



Bergvesenet

Postboks 3021, N-7441 Trondheim

Rapportarkivet

Bergvesenet rapport nr 6088	Intern Journal nr	Internt arkiv nr	Rapport lokalisering	Gradering
Kommer fra ..arkiv	Ekstern rapport nr BA	Oversendt fra	Fortrolig pga	Fortrolig fra dato:

Tittel

BA-rapporter vedrørende Ulveryggen kobberforekomst Finnmark

Forfatter	Dato	Bedrift (oppdragsgiver og/eller oppdragstaker)

Kommune Kvalsund	Fylke Finnmark	Bergdistrikt	1: 50 000 kartblad 19351	1: 250 000 kartblad Honningsvåg
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Fagområde Forekomstbeskrivelser	Dokument type	Forekomster (forekomst, gruvefelt, undersøkelsesfelt) Ulveryggen
Råstoffgruppe Malm/metall	Råstofftype Cu	

Sammendrag, innholdsfortegnelse eller innholdsbeskrivelse
Rapportene er kopier av NGU's bergarkivrapporter (BA):

BA 996, 1283, 4430, 4431, 4477, 4482, 4483,
4626, 4988, 4989

BERICHT

BERICHTIGUNG DES

Repperfjord

UND NORDISKA GRUF A.B.

IN H. B. T. I. S. S. C.

BERICHTIGUNG

von

Verbindender P. H. M. S. N.

Bergarkiv

Rapport nr: 996

Während mit Baumister J. L. O. aus Ludvika in Schweden habe ich von 20-2-1940 eine Reise nach Repperfjord bei Hammerfest in Finnmarken, Norwegen, gemacht, um verschiedene Fragen über eine baldige Inbetriebnahme des seit 1904 untersuchten Kupfererzorkommens, das der schwedischen Gesellschaft Nordiska Gruf A.B. gehört, an Ort und Stelle zu untersuchen.

In Oslo war mir mitgeteilt worden, daß man dreimal wöchentlich Autabusverbindung zwischen Hammerfest und Repperfjord hat. Es war deshalb eine Enttäuschung, bei der Ankunft in Hammerfest zu erfahren, daß der Autoweg der ganzen Winter von Schnee blockiert ist, sodaß überhaupt während des Winters keine Autoverbindung zwischen Hammerfest und Repperfjord besteht.

Nach einer stürmischen Reise mit Motorboot kamen wir auch an Repperfjord an und haben im Hause der Gesellschaft übernachtet, gesprochen und gewohnt.

TRANSPORTWEG.

Maschinen und Materialien für die Ablage und Materialien einen Betrieb müssen vom See nach dem Grubengebiet, sowie Koharz oder es tionskonzentrat von der Grube nach dem Hafen transportiert werden, zu ist deshalb nicht zu vermeiden, daß ein brauchbarer Autoweg zu 100 bis Jahreszeiten offen gehalten wird. Aus einer von Stabsfahnenjunk 1904 gezeichneten Karte (1:4000) kann man sehen, daß ein Ufer entlang nach Vestre Ariselv führt und von hier nach einer B. von etwa 90° dem Aris-Fluß nach dem Grubengebiet folgt.

Die Strecke vom Hafen (auf der Karte als "Kontor" bezeichnet) nach dem Ariselv ist ein Teil des Hauptwegs von Hammerfest nach Karasjok, etwa 3,6 m breit und im Sommer gut gehalten. Sie kann ohne jede Vor-

Arbeiten für schweren Autotransport benutzt werden.

Der private Grubenweg der Gesellschaft vom Ufer den Ariselv entlang nach den Gruben ist teilweise recht steil und überall schmal, durchschnittlich etwa 2 m, und für Pferdetransport gebaut. Ich habe den Eindruck, daß der Weg recht gut gebaut ist, wesentlich auf Steinboden. Auf einer Stelle, wo der Weg am steilen Abhang entlang gebaut ist, war die St. immer niedergebrosen, und es wird auf eine Strecke von etwa 50 bis 60 m nötig, den Weg umzubauen und zu verstärken.

Weiter nach den Gruben, an der genannten Stelle vorbei, geht der Weg wieder über flacheres Land und ist überall auf Gestein gelegt. Ich glaube, daß die Verstärkung und Verbesserung des Wegs keine Schwierigkeiten machen wird, aber es ist neben dem Bau eines neuen Kais unbedingt die erste Arbeit, die ausgeführt werden muß.

K. I. und HAFENVERHÄLTNISSE

Bei dem auf der Karte als "Kontor" bezeichneten Gebäude war ein Holz Kai als Verlängerung des der Gesellschaft gehörenden alten angefangenen Steinkais gebaut. Dieser Holz Kai wurde aber im Januar dieses Jahres vom Hjordsels zerstört.

