here is reported to occur in gravel beds near the Vermilion River.

In this present program, two areas on the ground have been explored to a very limited extent by diamond drill. These are referred to hereinafter as Zone No. 1 and Zone No. 2.

#### ZONE NO. 1

Exposures consist of two old rock trenches in the east central part of patented claim S 5265 (see Figures B and D). Heavy pyrite, pyrrhotite, pentlandite, chalcopyrite mineralization occurs in tongues of comparatively fresh diorite. The structure is very complex and it is apparent that the diorite is not a continuous dike, but occurs rather as a series of fingers, which strike in a generally northwesterly direction. Where cut by diamond drill holes it has a near-vertical dip. It is intruded between the contact of a somewhat contorted band of limestones on the southwest and a series of cherty quartzites on the northeast. A zone up to 30 feet thick of highly silicified breccia has been developed in the quartzite along the contact between it and the diorite, and locally, there are inclusions of this breccia within the diorite. The breccia is well mineralized with very finely disseminated, barren pyrite.

The ore mineralization occurs in solid masses and lacey networks in the massive, unsheared diorite - probably related in distribution to certain flat jointing and to bends and irregularities in the form of the intrusives.

In Trench No. 1, the lenticular body of mineralization has a length of about 50 feet and a maximum width of about 20 feet. Three drill holes, some 30 feet (vertical) below this trench, gave the following results: (See Figure "E").

#### Hole No. 1.

21.4 feet of core length assaying

Nickel - 1.76% Copper - 1.68% Platinum - 0.13 oz

Platinum - 0.13 ozs./ton Gold - 0.03 ozs./ton.

(\$28.85 combined value per ton at present market price.)

#### Hole No. 2

24.2 feet core length assaying

Nickel - 1.42% Copper - 1.44% Platinum - 1.12 oz:

Platinum - 0.12 ozs./ton
Gold - 0.56 ozs./ton

(\$28.85 combined value per ton at present market price.)

#### Hole No. 3

This hole passed about 15 feet off
the southeast end of the surface mineralization and the
diorite in it was found to contain very little mineralization.

In trench No. 2 the surface mineralization forms a tapering wedge some 80 feet long and 15 feet wide at the base of the wedge. Three drill holes about 40 feet below this trench gave the following results:

#### Hole No. 4

13.9 feet of core length assaying

Nickel - ) Assays not Copper - ) available at Platinum - ) the time of Gold - ) writing.

The mineralization in this hole is mainly concentrated within 3.2 feet of core and low values were obtained by the necessity of including, in the composite sample, scattered mineralization, which assayed well in gold alone, in two places. (\$8.40 and \$16.80)

#### Hole No. 5

This hole, in the northwestern end of the trench, cut no diorite and no mineralization of the ore type. This suggests a southeasterly rake.

#### Hole No. 6

A  $7\frac{1}{8}$  foot length of mineralized diorite was sampled together with fairly good mineralization in the

silicified breccia, separated by 10 feet of barren breccia.

This whole section of 32.2 feet core length assayed

Nickel - 0.78% Copper - 0.75%

Platinum - 0.06 ozs./ton
Gold - 0.01 ozs./ton

(\$12.84 combined value at present market prices.)

#### NOTE

In all of these holes, the core in the mineralized zones was split in short samples in the usual way, and assayed first for gold. Later, in order to obtain a more accurate assay of the four metals, composite samples were made up from aliquot parts of each sample within a given zone.

No attempt to channel sample the surface of this zone was made since surface oxydation would preclude the possibility of a representative sample. However, some grab samples were taken of the better looking material which gave values in Nickel, up to 1.40%; Copper 4.88%; Platinum .05 ozs.; and Gold 0.45 ozs.

A small amount of old test-pitting (Trench 3) was done in claim S 42676, starting about 100 feet southeast along the strike from No. 2 Trench. Here the geology is entirely different. A number of barren quartz ladder-veins out transversely across a highly carbonatized rock which lies

between two horizons of limestone. Two diamond drill holes were put down under this trench and though one of them cut nearly 30 feet of quartz-carbonate, (presumably a vein paralleling the hole) nothing of economic interest was found.

It should be noted that in the case of the mineralized diorite there is an almost complete lack of quartz and carbonate.

#### ZONE NO. 2

old rock work and stripping in claim S 39838. An undulating contact strikes west-northwesterly between limestone on the northeast and highly silicified breccia on the southwest. Within the limestone and near this contact, a very irregular strike-vein of quartz is exposed for about 50 feet, with a maximum width of 5 to 6 feet. Off the southeast end two trenches show that it either terminates or pinches to insignificant width. On surface the vein is open to the northwest. A large number of straight, parallel stringers branch off the main vein in a southwesterly direction giving a "half-feather" pattern to the quartz. (See Fig. C) One of these is a little wider than the met and apparently the trenching follows this, rather than the main leader vein.

Lenses of heavy galena-sphalerite mineralization

with minor amounts of pyrite and chalcopyrite are found locally in the quartz, and the breccia is well mineralized with finely disseminated pyrite with occasionally a little coarse pyrite and arsenopyrite. The quantity of galena and sphalerite present in the lenses is intriguing but not to the degree that mining of lead and zinc might be considered. However, grab sampling of the better mineralization here, gave values up to \$8.05 in gold, with some silver and it was hoped that this might prove worth while.

Four drill holes under this trench met with discouraging results. Although the quantity of quartz vein-matter intersected corresponded with that on surface, the galenasphalerite mineralization was found in only two short sections. In hole No. 10, 2.7 feet of quartz with galena, sphalerite, chalcopyrite and pyrite assayed \$3.50 in gold, and 2.30 ozs. in silver. Hole No. 11 cut 10 inches of quartz with other stringers containing a little arsenopyrite, chalcopyrite, galena and pyrrhotite which assayed \$5.25 in gold over 2.0 feet. Casual channel sampling in this vein zone gave the following assays in gold.

Value	Width
Traco	8.0 feet
\$0.70	8.0 feet
Trace	6.0 feet
\$9.80	6.0 feet
Trace	5.0 feet
\$1.40	6.0 feet

A fifth drill hole was aimed in the opposite direction across a narrow swamp. This cut limestone strata and

finally passed through an unmineralized contact into a massive grey rock, probably quartzite, but possibly a variant of the diorite in which the copper-nickel mineralization is found in Zone No. 1.

#### CONCLUSION

The feature of greatest interest on this property lies in Zone No. 1, where two lenses of copper - nickel - platinum - gold mineralization are found. The surface exposure of these two lenses has a combined area which indicates roughly 100 tons per vertical foot.

Of the six drill holes intended to intersect this zone (as nearly as possible at right angles to the strike and dip), four cut through mineralization of ore grade and width as follows:-

Number Length Nickel Copper	Platinum	Gold
2 24.2 ft. 1.42% 1.44% 4 13.9 ft. Assays not yet av	0.13 ozs. 0.12 ozs. vailable. Cor 0.06 ozs.	0.06 ozs. nb. values \$36.00

Of the other two holes, one passed through unmineralized host-rock and the other cut none of the host-rock.

The mineralogy and general characteristics of these bodies, although they are small, bears striking resemblance to the ores of the Sudbury nickel eruptive and are

associated with basic dike-like tongues of rock of the wellknown offset type. Similar basic dikes strike morthwasterly through the property and very little exploratory work has been done along this strike.

Under the foregoing circumstances, it is highly desirable that further work be carried out along the strike of this No. 1 Zone. The results already obtained from a very short program indicate reasonably good prospects of locating other bodies of nickel-copper - platinum - gold ore.

#### RECOMMENDATIONS

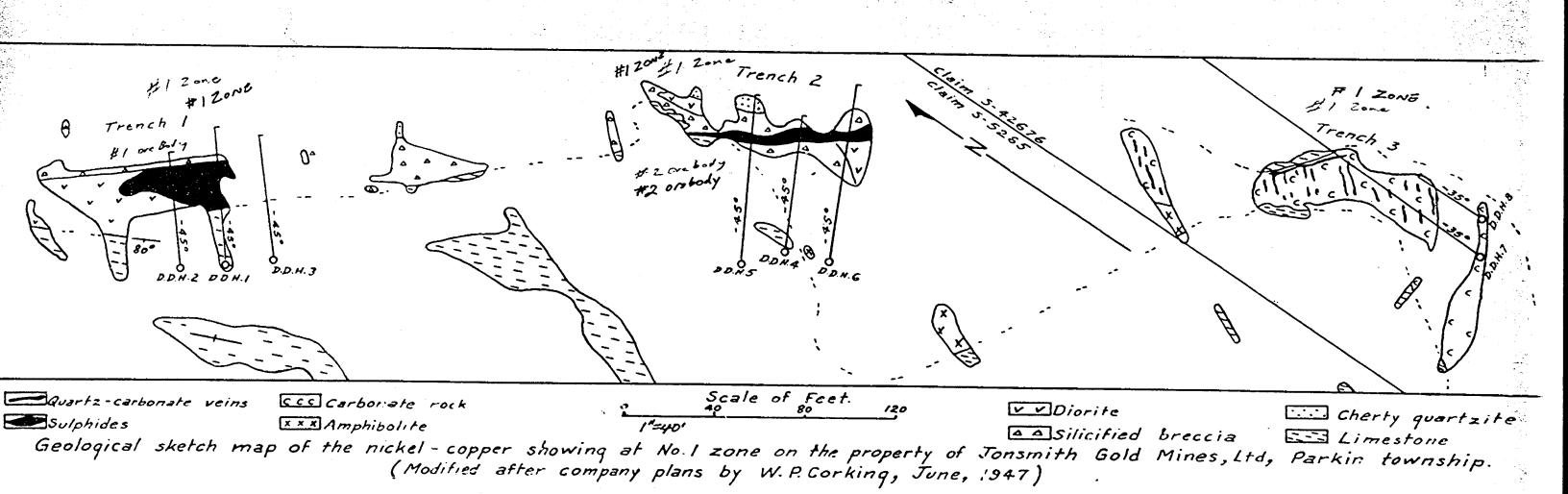
- Since magnetic pyrrhotite forms such a prominent part of the mineralization, there can be little doubt that a magnetometer survey would indicate anomalous magnetic intensities in the vicinity of such bodies. It is recommended that a base line be cut northwesterly through the property and that a (Vertical type) magnetometer survey be carried out, taking readings at 100 feet intervals along lines 200 feet apart, at right angles to the base line.
- 2. On completion of this survey, diamond drilling should be carried out in accordance with the anomalous magnetic pattern. It is of course, not possible to estimate the footage that may be required until the survey is completed.

It should be stated that the present tote road from the Capreol-Milnet road to the camp is almost useless, at

present, for freighting purposes. Although much of the road is suitable for trucks, there is perhaps \( \frac{1}{4} \) to \( \frac{1}{2} \) mile of swamp, which makes it almost impassable, even for horses. The geophysical survey will not require much freighting and it is suggested that this party be supplied by air, and that the road and camp situation be dealt with, when the diamond drilling situation develops.

E. G. BISHOP and W. P. CORKING

June 27th, 1947 Haileybury, Ontario.





020

NICKEL-COPPER PROPERTY

of

#### JONSMITH GOLD MINES, LTD.

#### PARKIN TOWNSHIP

#### HISTORY AND DEVELOPMENT

In 1949 Jonsmith Gold Mines, Ltd. held a block of 30 mining claims in lots 3-7, Concessions 2 and 3, Parkin township. The property lies about 8 miles north-northeast of Capreol on the Canadian National railway and is reached by approximately 5 miles of tote road from Milnet station.

Interesting values in nickel, copper, and precious metals were obtained from surface grab samples of sulphide mineralization. This led the company to undertake a program of surface work and diamond drilling in 1947. Thirteen holes, aggregating 1670 feet, were drilled at that time. A magnetometer survey of the property was also made.

#### ACKNOWLEDGMENTS

The following summary has been prepared from the maps and report on the property prepared by W.P. Corking, consulting geologist, in June 1947. These were kindly furnished by the Company. The writer has not visited the property.

#### GEOLOGY

The general geology of Parkin township is shown on Map No.41c of the Ontario Department of Mines. The south

boundary of the Jonsmith property lies 2 - 5 miles north of the Whistle mine, located on the northeast corner of the Sudbury nickel irruptive. The long axis of the Jonsmith block of claims extends for approximately 2% miles along the northwesterly-southeasterly trending series of sedimentary rocks. Several tongues of basic rock of the Sudbury nickel irruptive "offset" type intrude the sediments parallel to their strike. Two areas on the property, referred to as the No.1 and No.2 zones, have been explored by diamond drilling.

#### NO.I ZONE

This is represented on the surface by two old rock trenches in the east central part of claim S-5265 and No.3 trench on claim S-42676 (see Fig. ). At trenches No.1 and No.2 heavy pyrite, pyrrhotite, pentlandite, and chalcopyrite mineralization occurs in tongues of comparatively fresh diorite. The diorite occurs as a series of fingers intruded between a contorted band of limestone on the southwest and cherty quartzite on the north east; A zone of highly silicified breccia, up to 30 feet in width, has been developed in the contact zone of the quartzite; locally there are inclusions of this breccia within The ore mineralization occurs in solid masses and the diorite. lacy networks in the massive diorite. In trench No.1 the lenticular body of mineralization has a length of 50 feet and a maximum width of 20 feet. In trench No.2 the surface mineralization is wedge-shaped, being 80 feet in length and 15 feet wide at the base of the wedge. At trench No.3 the geology is quite different. Here a number of quartz ladder veins cut across a highly

carbonatized rock which lies between two horizons of limestone.

<u>Drilling Results.</u> A summary of drilling results at trench No. 1, as given in the report by W.P. Corking, is listed below. The location of the holes is shown on Fig. . The intersections were about 30 feet below the trench.

Hole No.	Core Length (feet)	Nickel (per cent)	(per cent)	Platinum (oz/ton)	Gold (oz/ton)
1	21.4	1.76	1.68	0.13	0.03
2	24.2	1.42	1.44	0.12	0.06
3	1,	litt	le minerali	zation.	

Holes Nos. 4, 5, and 6 were drilled under trench No.2. Hole No.4 showed 13.9 feet of mineralized core but most of the mineralization was concentrated within 3.2 feet. At two places in this sample gold assays of \$8.40 per ton over 2.5 feet and \$16.80 per ton over 1.7 feet (gold valued at \$35.00 per ounce) were obtained. Hole No.5 cut no mineralization of the ore type. In hole No.6, 32.2 feet of core assayed 0.78 per cent nickel, 0.75 per cent copper, 0.06 ounces platinum per ton and 0.01 ounces gold per ton. Holes 7 and 8 did not show any values.

#### NO. 2 ZONE

The surface showing at No.2 zone consists of a small amount of old rock work and stripping on claim S-39838. This lies about 4200 feet north west of No.1 zone. It consists of a quartz vein that is exposed for 50 feet along the strike and has a maximum width of 6 feet. Stringers of quartz branch from the main vein. Lenses of heavy galena-sphalerite mineralization

and minor amounts of pyrite and chalcopyrite are found locally in the quartz. Grab samples of the mineralized material show gold values on assay. Six holes were drilled into this vein but revealed only non-commercial values in gold and silver.

The mine underground is serviced by a three compartment shaft, to a depth of 475 feet, with levels at the 190', 300', and 465' horizons. The shaft compartments are all 5'0" x 6'0". Compartment # 1, the skip compartment, is serviced by a 2 ton automatic dumping slip. Compartment # 2, the cage compartment, services men and material. Compartment # 3 is the manway compartment and power services way, with 6" compressed air, 2" water, two 3" pump discharged, two 2" level drain lines and the necessary power cables.

Drifts and crosscuts on the three levels totalled 1,955.0 lineal feet, 610.0 feet on the 190' level, 630' feet on the 300' level, and 715' on the 465' level at the time Jonsmith Mines Limited took over the underground operation. The aforementioned work was by Milnet Mines Limited, just that necessary to extract the indicated #1 and #2 ore bodies.

A loading pocket station at the 425' horizon plus ore pass raises to the 190 level, services the broken ore.

Emplty stopes provide partial waste disposal, the remaining waste is decked to surface.

The mine water is pumped from the 465 pump station and averages 100 gallons per minute. Most of this water is made in the top 100.0 feet of the mine openings, with little or no water made on the 465 horizon.

#### GEOLOGICAL DATA AND ECONOMIC POSSIBILITIES

Jonsmith Mines Limited is located in Parkin Township, approximately four miles north of the northeast corner of the Sudbury Basin.

A quartz diorite dyke, the Norman-Parkin offset inrudes the property for more than a mile and it is along this dyke that ore deposits have been found.

The dyke is similar geologically and mineralogically to several others which radiate outwar from the rim of the Basin and which are the host rocks for several producing mines. Typical of these are the Nickel offsets Mine on the north side of the Basin; the Copper Cliff No.1 and No.2 and the Evans Mines on the south side and the Worthington Mine along the Worthington offset. The Frood Mines occurs along a quartz diorite dyke which is parallel to the rim of the Basin. The Whistle Mine, two miles south of Jonsmith Mines is a deposit along the Norman-Parkin offset and was formerly a producer of Nickel-copper ores. It is now owned by the International Nickel Company who plan to resume operations in the near future.

POOR QUALITY ORIGINAL TO FOLLOW

The Norman-Parkin offset on the Jonsmith Mines property is a quartz diorite dyke from forty feet to five hundred feet in width and strikes in a northeast direction for 4,000 feet where it bends to the northwest. It was along this northwest limb that the two ore bodies occur. The present surface surveys only extended a few hundred feet northwest of the mine workings and there is a suggestion that the dyke resumes its original course and continues for three claims to the north boundary of the property.

As in the other mines in the Sudbury area the ore occurs in breccia zones the result of either faulting or foldings. On the Jonsmith property the breccia zones were caused by northwest trending faults which shattered the quartz diorite host rock sufficiently to provide favourable receptacles for nickel bearing sulphides.

This breccia is also found in the Burton showing near the south boundary of the property where diamond drilling has revealed a narrow but high grade nickel-copper zone and where deeper drilling is now in progress.

The area north of the shaft warrants complete investigation to determine whether the northwest trending faults intersect the intrusive along the assumed northerly extension and produce favourable breccia zones.

The decepst level, the 465 foot horizon has a geological pattern similar to that on surface and it is highly probable that this will continue to depth. Thus the possibilities of locating new ore bodies below the 465 feet level are very favourable.

From the geological aspect the whole length of the dyke is prolific prospecting ground for nickel-copper deposits. At present a length of 5,200 feet has been mapped and there are indications that it continues for another 5,000 feet to the north boundary of the property.

Along the mapped portion of the dyke several anomalous zones were indicated by a magnetic survey and these corresponded to the areas outlined by a geochme.cal survey. These anomalies are probably signficant indications of sulphide zones and warrant detailed investigation.

#### Milnet Mines Ltd. - Development and Production

Prior to 1952 Jonsmith Mines Limited completed a partial geological surface survey of their property and had outlined by closely spaced diamond drill holes the #1 and #2 ore bodies. Subsequently Milnet Mines Ltd. leased a block of ground 1000 feet x 500 feet to 500 feet in depth, embracing #1 and #2 ore bodies, for the purpose of mining the nickel-copper ore contained in this block.

POOR QUALITY ORIGIN.
TO FOLLOW

During 1952 Milnet Mines constructed the aofresaid surface and underground plant and purchased the necessary equipment. Shaft sinking and the level stations were completed by Oct. 31, 1952. Development of the 190' and 300' levels was started with the first ore delivereries to Falconbridge Nicekl Mines Ltd. in the latter part of Dec., 1952. Development of and production from the #1 and #2 ore bodies from 465 level to surface continued to completion as of July 18, 1954. In Feb. 1954 Milnet Mines shipped to Falconbridge Nickel Mines 13,035 dry tons of ore with plant at approximately 80% of its hoisting capacity.

The #1 ore body as mined by Milnet Mines extended from surface to 350 feet below surface. The surface outcrop was 60 feet in length. The longest stoping length was 240 feet; the average stoping length was 120 feet. The surface outcrop of #2 ore body was 50 feet in legnth; the average stoping length was 55 feet. This vein structure persisted to below the 465' horizon. Stoping was completed from the 400' horizon to surface, below this the vein structure was below shipping grade, that is nickel 1.00%, copper 1.00%.

Milnet Mines Limited from Dec. 15, 1952 to July 18, 1954, mined and shipped to Falconbridge Nickel Mines a total of 157,755.70 dry tons of nickel, copper and precious metals ore. This ore assayed - nickel 1.49%, copper 1.54% (to date); gold 0.027 oz., platinum 0.066 oz., palladium 0.087 oz. and combined iridium, rhodium, and ruthenium 0.0032 oz. (to the end of Apr. 1954, the latest availabel assays)

Total contained metals in the above 157,755.70 dry tons were as follows:

Nickel	(to date)	4,711,119.0 lbs.
Copper	(to date)	4,846,847.0 lbs.
Gold	(To Apr. 30/54)	3,834.5 oz.
Platinum	n n	9,299.6 oz.
Palladium	**	12,234.4 oz.
Iridium,		
Rhodium &		
Ruthenium	(")	457.7 oz.

Total recoverable metals in the above 157,755.70 dry tons were as follows.

Nickel	(t0 date)	3,837,088.0 lbs.
Copper	(to date)	3,762,967.0 lbs.
Gold	(to Apr. 30/54)	932.7 oz.
Platinum	- w	5,860.9 oz.
Palladium	n .	7,695.3 oz.
Iridium,		
Rhodium &		
Ruthenium	( ")	285.9 oz.

POOR QUALITY ORIGIN. TO FOLLOW

#### Jonsmith - Current Exploration and Development

On July 19, 1954, Jonsmith Mines Limited purchased the surface and underground plant and equipment from Milnet Mines Limited.

On the same date Jonsmith resumed lateral work northwest and southeast on the 300' level, and southwest on the 465' level along strike of the favourable diorite. Underground diamond drilling at 50 foot intervals to probe the full width of the favourable diorite was started.