Tiefenmessungen im Meer sind früher außerhalb des Kais gemacht und zeigen, daß die Verhältnisse für den Bau günstig sind. Ich habe in Hammerfest mit einem Herrn H. U. N., der sowohl Kaianlagen wie auch andere Holz- und Betonarbeiten über ganz Finnmarken gebaut hat, gesprochen. Er meinte, daß es wegen der Eisverhältnisse in Repperfjord zu empfehlen ist, den Kai aus Beton zu bauen. Um die Maschinen und Materialien für eine Anlage empfangen zu können, kann man aber vorläufig einen Holz Kai im Laufe von wenigen Wochen bauen. Für den Betrieb wird es jedoch nötig sein, einen starken Kai, der Eisdruck aushalten kann, zu bauen. Weiter muß ein Lagergebäude für Flotationskonzentrat (1500 bis 2000 t) mit Transportband und Schrapperanlage gebaut werden.

SÜBWASSER - ZUFUHR

Wie man auf der Karte sieht, laufen zwei Flüsse: der Østre Ariselv und der Vestre Ariselv in der Richtung SW-NO nach dem Repperfjord, also parallel der Streichrichtung des Repperfjord-Erzvorkommens. Von diesen zwei Flüssen ist der Vestre Ariselv dem Grubengebiet am nächsten. - Die zwei nahliegenden Flüsse sind durch einen Gebirgsrücken von etwa 30 bis 40 m Höhe von einander getrennt.

Der Østre Ariselv kommt vom See Arisvand, der eine nicht unbedeutende Wasserfläche und Tiefe hat und nach Angaben recht fischreich

ist, im Winter also nicht bis zum Boden gefroren ist.

Der Vestre Ariselv dagegen kommt von unbedeutenden Gebirgsseen, die wie man befürchten muß in der kalten Jahreszeit bodenfrieren werden.

Während unserer Besichtigung oben am Ariselvtal bis zum Grubengebiet war der Fluss vollkommen unter einer hartgepreßten mehrere m dicken Schneedecke verschwunden. Nach Angaben von Hepperfjord-Binwohnern über die Wassermengen in verschiedenen Jahreszeiten fürchteten wir, daß die Wassermenge, die wir im Vestre Ariselv in der jetzigen trockenen und kalten Jahreszeit finden würden, ungenügend für Flotation, Kompressor-Kühlwasser usw sein würde.

Es war deshalb eine erfreuliche Überraschung, durch Weggraben von Schnee und Eis einen frischen Strom von etwa 1 m Breite und 20 cm Tiefe zu finden, von dem ich eine 10 Liter Probe für Wasseruntersuchungen nahm.

Die für die Anlage nötige Wassermenge wird etwa bei vollem Ausbau bis 120.000 t Koharz per Jahr:

für Flotation	1350 l/min.
" Kompressor	150 " "
" Haushalt 100 Personen .	15 " "
" Spülwasser Bohrmaschinen	50 " "
" Dieselmotor	50 " "

zusammen 1615 l/min., also etwa 27 l

Wesentlich mehr als diese Wassermenge haben wir jetzt in Vestre Ariselv beim Grubengebiet beobachtet. Nach Angaben sind eben die Monate Februar und März die trockensten und wasserärmsten. Wir sehen die Frage der Wasserzufuhr für eine Grubenanlage durch Versorgung von Vestre Ariselv mit einer Pumpenanlage, die natürlich frostfrei montiert werden muß, als gelöst an.

DIE WETTERVERHÄLTNISSE

haben wir durch Erkundigungen bei Leuten von Hammerfest und Hepperfjord so genau wie möglich untersucht, da ich es für wichtig halte, daß man sich so genaue Kenntnisse wie möglich über die Schwierigkeiten für Anlage und Betrieb durch die in West-Finmarken im Vergleich zu den skandinavischen Verhältnissen harten Wetterverhältnisse durch Schnee, lange Kälteperioden und Stürme verschaffe.

Allgemein kann man damit rechnen, daß die Kälteperiode mit Temperaturen unter 0° Mitte November bis Mitte Mai oder Anfang Juni dauert.

In den letzten Jahren ist der Boden bis Weihnachten schneefrei geblieben, dagegen schneebedeckt von Januar bis Juni.

Im übrigen kann man sagen, daß die Gegend von Hepperfjord stürmisch ist. Die schlimmsten Winde sind die südlichen und südwestlichen, die mit kaltem Wetter kommen. Die nördlichen und rein westlichen Winde kommen mit feuchtem und etwas milderem Wetter und sind deshalb nicht so gefürchtet. Die Anlagen bei der Grube werden in einer Meereshöhe von etwa 350 bis 360 m liegen; die Temperatur ist hier einige Grad niedriger als am Fjord. Es ist da oben auch beinahe immer Wind ~~und~~ oder Sturm. Die Anlagen müssen deshalb möglichst im Berg eingeschossen werden (Lagerbunker, Verkleinerungsanlage^{usw}). Von Baumeister JANSEN wurde vorgeschlagen, Bohrschmiede, Reparatur von Bohrmaschinen, Lager für Materialien, Grube usw. in einem Querschlag des Olle-Stollens, nachdem genügend Raum durch Strobearbeit geschaffen worden ist, einzubauen. Dieser Vorschlag ist nach meiner Meinung gut.