Jonsmith recently completed surface geological magnetic, electro-magnetic, and geochemical surveys on part of the Jonsmith Mines holdings; on receipt of this, sufrace diamond drilling was started on the "Burton shwoings". Other favourable areas are to be dimaond drilled on completion of the drilling in the "Burton Shwoing" area.

A report by C. E. Bowker, Mine Manager, dated August 10th, 1954 estimated that to thoroughly prespect the favourable diorite along its known strike length and to 1,000 feet plus in depth would require an additional \$850,850.00. As this amount of money is not available to Jonsmith Mines Limited, monetary assistance was sought from the Materials Division of Emergency Procurmenet Services, Washington, D.C., U.S.A.

After discussions with officials of the above Division on Sept. 3, 1954, they suggested a more comprehensive report and to break the above exploration and development program into projects. The rating or priority of the projects to the bidrectly related to the speed of possible nickel ore deliveries.

#### Projects - Priority and Costs.

#### Project #1

- (a) Deepen the shaft to 1,000' horizon, cutting three levels at 175' intervals.
- (b) Drift out the favourable diorite on these three new levels, completing as a minium 2,000 feet of drifting and cross-cutting on each of the new levels.
- (c) Diamond drill the 640', 815', and 990' horizons at close intervals.

The estimated cost of Project #1 is as follows :

(a) Deepen the shaft to 1,000' horizon with stations at 175' intervals; shaft, station,s and pump - 635 feet 0 \$210.00 per ft. \$133,350.00

DUPLICATE COPY POOR QUALITY ORIGINAL TO FOLLOW The mine underground is serviced by a three compartment phart, to a depth of 475 feet, with levels at the 1901, 3001, end 4651 herizons. The shaft compartments are all 51-07 x 61-07. Compartment \$1, the skip compartment, is serviced by a 2 ton automatic dumping skip. Compartment \$2, the cage compartment, services men and material. Compartment \$3, is the manual compartment and power service acy, with 64 compressed air, 25 water, two 35 pump discharges, two 27 level drain lines and the necessary power cables.

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#### September 7th, 1954.

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clong the mapedd portion of the dyke several anomalous zoned were indicated by a nagnetic survey and these corresponded to the areas outlined by a geochemical survey. These anomalies are probably significant indications of sulphide nones and warrant leteiled investigation.

#### "ilnet "ines "inited - Development and Production

Prior to 1952 Jonanith Mines limited completed a partial geological surface nurvey of their property and had outlined by closely speed diamond drill holes the \$1 and \$2 ore bodies. About ently "linet Mines itd. leased a blook of ground 1,000 feet x 500 feet to 500 feet in depth, embracing \$1 and \$2 ore bodies, for the purpose of mining the nickel-copper ore contained in this block.

#### September 7th, 1954.

During 1952 Filmet Wines constructed the aforesaid surface and underground plant and prochased the necessary equipment. That sinking and the level stations were completed by let. 31, 1952. Levelopment of the 190' and 300' levels was started with the first are deliveries to Falconbridge Rickel Vines 1td. inthe latter part of Dec. 1952. Pevelopment of and production from the 51 and 52 are bodies from 465 level to surface continued to completion as of July 18, 1954. In Feb. 1954 Milnet Mines shipped to Falconbridge Rickel Vines 13,035 dry tons of ore with the plant at approximately 80% of its hoisting capacity.

The \$1 ore body on mines by Milnet Mines extended from surface to 350 feet below surface. The surface outcrop was 60 feet in length. The longest stoping length was 240 feet; the average stoping length was 120 feet. The surface outcrop of \$2 ore body was 50 feet in length; the average stoping length was 55 feet. This wein structure persisted to below the 465 horizon. Stoping was completed from the 400 horizon to surface, below this the vein structure was below shipping grade, that is nickel 1.00%, copper 1.00%.

illnet lines simited from Dec. 15, 1952, to July 18, 1954, mined and shipped to Palconbridge bickel Mines a total of 157,755.70 dry tons of mickel, copper, and precious metals ore. This ore usuayes - mickel 1.49%, copper 1.54% (to date); gold 0.027 oz., platinum 0.066 oz., palladium 0.037 oz., and combined iridium, rhodium, and ruthenium 0.0032 oz. (to the end of apr. 1954, the latest available assayo).

Total contained metals in the above 157,755.70 dry tons were as follows:

Hickel Copper		date)		4,711,119.0	lbs.
Cold			1041	4,846,847.0	
Tlatinus	3 6.	o Apr. 30.	1291	3,834.5	
	}	н	}	9,299.6	
Valledium Tridium,	(	•••	,	12,234.4	02.
Bhodium, &					
"uthonium	(	n	)	457.7	02.

Notal recoverable metals in the above 157,755.79 dry tons note as follows:

Nickel Copper Cold Flatinum	(to	date) date) Apr. 30/	54)	3,837,086.0 3,762,967.0 932.7 5,860.9	1bs.
Palladium Iridium, Ehodium, & Authonium	(	u n	)	7,695.3 285.9	oz.

#### Jonarith - Current exploration and Development

on July 19, 1954, Wonsmith Fines Limited purchased the surface and enderground plant and equipment from Filmet Mines limited.

on the same date longmith resumed lateral work northwest and southwest on the 300' level, and southwest on the 465' level, along strike of the favourable diorite. Underground diamond drilling at 50 foot intervals to probe the full width of the favourable diorite was started.

Jonemith recently completed surface goological, magnetic, electro-magnetic, and geochemical surveys on part of the Jonemith Mines holdings; on receipt of this, surface diamond drilling was started on the "Burton Chowing". Other favourable areas are to be diamond drilled on completion of the drilling in the "Eurton Chowing" area.

A report by C. E. Bowker, Mine Manager, dated August 10th, 1954, estimated that to the roughly prospect the favourable disrite along its known strike length and to 1,000 feet plus in depth would require an additional \$250,850.00. In this amount of money in not available to Jossmith Mines Limited, about are usual sought from the Materials Mivision of emergency requirement Services, Mashington, J. C., U. J. A.

Latter discussions with officials of the above fivision on dept. 3, 1954, they suggested a more comprehensive report and to becar the above exploration and development progress into projects. The rating or priority of the projects to be directly related to the speed of possible nickel are deliveries.

#### Projects - Fierity and Costs

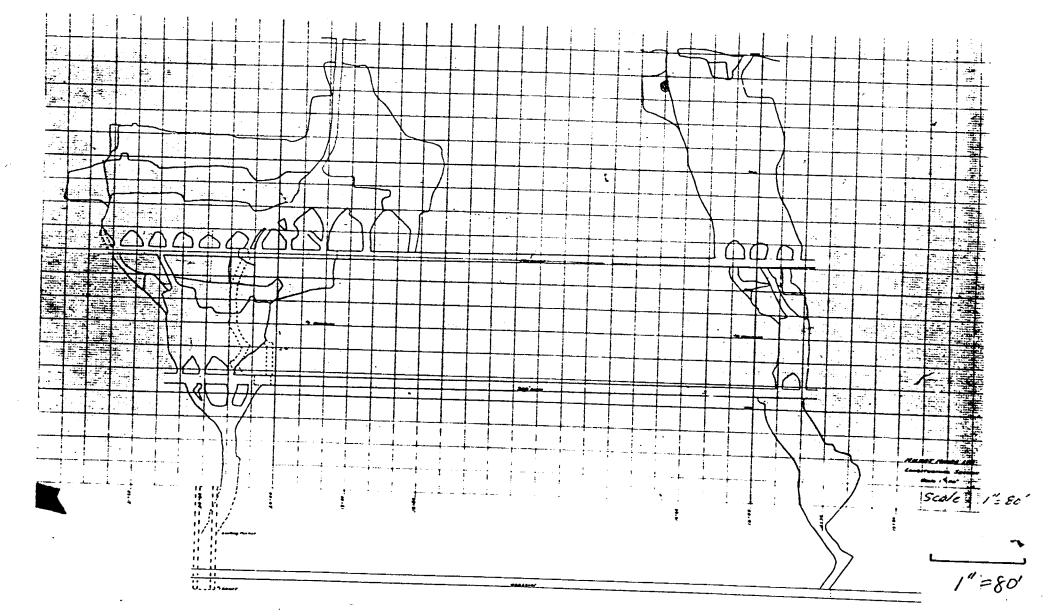
#### Project of

- (a) Region the sauft to 1,000' horizon, outling three levels at 175' intervals.
- (b) Drift out the favourable diorite on these three new levels, completing as a minimum 2,600 feet of drifting and cross-cutting on each of the new levels.
- (c) isomord drill the 640', 815', and 990' horizons at close intervals.

The estimated cost of Project Al is as follows:

(a) Teepen the shaft to 1,000' horizon with stations at 175' intervals; shaft, stations, and sump - 635 feet @ {210.00 per rt. 1133,350.00





Longitudinal section through Milnet mine, Jensmith Mines Lil, Farkin township. After plans of Monet Mines Lil

NOTE: THE FOLLOWING REPRESENTS H. WILSON'S CALCULATIONS FOR TONNAGES + GRADE OF NO. 1 AND NO. 2 OREGODIES ON JONSMITH PROPERTY.

JONSHITH SAMPLES

# Calculations for Average Grade of No. 1 Oreshoot.

		Taneth	Tons	cu.\$	Tx5	N1.5	TXX	Pt.oxe.	T X OEB.
Bection	Aros	Length		1.68	35,212.8	1.24	25,990.4	0.111	2,326.56
38-6-5	5,240	40	20,960			1.47	78,511.2	0.126	6,729.53
4-25 etc	9,370	57	53,409	2.09	111,624'.6		88,546.8	0.133	4,261.32
3-45-42	5,340	60	32,040	3.17	101,566.8	2.67			4,146.37
1-2-40	4,120	68	28,016	2.11	59,113.8	0.84	23,533.4	0.148	
		36	3,672	2.40	8,812.8	1.39	5,104.1	0.077	282.74
13	1,020					-		0.128	17,746.52
•		2614	156,097	2.29	316,331.0	1.58	218,685.9	0.120	
					MARISTO			* .	
		47 <sub>-</sub>			TOTALS.			i.	
				2.89	516,331.0	1.58	218,685.9	0.128	17,746.52
No. 1 0r	epody	Ş	138,007			1.27	43,780.7		
No. 2 Or	еро дд		34,600	1.32	45,686.5	1.21	*****		
			172,697	2.10	362,017.3	1.52	262,466.6		

#### Calculations for Average Grade of No. 2 Ore Shoot.

Section	Area	Tons	Cu.\$	Txs	N1.5	TXS
1	355	710	2.47	1,753.70	0.93	660.30
. <b>3</b>	<b>33</b> 0	660	2.85	1,551.00	0.59	389.40
8	1,045	2,090	1.80	3,762.00	1.65	3,448.50
4	2,125	4,250	1.17	4,972.50	0.80	3,400.00
5	5,795	11,590	0.93	10,778.70	0.91	10,546.90
6 / 6 /	3,165	6,330	1.33	8,418.90	2.18	13,799.40
7	2,925	5,850	1.51	8,833.50	1.30	7,605.00
8	1,560	3,120	1.80	5,616.00	1.26	3,931.20
		54,600	1.38	45,686.30	1.27	43,780.70
			TOTALS			
		***				
No. 1 0r	ebody	138,097	2.29	316,331.00	1.58	218,685.90
2 1	•	34,600	1.32	45,686.30	1.27	45,780.70
Total		172,697	2.10	362,017.30	1.52	262,466.60

# Calculations for Average Grade of Sections - No. 2 Orebody.

Bection	Hole No.	Length	<u>Cu.≰</u>	SXL	N1 . \$	5 x L	
	8-1	19.0	1.72	32.87	0.73	13.87	
	S-2	5.0	4.71	23.55	0.07	00.35	
		24.0	2.35	56.42	0.59	14.22	
8	8-3	30.0	1.47	44.10			
	8-4	35.0	2.08	72.80	1.02 8.19	30.60	
	ရ ပျိုက်ပြီးသည် ရေးသည်	65.0	1.80	116.90	1.65	76.65 107.85	
	5-5	44.0	1.06	46.64			
Salahar Salaha Salahar Salahar Salah	8-6	61.0	1.06	64.66	0.47	20.68	
r Pri Openia	3-,1B	15.0	0.69	10.35	0.76	69.54	
	5-22	30,0	1.53	45,90	0.59	11.40	Area
	17.	11.9	1.81	21.54	0.85	17.70	2125 A
		181.9	1.17	189.09	0.80	10.12 129.44	
8	S-7	46.0	0.46	01 16			
	8-8	65.0	1.20	21.16 78.00	0.38	17.48	
	8-17	79.0	1.00	70.00	1.55	100.75	
	47	47.3	1.33	62.91	1.04	82.16	
	8-23	72.0	0.64	46.08	1.07	50.61	*
		The Manager			0.43	30.96	
		309.3	0.93	287.15	0.91	281.96	
6	S-9	39.0	0.81	31.59	1.33	51.67	
	S-16	74.0	1.72	127.28	2.82	208.67	
	S-10	35.0	1.53	53.55	1.26	44.10	
	8-24	36.0	0.89	32.04	2.70	97.20	
		184.0	1.33	244.46	2.18	401.85	
7	8-12	15.0	1.33	19.95	0.35		
	S-14	43.0	0.86	36.98	0.38	£5.25	
	8-25	84.5	1.87	158.02	1.68	37.84 141.96	
		142.5	1.51	214.25	1.30	185.05	
В	8-26	42.5	0.67	28.48	0.66	28.05	
	48	45.0	1.81	81.45	1.58	71.10	
	3-27	49.0	2.76	135.24	1.50	73.50	
		136.5	1.80	245.17	1.26	172.65	

## Recalculating Creelman Averages - JONSMITH Sampling

No. 2 Ore Zone.

•					NO. 2 U	re zone	<u>)</u> .				•	
									Pa	lladium		
Sam	nia T+	Cu	4 - 7	- 174			tinum				-	
No		· Cu.	XIL	N1.	%xL	rt.	OZB.	OZS.X L	Jt.	028.	ors.x L	
	8-1					·	14.0	to 29.0'				
208	5.0	1.91		0.29		5.0	0.061				•	
809	5.0	1.83		1.00		5.0	0.051		5.0	0.063		
210	5.0	1.44		0.90		5.0		,	5.0	0.028		
	15.0	1.73		0.73		. ——	0.025		5.0	0.018		
Role	8-3	2170		0.75		15.0	0.046		15.0	0.036		
218	5.0	. 8.73.	$r_{\rm w}$	6 <b>1.13</b>		10.0		0 34.01				ř.,
213	5.0	1.57	1	1.63	range of the	10.0	0.108	1.08	20.0	0.026	-260	
814	5.0	2.48		1.95								
215	5.0	0.16		0.28	1	10.0	0.174	1.74	10.0	0.058	.580	
216	5.0	0.42	ss • €	0.09	•	5.0	0.033	.165		0.045	1 6 m	_
	25.0	1.47		1.02					8.0	0.045	2.255	D
Pole	8-4	****		1.02		25.0	.119	2.985	25.0	0.043	1.065	
							, 38.0°	to 73.0	•			
217	5.0	0.73		0.41		5.0	0.033	.165	5.0	0.045	.225	D
218 219	5.0	1.87		1.19	)	10.0	0.028	.280	10.0	0.040		
220	5.0	2.58		1.10	, ,)	2010	0.000	•200	10.0	0.049	<b>.49</b> 0	
221	5.0 5.0	2.22		5.20	,	10.0	0.068	.680	10.0	0.039	.390	
222	5.0	3.91		0.90	,		• • • • • • • • • • • • • • • • • • • •	••••	20.0	0.005	•390	
223	5.0	2.00		4.22		10.0	0.412	4.120	10.0	0.297	2.970	
	35.0	8.08		2.19	,	-	-		-	-		
***		a.00		2.19	,	35.0	0.150	5.245	35.0	0.116	4.075	
Hole						ě	6.0	to 27.	0*			
239	10.0	0.46	4.60	0.15	1.50	10.0	0.065	.650	10.0	0.035	•350	
240	5.0	0.36	1.80	0.45	2.25	5.0	0.012	•060	5.0	0.007	.035	
241	6.0	2.35	15.90	1.04	6.24	6.0	0.088	.528	6.0	0.050	.300	
	31.0	1.06	22.30	0.47	9.99	21.0	0.059	1.238	21.0	0.033	.685	
Bole	8-6						32.01			V.000	•005	
245	10.0	1 00	10.00	3 48	34.50							
346	10.0	1.00 0.88	10.00 8.80	1.47	14.70	10.0	0.028	.280	10.0	0.033	.330	
247	5.0	1.41	7.05	0.47	4.70 2.45)	10.0	0.031	.310	10.0	0.027	.270	
248	5.0	2.87	14.35	2.36	11.80)	10.0	0.039	.390	10.0	0.037	.370	
249	5.0	0.59	2.95	2.73	13.65)					0.001	.070	
250	5.0	0.76	3.80	0.81	4.05)	10.0	0.053	.530	10.0	0.049	.490	
251	5.0	1.46	7.30	0.87	4.50)							
252	5.0	0.48	2.40	0.70	3.50)	10.0	0.048	<b>.4</b> 80	10.0	0.032	.320	
253	5.0	0.60	5.00	0.28	1.40)	•••						
254	6.0	0.89	5.54	1.47	8.82 )	11.0	0.064	.704	11.0	0.055	.605 0	•
	61.0	1.06	64.99	1.14	69.42	61.0	0.044	9 404	43.0			
Hole					00.42	01.0		2.694	61.0	0.039	2.385	
							a). U	' to 47.	D*			
256	5.0	0.88	4.40	0.55	2.75							
257	5.0	0.49	2.45	0.57	2.85							
258 259	5.0 5.0	0.43	2.15	0.26	1.30							
260	7.5	0.28 0.51	1.40 2.32	0.30	1.80							
		0.31	2.32	0.26	1.95							
	27.5	0.46	10.72	0.38	10.35							

O Not weighted.

Sample No.	Too t	Cu.	5 x L	N1.	<b>%</b> x L	
Hole 6-8	******		direction.	£21	A. A	
865	5.0					83.0' to 98.0'
266	5.0	1,09	5.45	1.22	6.10	
267		1.21	6.05	1.03	5.15	
268	5.0	1.00	5.00	0 19	0.95	
269	5.0	0.52	2.60	0.15	û <b>.7</b> 5	
270	10.0	0.71	7.10	0.28	2.80	
270 271	10.0	0.70	7.00	0.64	5.40	
	5.0	1.12	5.60	2.36	11.80	
272	5.0	0.67	3.35	5.80	29.00	
273	5.0	2.39	11.95	4.54	22.70	
274	5.0	3.97	19.85	2.25	11.25	
875	5.0	0.83	4.15	0.81	1.05	
	65.0	1.20	<b>76.30</b>			
Hole 8-9	99.0	1.20	78.10	1.55	100.95	
			4			16.0' to 33.0'
277	5.0	0.78	3.60	2.86	14.30	
278	5.0	1.02	5.10	0.90	4.50	
279	7.0	0.72	5.04	<u>C.55</u>	3.05	
	17.0	0.81	13.74	1.33	22.65	1. S. C.
Hole 8-10					22100	44.0' to 79.0'
283	5.0	1.70		3.08		
284	5.0	1.50		1.78		·
285	5.0	1.48		2.57		
286	<b>5.0</b>	1.77		0.22		<b>√</b>
287	5.0	1.23		0.45		
288	5.0	1.23		0.52		
289	5.0	1.78		0.18		
	35.0	1.53		1.26		•
Hole 8-12						54.0' to 69.0'
290	5.0	0.91		0.52		
<b>291</b>	5.0	0.99		0.18		
292	5.0	2.09		0.36		
	15.0	1.33		0.35		
Hole 8-14						109.0' to 152.0'
294	10.0	1.23	12.30	0.33	3.30	20710 10 20210
295	10.0	0.94	9.40	0.20	2.00	
298	5.0	0.95	4.75	0.92	4.60	
297	5.0	0.95	4.75	1.52	7.60	
298	5.0	0.44	8.20	1.57	7.65	
299	5.0	0.38	1.90	0.79	8.95	
300	3.0	0.54	1.62	2.83	8.49	
	4.5.0	0.86	36.92	0.68	37.79	

Sample No.	Yest	Cu.	<b>XXL</b>	Ni.	5 x L			
Hole 8-16		principile	<del>diministratio</del>		<del></del>	51.01	<b>t</b> o	125.0
	# Ó	1.41	7.05	5.03	25.15			
301 302	5.0 5.0	2.31	13.05	10.13	50.65		•	
303	<b>5.</b> 0	1.42	7.10	1.72	8.60			
304	<b>5.</b> 0	1.50	7.50	0.91	4.55			
305	<b>5.</b> 0	0.54	2.70	1.24	6.20			, · · · · · · · · · · · · · · · · · · ·
306	5.0	0.60	3.00	3.04	15.20			
307	5.0	0.34	1.70	6.08	-3.01			
308	5.0	0.92	4.60	3.49	17.45	* * *	. : - '	
309	5.0	1.28	6.40	1.42	7.10			
310	5.0	0.86	4.30	0.88	4.40		* -	
311	5.0	1.07	5.35	1.79	. 8.95		2	
312	5.0	1.22	6.10	1.76	8.80			
313	5.0	2.00	10.00	3.21	16.05			
314	5.0	5.80	29.00	3.83	19.15	·		
315	4.0	4.81	19.24	3.33	13.38			
	74.0	1.72	127.09	2.82	208.58			* * * *
Hole 8-17				, , ,		33.01	to	יסיפנו
317	10.0	0.70	7.00	0.64	6.40			
318	10.0	0.90	9.00	0.68	6.80			
319	10.0	0.14	1.40	0.15	1.50			
320	10.0	0.48	4.80	1.12	11.20			•
321	10.0	0.59	8.90	1.45	14.50			
322	10.0	0.86	8.60	0.86	8.60			
<b>32</b> 3	10.0	1.46 3.66	14.60 18.30	0.37 4.17	3.70 20.85	•		
324 325	5.0		9.32	2.16	8.64			
320	4.0	2.33		2.10	0104			
77-1- A 10	79.0	1.00	78.92	1.04	82.19	83.01	•-	00.01
Hole 8-18				0.43		9310	to	\$0.0.
327	B.0	0.81		0.61				
328	5.0	0.64		1.42				
329	5.0	0.61		0.24				
7.1.000	15.0	0.69		0.76		00 01	4.	4% 01
Hole 6-20	<b>_</b> -					20.0'	10	43.0'
<b>33</b> 0	5.0	1.24		0.48				
331	5.0	3.90		1.34				
332	5.0	2.28		0.97				
, 	15.0	2.47		0.93		100.00		300.0
Hole 5-22						TES.O.	<b>to</b>	152.0'
333	10.0	1.51		0.59				
334	10.0	1.80		1.07				
335	10.0	1.28		0.12				
	30.0	1.53		0.59				
1.0								