BEWÖHNUNG DER ANLAGE

Für rechnen mit einer Belegschaft von insgesamt 100 Arbeitern, 5 bis 6 Beamten sowie 2 Ingenieuren, für die Wohnungen geschaffen werden müssen, d.h. für die Arbeiter Schlafzimmer mit 4 Betten, Kantine und Aufenthaltsraum samt Küche möglichst mit finnischem Dampfbad.

Die Beamten können in Einzelzimmern im ersten Stock des Bürogebäudes wohnen. Für die 2 Ingenieure muß ein kleines Haus mit 3 Schlafzimmern samt Kuchenzimmer und Bad gebaut werden. Man muß also an Wohnungen bauen:

- 1 Ingenieurwohnung im 1. Stockwerk
- 1 Büro mit Zimmern für Beamte im ersten Stock
- 4 Arbeiterwohnungen für je 25 Mann
- 1 Kantine und Küche.

Wie sonst in Norwegen werden sämtliche Gebäude am besten aus Holz gebaut. Es ist zu empfehlen, daß sämtliche Häuser von einer Spezialfirma gebaut werden, natürlich nach eingeholten Angeboten von verschiedenen Firmen. Wir haben vorläufig eine Besprechung gehabt mit der Hæverfestfirma. H. Hæver Materialhandel, dessen Inhaber große Erfahrung von Bauarbeiten in Finnmarken hat und der uns Angebote vorläufig geben wird.

ENERGIE

Für die ganze Anlage wird man anfangs etwa 500 PS benötigen, nämlich für:

Flotation	300 PS
Kompressor	120 PS
Licht	20 PS
Schneid- und Verastatt	20 PS
Ventilatoren u. Kleinmaschinen	
Grube	10 PS
Sonstiges	10 PS

Man kann für eine Kraftlieferung mit 3 Möglichkeiten rechnen:

- 1- eigener Kraftausbau
- 2- Kraftmiete von Porsa Kraftzentrale, die der Gemeinde Hammerfest gehört
- 3- Dieselmotoranlage bei der Grube.

Die Verhältnisse liegen für eigenen Kraftausbau nicht günstig, und ich nehme an, daß ein solches Projekt auch wegen Zeitverlust nicht in Frage kommen kann. Die Porsa Kraftzentrale, seiner Zeit von Porsa Kobbergruber gebaut, gehört jetzt der Gemeinde Hammerfest, die eine Hochspannungslinie mit 22.000 Volt, teils in Kabel, gebaut hat.

Es ist weiter eine Hochspannungslinie von Porsa nach Faegfjord zur Versorgung der Kvalsund Gemeinde geplant. Von Faegfjord nach Repperfjord ist es nur etwa 5 km. Eine Kraftüberführung von Porsa nach den Repperfjordgruben scheint somit günstig zu werden. Jetzt sind aber nur 500 PS von den im Wasserfall vorhandenen 1200 PS ausgebaut, es wäre als nötig, sowohl Wasserröhrlleitung wie Turbine mit Generator zu beschaffen.

Die Gemeinde Hammerfest ist in einer schlechten ökonomischen Lage und es wurde mir mitgeteilt, daß es ausgeschlossen ist, daß die Gemeinde das nötige Kapital für diesen Ausbau selbst beschaffen kann.

Ich glaube aber, daß es möglich ist, eine günstige Kraftmiete durch Vorauszahlung eines Betrags, der den weiteren Ausbau der Kraftzentrale ermöglicht, zu erhalten.

Man kann weiter damit rechnen, daß der Kraftverbrauch in Finnmark in der hellen Jahreszeit von Mai bis August gering ist, weil es dann Tag und Nacht hell ist. Die muß man für eine Grubenanlage günstig nutzen können. etc

Ich habe vorläufig kurz mit dem Stadttechniker in Hammerfest über die Kraftfrage gesprochen, habe aber vor meiner Abreise nach Repperfjord ein Zusammenkommen verabredet. Die verschiedenen Möglichkeiten werden weiter unten diskutiert.