Sample No.	<u> Foot</u>	Cu.	FXL	N1.	s by L	
Hole S-23		en e		•		73.0' to 145.0'
339	10.0	0.57	3.70	0.74	7.40	1010 10 21010
340	10.0	0.32	3.20	0.71		
341	10.0	0.17	1.70	0.04	7.10	
342	10.0	0.43	4.30		0.40	
343	7.0	0.12	0.84	0.30	3.00	
344	5.0	1.34	6.70	0.05	0.35	
345	5.0	1.71	8.55	0.42	8.10	
346	5.0	0.88		0.62	3.10	
847	5.0	1.16	4.40	0.21	1.05	
348	and the second s		5.80	0.81	4.05	
	5.0	1.42	7.10	0.48	2.40	
	78.0	0.64	46.29	0.43	30.95	
Hcle 8-24						122.0' to 153.0'
356	5.0	0.63	3.15	1.99	9.95	
357	5.0	1.33	6.65	3.98	19.80	
358	5.0	0.84	4,20	3.61	18.05	- 1
359	5.0	1.10	5.50	2.04	10.20	
360	5.0	0.54	B.70	5.89	19.45	
361	5.0	0.82	4.10	3.24	16.20	
362	6.0	0.96	5.76	0.61	3.66	
	36.0	0.89	32.06	2.70	97.31	
Hole S-25	*			2110	V/ 101	108.0' to 192.5'
363	10.0	0.43	4.30	0.39	m 00	TOOLO TO TATIO
364	10.0	0.89	8.90	0.33	3.90	
365	10.0	1.53	15.30	0.37	3.30	
366	5.0	1.96	9.80	0.22	3.70	
367	5.0	2.23	11.15	0.26	1.10	
368	5.0	2.80	14.00	1.10	1.30	
369	5.0	3.39	16.95	0.87	5.50	
370	5.0	2.54	12.70	0.94	4.35	
371	5.0	1.28	6.40	2.37	4.70	
372	5.0	1.64	8.20	1.85	11.85	
378	5.0	6.43	32.15	4.68	9.25	
374	<b>B.</b> 0	1.29	6.45		23.30	
375	5.0	1.96	9.80	3.61 3.53	18.05	
376	4.5		1.75		17.65	·
		0.39		7.58	34.11	
	84.5	1.87	157.85	1.68	142.06	
Hole S-28	_ :			•		190.0' to 232.5'
381	10.0	1.09	10.90	0.26	2.60	
382	10.0	0.60	6.00	1.12	11.20	
	10.0	0.60	6.00	0.81	8.10	
384	10.0	0.49	4.90	0.24	2.40	
385	2.5	0.40	1.00	1.43	3.58	
	42.5	0.67	28.80	0.66	27.88	

expension of the common of t

							***		
Sample	Tost	<u>a.</u>	% x L	N1.	SXL	Pt.02.	Lxoz.	Pd.02.	Lx 02.
No. Hole 8	_07	* * * * * * * * * * * * * * * * * * * *			•			306.01	to 355.0'
11010	-61				1			500.0	•0 000.0
389	4.0	1.01	4.04	2.96	11.84				· · · · · · · · · · · · · · · · · · ·
390	5.0	7.98	39.90	2.44	12.20				
391	5.0	2.90	14.50	1.88	9.40				
392	5.0	1.01	5.05	0.16	0.80				
373	5.0	3.99	19.95	1.75	8.75	) . · · · · · · ·			
394	5.0	1.89	9.45	0.50	1.50				in the second
395	5.0	0.51	2.55	0.14	0.70			*	ting a second
898	5.0	3.07	15.55	3.58	16.90				
397	5.0	2.28	11.40	1.48	7.10			• •	
398	5.0	2.62	13.10	0.84	4.20				
	49.0	2.76	135.29	1.50	73.39		. 13	13.6 to	125.5'
Hole /					0.30	004			
67	1.4	1.22	1.71	0.13	0.18	•004	.0056	.018	.0252
68	2.0	5.60	11.20	2.66	5.32 1.55	.052	.1040	.028	.0560 .0650
<b>69</b> 70	5.0	1.02	5.10 3.54	0.31	3.08	.018	.0630	.007	.0245
70	3.5	1.01	3.04	0.88	3.00	.010	•0000	.007	10240
	11.9	1.81	21.55	0.85	10.13	.029	.3426	.014	.1707
Hole	118							135.0	to 143.5'
77	3.8	0.84	3.19	0.66	2.51	.021	.0797	.094	.357
	3.2			*					,
78	1.5	0.51	0.77	0.24	0.36	.011	.0165	.078	.117
	8.5	0.47	3.96	0.34	2.87	.011	.0962	.055	.474
Hole /	119					•	129	0.01 to	134.0'
87	2.5	0.42	1.05	0.19	0.475	300.	.005	.012	.030
88	2.5	0.75	1.88	0.67	1.675	.003	.007	.014	.035
							*******	-	-
	5.0	0.58	2.93	0.43	2.150	.0025	.012	.013	.065

### Comparison of Platinum Assays

		Ledo Composite	oux Samples	Low Jonemit	calculated)	
Hole No	yt.	Pt. ozs.	ozs. x L	028.	Ozs. x L	Sample No.
35	123.0	•10	12,200	.115	14.0300	129
1	26.0	•07	1.820	-079	2.0540	130
8	27.0	.21	5.670	.149	4.0850	131
3 🐧	8.0	.03	.240	.017	.1360	132
4	12.6	.19	2.394	.189	2.5814	133
5	30.5	.07	2.135	-045	1.3725	134
6	27.5	.09	2.475	-134	3.6850	135
13	25.0	.17	4.250	.077	1.9850	136
23	24.0	.21	5.040	.173	4.1520	137
27	57.0	.14	7.980	.068	3.8760	158
38	63.0	•04	2.520	.128	8.0640	139
40	<b>50.0</b>	.82	18.800	.566	16.9800	140
				.218 (out)		
42	77.0	.17	13.090	.171	13.1670	141
	529.6	.148	78.414	.143	75.8459 (uncu	t)

If .30 substituted for ansays in hole 40 cut averages become .130 .128

Ledoux P. M. Assays - Hole No. 35.

Sample No.	Feet	Pt. 020.	028. x L	Au. ozs.	ozs. x L.
148	11.4	0.19	2.166	0.04	.456
143	29.5	0.04	1.180	nil	
144	5.5	0.12	•660	nil	
145	17.0	0.03	.510	nil.	
146	12.9	0.08	.774	nil	
147	19.4	0.09	1.746	0.09	1.746
149	17.2	0.15	2.580	0.39	6.708
149	6.5	0.08	.520	0.12	. 780
	130.4	•		•	-
•	119.4	0.005	10.136	0.081	9.690

# Weighted Average Analysis - Composite Samples No. 1 Orebody.

Tons	ozs.Au/ton	4.Cu.	\$ NI.	<u>\$ 70.</u>	<b>≸</b> 8.	\$ 810 <sub>8</sub>	% insol.
138,097	.039	8.26	1.43	16.90	9.91	34.99	49.40

Analysis	4	Chalcopyrite %	Pontlandito %	Pyrrhotit	o & Excess \$
Cu.	2.25	2.26			
Ni.	1.43		1.43		*
Jo.	16.90	2.00	2.73	8.05	4.12
9.	9.91	2.29	2.34	5.28	
% of ore	30.50	6.55	6.50	15.33	4.12

Total sulfides = 30.50 less excess Fo. (4.12) . 26.38% of ore.

## Core Stored on proporty. (april, 1951)

Hole No. 34 Sheet No. /

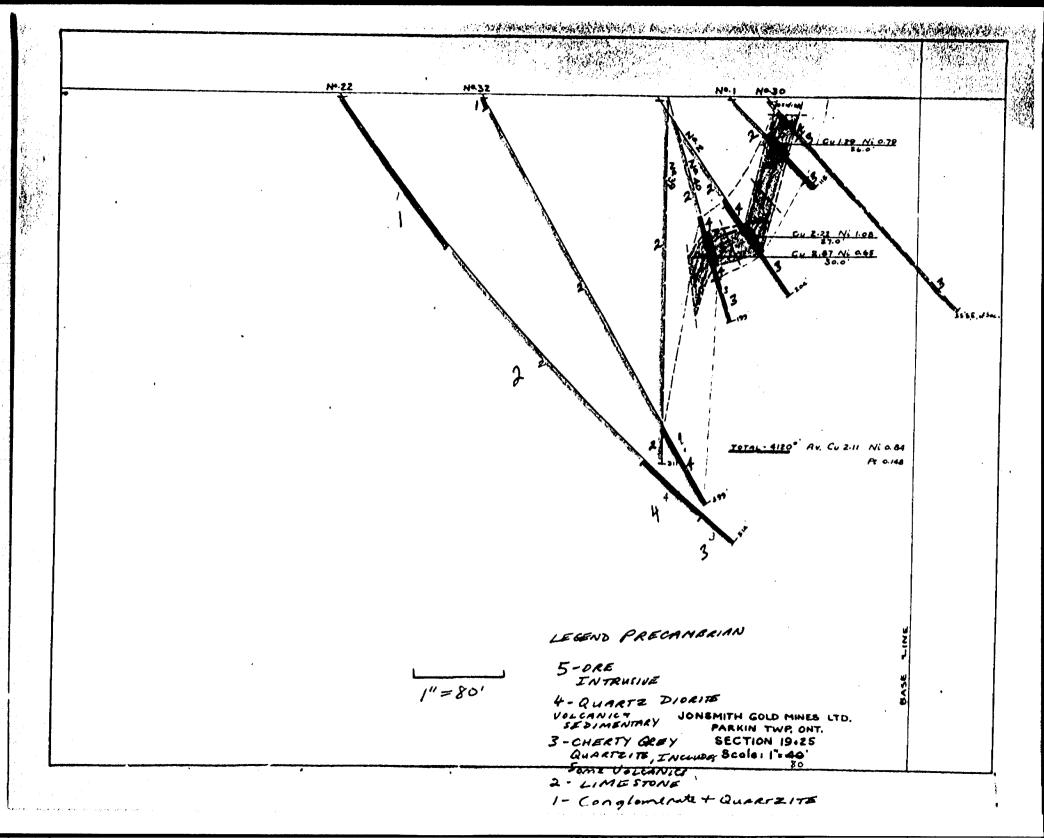
## DIAMOND DRILL RECORD

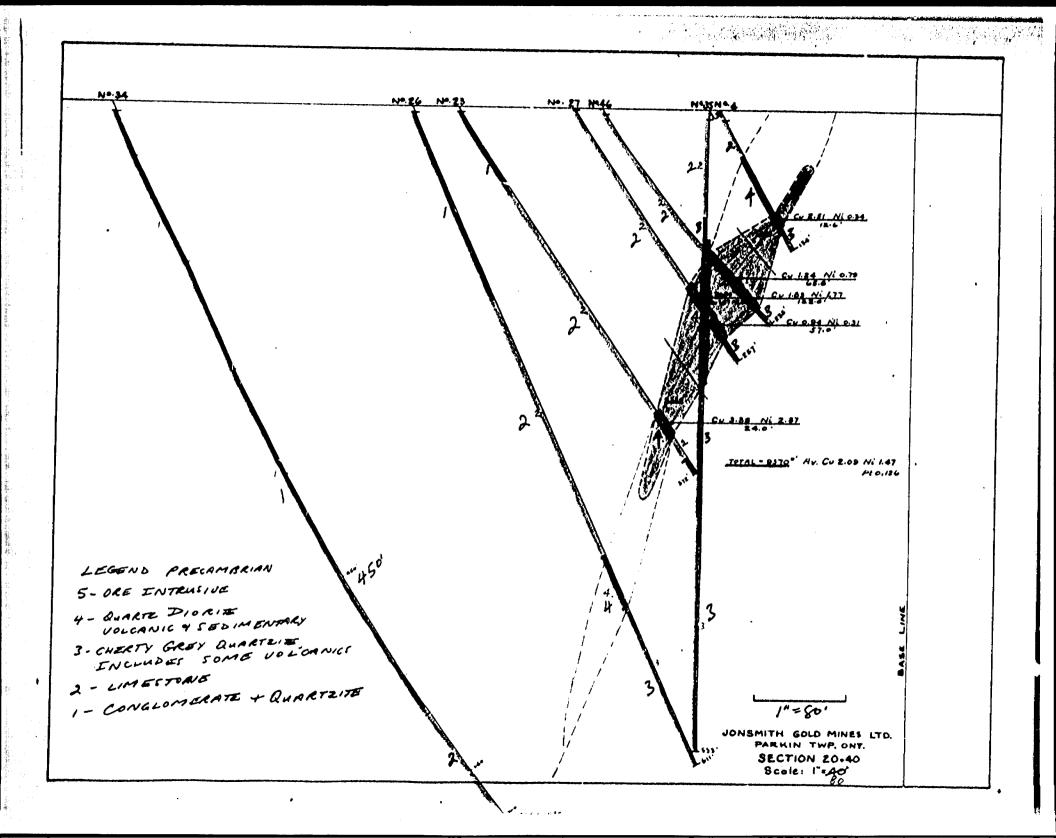
Property JONSMITH GOLD MINES LED		Dip 1	Elev. Collar
Location PARKIN TWP SUBBURY	20	64°	Datum
	/00	64	Date Started JULY 22 1951
	200	64	Date Completed Que 12
Latitude 635 5 of No. 1 2057 42677	600	55	Drilled by LUNGYEAR
Departure AND 525' W.	980	48	Logged by F. Q. LILGE.

Total Footage 1038

Foot		Formation	Sample	Sample	Gold	Gold	Pt
From	To	FUI IIIduvii	Number	Width	Sample	Sludge	Remarks
0	589	CONGLOMERATE Brill (BRUCE)					
589	728						
728	1038	FINE GRAINED SEDIMENTS SOME SOME (SE	RIENT)				
		A SECTION FROM 900 TO END OF HOLE  HAS BLITOST THE APPEARANCE OF  AND PRODUTE. SOME HORNBLENDE					
		PRESENT					
		1038 END Of HOLE.					
		NO MINERALIZED SECTIONS					No SALARLING
	<u> </u>				<u> </u>		

5. Thulman





AY. CV 0.93 Ni 0.91

LEGEND PRECAMBRIAN

5 - ORE INTRUSIVE

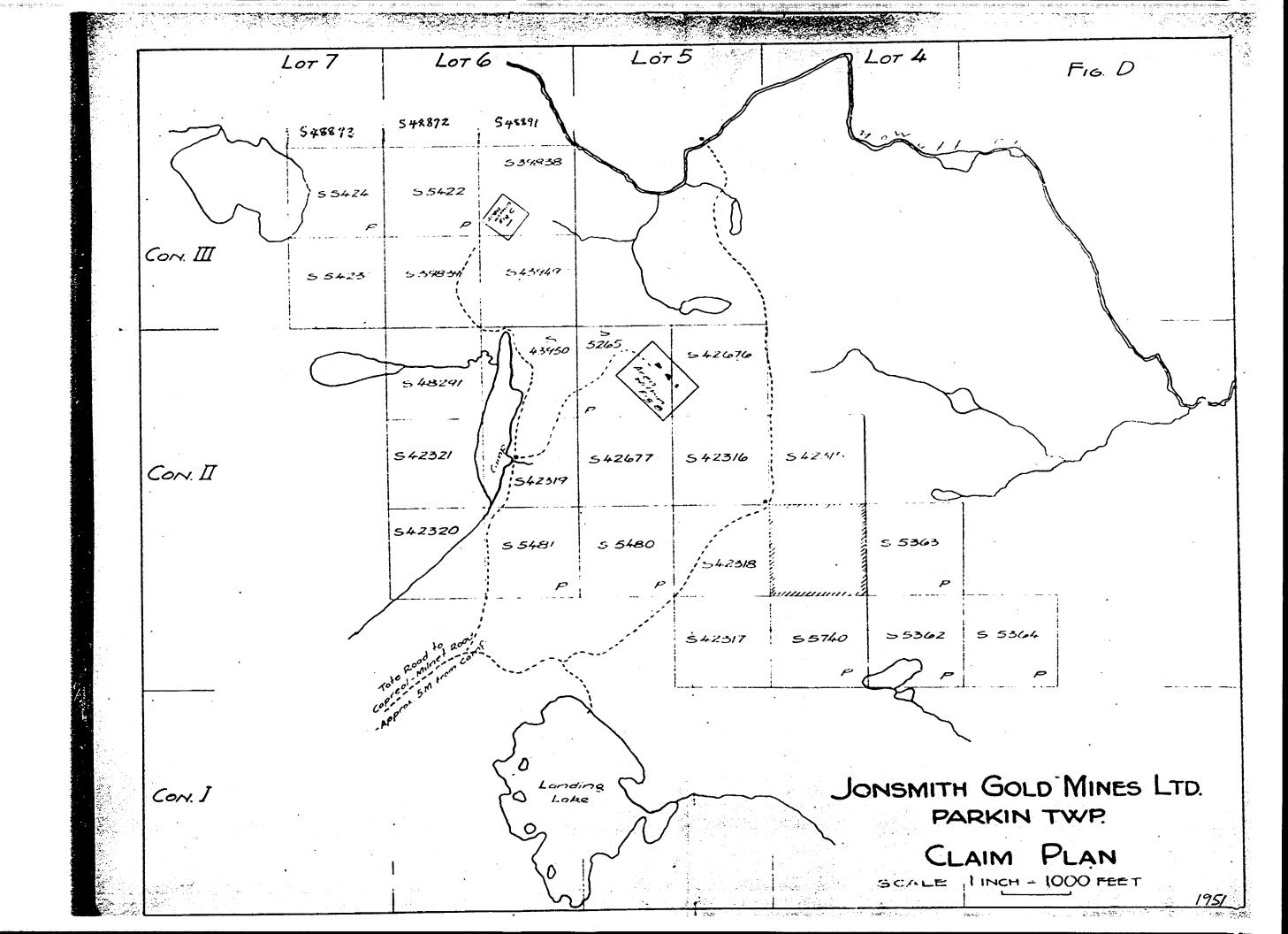
4 - QUARTE DIORITE VOLCANIC + SEDIMENTARY

3- CHERTY GREY QUARTEINE INCLUDES SOME. VOLUNICE

2- LIMESTONE

1- CONGLOMERATE + QUARTZITE

JONSMITH GOLD MINES LTD. PARKIN TWRONT SECTION 16-20 Scoler 1 . 40



ALLISSWALOG PARKIN34 PARKIN

900

June 28, 1947

The President and Directors, Jonsmith Gold Mines Limited, 17 Queen St. East, Toronto, Ontario.

Gentlemen:

Herewith my Report on the Property of your Company in Parkin Township, District of Sudbury, covering the recent programme of development carried out under my direction.

Yours very truly,

WPC/TM

E. G. Bishop and W. P. Corking.

MONETA PORCUPINE MINES, LIMITED
(NO PERSONAL LIABILITY)

SUITE 1808 - 320 BAY STREET
TORONTO 1, ONT.

PLEASE REPER FILE:

November 5th, 1951.

Dr. J. E. Thompson, Ontario Department of Mines, Parliament Bldgs., Toronto, Ont.

Dear Ed:

I attach Hal Wilson's calculations for the tonnages and grade of No. 1 and No.2 orebodies on the property of Jonsmith. I am also sending you Wilson's surface plan and his vertical sections. These are the only copies I have available of the plan and sections, so I may ask to have them back. In my opinion, Wilson has been ultra-conservative in estimating the tonnage. In my figures, after allowing for reasonable extensions of the oreshoot, I arrived at a figure of 180,460 tons with a grade of 2.09% copper and 1.55% nickel. Allowing for 10% dilution, this would be 198,506 tons. In calculating grade I have assumed that the dilution would not contain any metals, and I have also reduced the copper values by 12 percent and the nickel by 4 percent to take care of what I consider may be possible errors in core splitting. On the basis of these figures I arrive at a grade of 1.68% copper and 1.35% nickel, with 0.10 oz. platinum and 0.04 oz. gold. I might say that I think this would be the minimum grade, and I would expect that it would be somewhat higher.

If there is any other information you require, I would be most pleased to supply it to you. All of our technical information is available for your inspection.

Yours very truly,

LR. Burton

Encls.

F. R. Burton

DEPARTMENT OF MINES

PARLIAMENT BURDINGS

(Copred on August 24, 1966)

Sept 7.1954

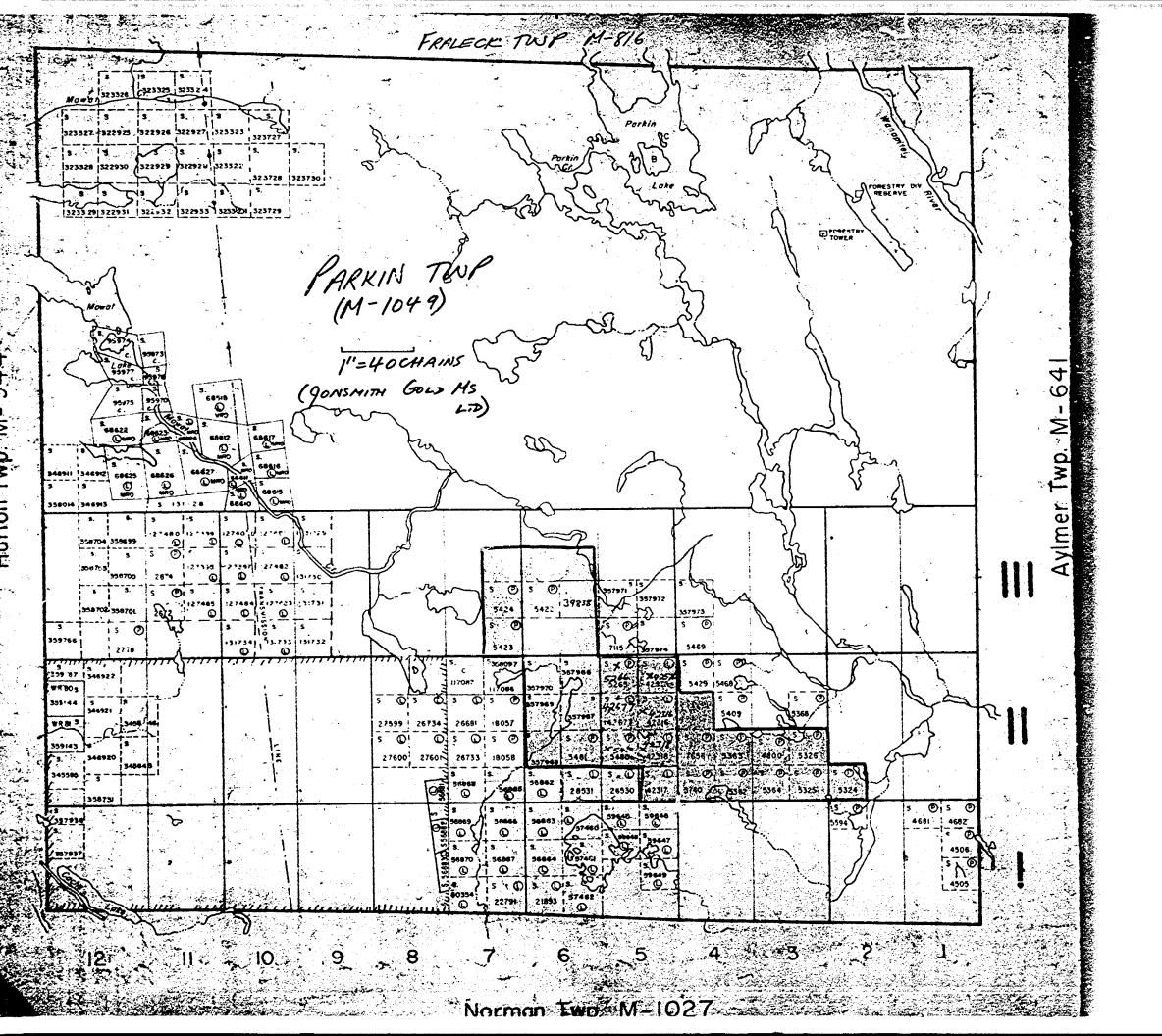
The President and Directors Jonsmith Mines Ltd. Room 906 327 Bay Street Toronto, Ontario

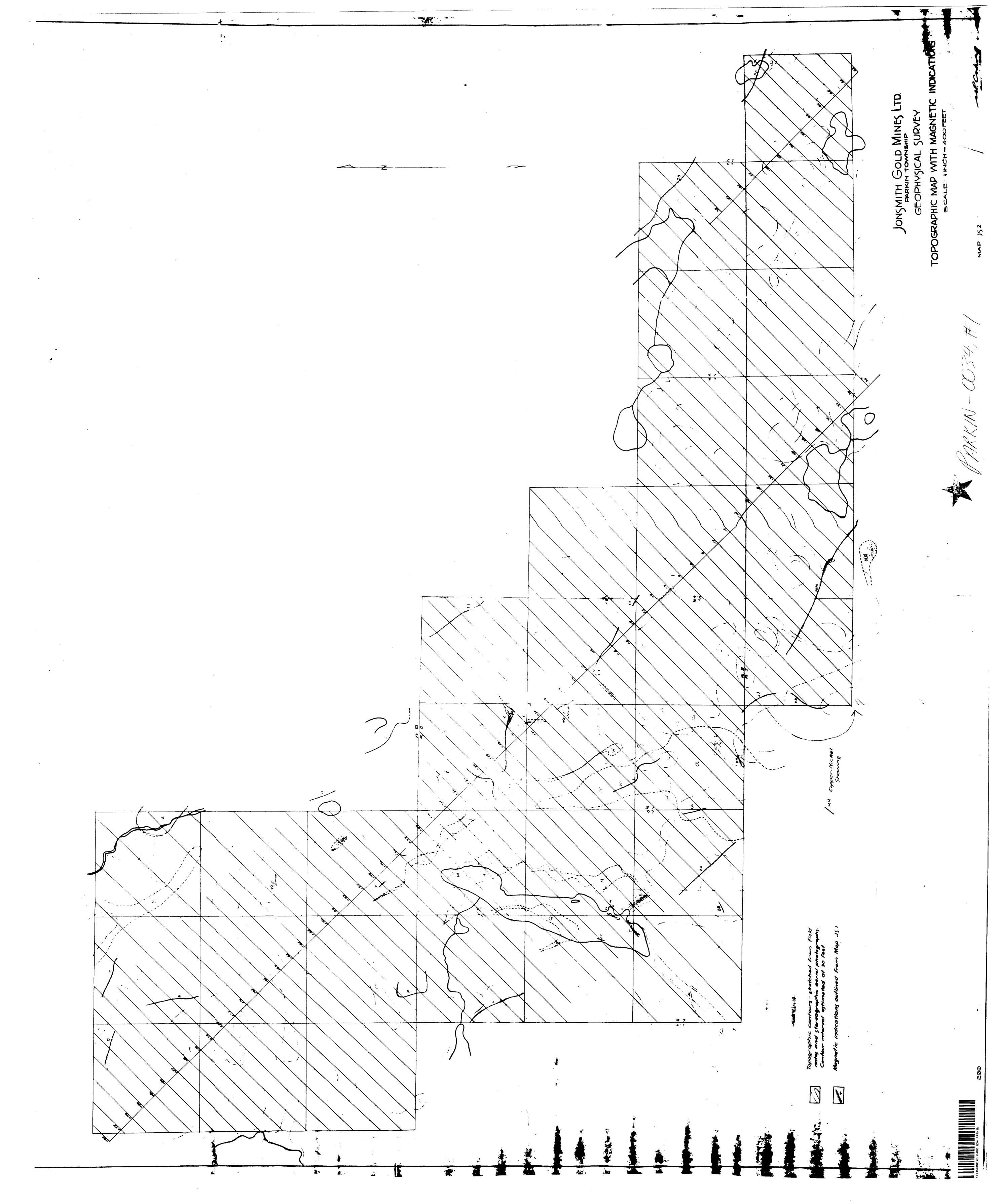
Gentlemen: -

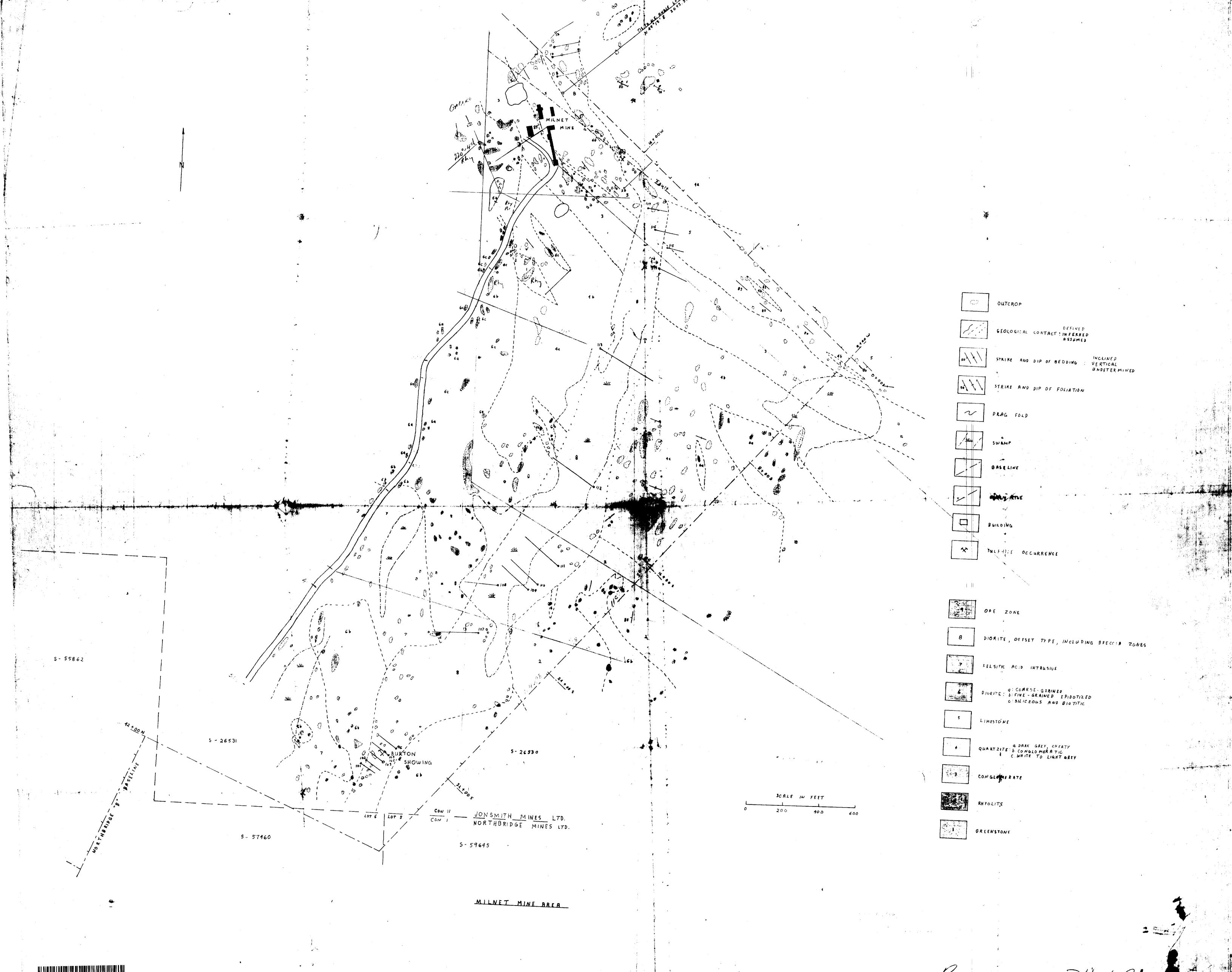
Enclosed is the report on Jonsmith Mines Ltd. along the lines suggested by officials of U.S. Division of Emergency Procurement Services.

The Jonsmith Mines Ltd./is comprised of/29 claims, approximately 1160 acres, in the Township of Parkin, Concession II and III, Lots 2 to 7, District of Sudbury, Province of Catario, Canada, Sudbury Mining Division. The property is approximately 10 miles north of the town of Capreol situated on the northeast rim of the Sudbury Basin.

The property is serviced by paved roads and railroad to Capreol, thence by good gravel roads to the mine property proper. Electric power is supplied by Hydro-Electric Power Commission of Ontario by 8.8 miles of 44.000 volt transmission line, owned by the mine and built in 1952 at a cost of \$104,000.00. Power is supplied at \$45.00 per K. W. year.

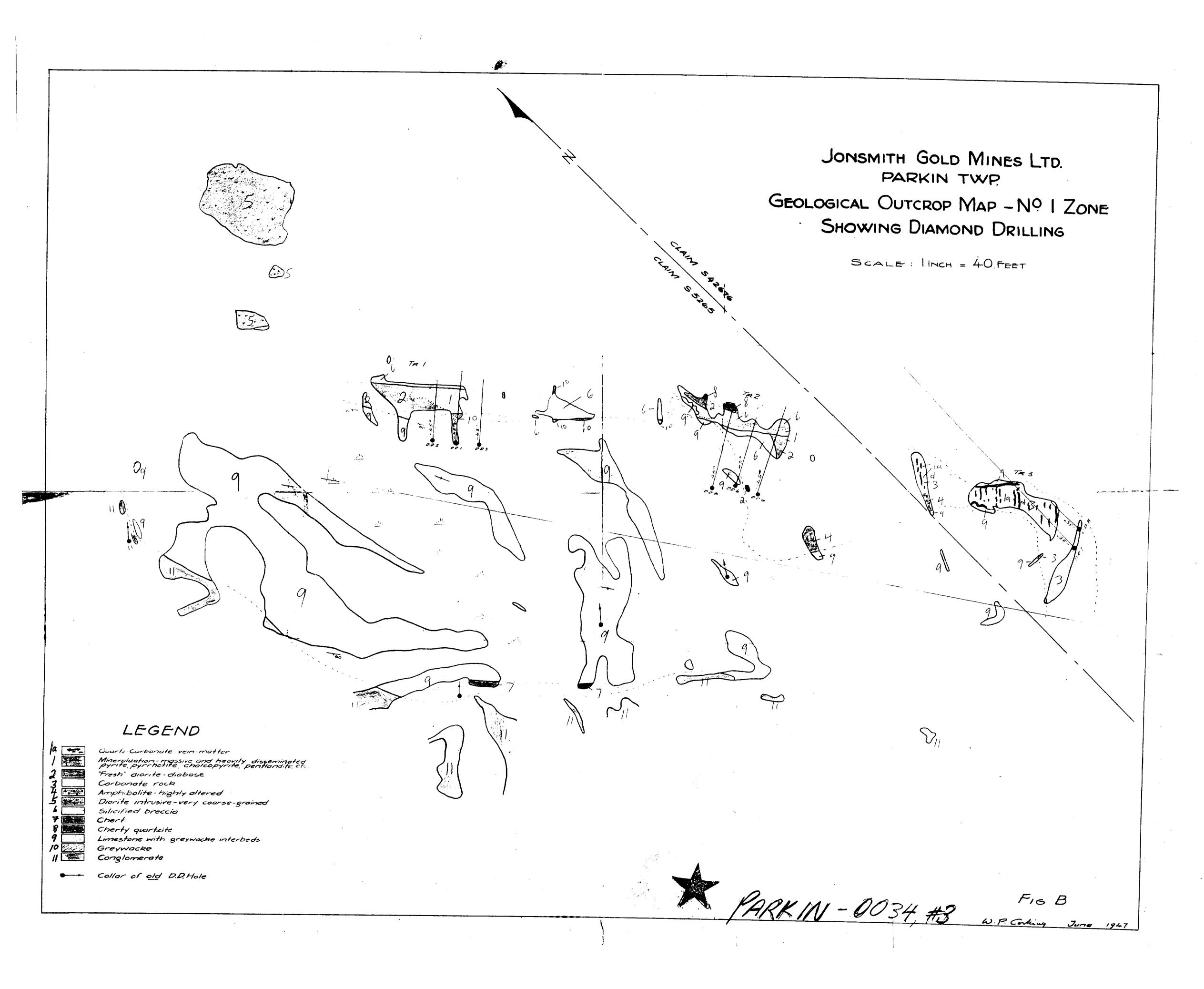






41115SW0106 PARKIN34 PARKIN

PARKIN-0034, #2



### LEGEND

## Post Huronian

Quartz diorite (offset type) including

6 Diorite

Huronian

5 ቆ Limestone

4 🥻 Quartzite

→ Conglomerate

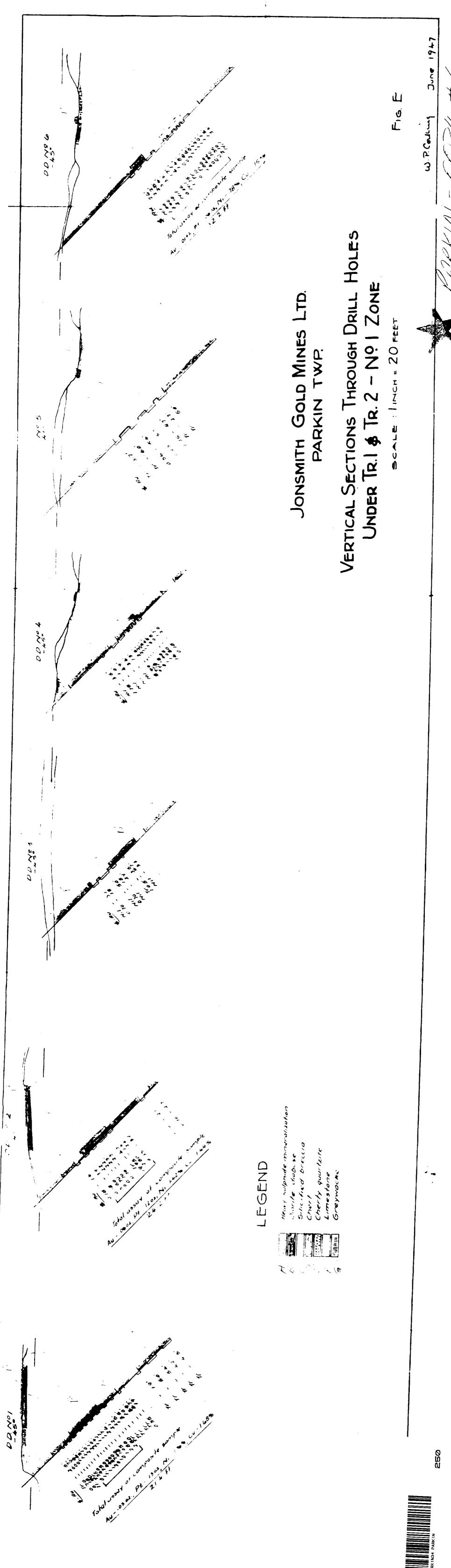
A Surface	8
190'	D:
300'	LS 1 0, 80
4-109	
465' 4-109	
600	5
700'	
800	N. 18 812-14
900	5 2.
1000	4-118
1100'	40 4
1200	4-134
1300' 1400'	10 of 1 130 (s.
AA - Breccia	at
What was	cq Bracha
4-108	4-124
MILNET MINE, PARKIN	TWP 3 and
Scale: 200':1"	e 465' Level Plan. 4-19 from Mine Plans, July 21. 193

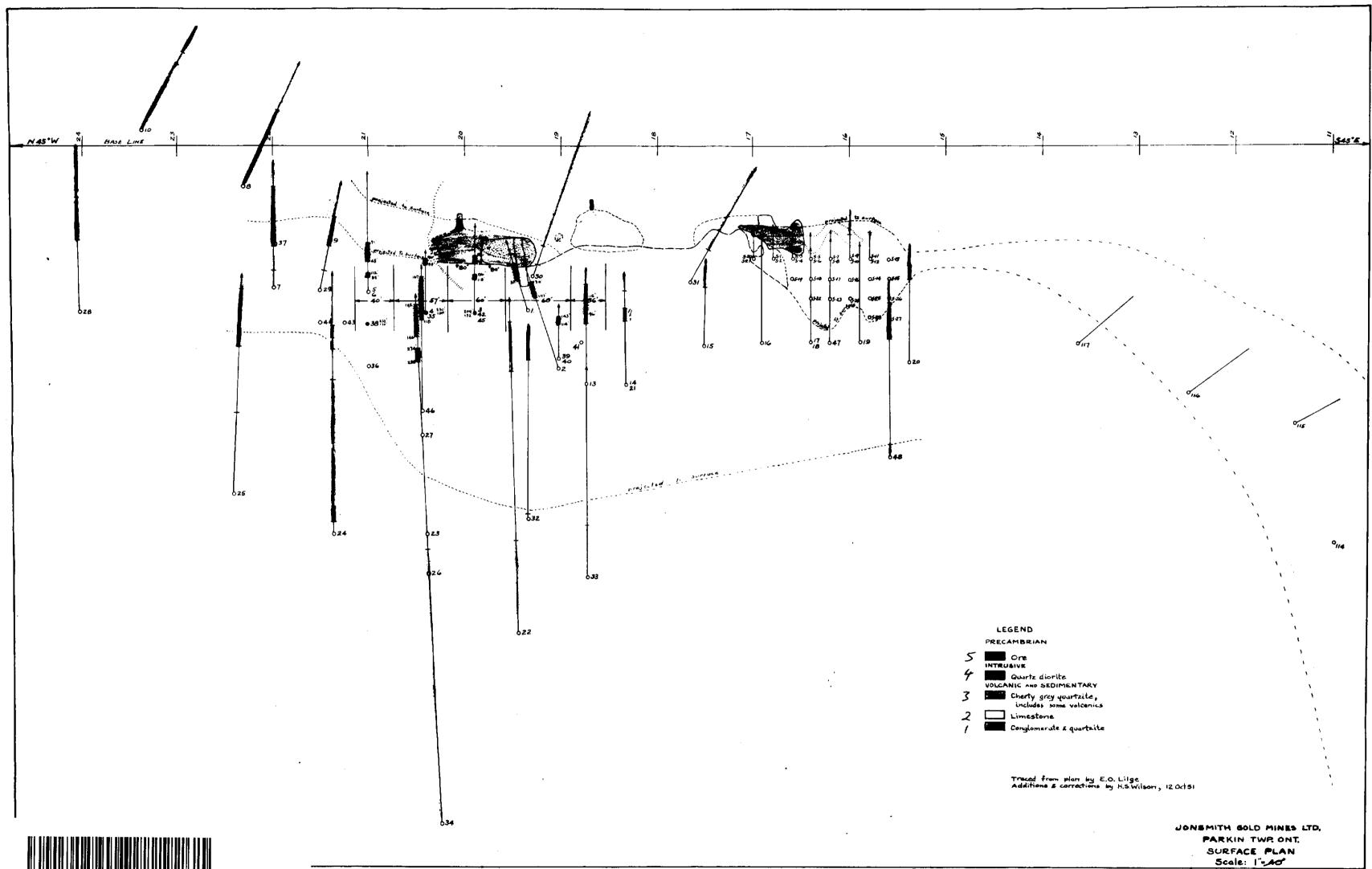


PARKIN-0034,#4

9000 N 8000 N MILNET MINE 465 LEVEL lin = 200 ft.







PARKIN - 00347