Die letzte und wahrscheinlich im Anfang sicherste Kraftquelle ist eine Dieselmotoranlage mit direkt gekoppeltem Generator. Sowohl Bre-

...sondierel sind in Preis gestiegen, und auch wegen dem relativ teuren Transport der Ole um der andern Materialien wird die Dieselmotorkraft im Vergleich mit Wasserkraft jetzt noch teurer als vorher. Ich fürchte, das man trotzdem mindestens anfangs unter allen Umständen mit einem Dieselmotor anfangen muß, empfehle aber, doch eine Hochspannungslinie zu bauen, selbst wenn die Gemeinde Hammerfest vorerst nur ungenügende Energiemengen abgeben kann.

UNTERSUCHUNGEN

Ich hatte gehofft, das ich sämtliche Untertagearbeiten hätte besichtigen können. Nachdem wir aber einen ganzen Tag durch hartgepreßte Schneedecke etw. 4 m tief gearbeitet hatten, um die Stollenöffnung des Olle-Stollens zu finden, mußte ich es in dieser Jahreszeit aufgeben, auch die Johns- und Erichs-Gruben zu besichtigen.

Der Johns Stollen ist übrigens Anfang Februar (bevor die später gekommenen großen Schneemengen die Arbeit erschwerten) eine Erzprobe von etwa 1,2 t an Professor B. ING, Stockholm, geschickt worden.

Ich bin aber die ganze Olle-Grube durchgegangen. Diese besteht aus einem etwa 180 m langen Stollen, dessen Mundloch 356 m über dem Meer liegt, senkrecht zur Streichrichtung des Vorkommens. Man kann durch auf dem Stollenstoß abgesetzten grünen Kupferkarbonatstreifen und Flächen leicht feststellen, wo man erzführende und wo man erzfreie Partien hat.

Es scheint die erste Strecke des Stollens erzfrei zu sein. Etwa 20 m vom Mundloch ist ein kurzer Querschlag nach Norden getrieben, ohne Erzmassen von Bedeutung getroffen zu haben. Ein Feldort senkrecht auf dem Stollen also in der Streichrichtung des Vorkommens ist sowohl gegen SW wie NE etwa 80 m getrieben und hat augenscheinlich recht gute Erz getroffen. Dieses Feldort ist nicht im Bericht des Ingenieurs GOLDKUHLE genannt und ist deshalb wahrscheinlich später getrieben. Ich habe Proben für Flotationsversuche, die von mir ausgeführt werden, genommen. Weiter habe ich Proben für Flotationsversuche von dem großen Schurf, der etwa grade über dem Olle-Stollen getrieben ist, genommen. Weiter habe ich noch Proben von Erz aus den Bergeshalden außerhalb Johns Stollen genommen.

Sowohl nach dem Bericht von GOLDKUHLE, wie noch mehr nach dem Berichte von OTTO STILMANN (1913) liegen wahrscheinlich die ganz überwiegenden Erzmengen in "Olles Malmer". Es muß deshalb interessant sein, Flotationsversuche von diesen Erzen auszuführen.

Nach Besprechungen mit Bauleiter JAKSON und nach Besichtigungen der Ort und Stelle werden wir aus diesem Grunde auch vorschlagen, daß zur Transportverbindung zwischen Grube und Aufbereitungsanlage der Olle-Stollen als vorläufiger Förderstollen gewählt wird, weil hierrdurch wahrscheinlich die durch die Arbeiten in "Kvadraten", "Segmenten", "Triangelen" und Olle-Stollen gefundenen relativ großen Erzpartien, wesentlich über der Olle-Stollen-Sohle liegend, hierrdurch einfach und schnell abgebaut werden können. Ein anderer Grund, weshalb wir bei diesem Plan geblieben sind, ist, daß wir zwischen Olle-Stollen und John-Stollen einen günstigen Baugrund für die Aufbereitungsanlage gefunden haben, wo die Oberfläche des Erzbunkers in demselben Niveau wie der Olle-Stollen liegt. Falls man wünscht, wird es auch keine Schwierigkeiten machen, die wahrscheinlich etwas reicheren John-Erze direkt vom John-Stollen nach dem Erzbunker der Aufbereitungsanlage unseres Plans zu fördern.

ZUGANG VON GELEHRTEN ARBEITERN

Wegen der isolierten Lage der Gruben und wegen der harten Wetterverhältnisse wäre es vorzuziehen, sämtliche Arbeiter aus der Gegend oder mindestens von West-Finmarken zu nehmen.

Ich muß leider sagen, daß mein Eindruck von den Einwohnern, die ich gesehen habe, und auch die Erfahrung anderer Betriebsingenieure mit den Leuten hier nicht sehr günstig sind. Die Bevölkerung ist oft eine Mischung von Norwegern oder Lappen mit Lippen, oder auch von rein finnischer Abstammung. Ich glaube, daß man alle oder die meisten Facharbeiter, wie Holzarbeiter, Motorarbeiter, Mechaniker, Elektriker, Grubenarbeiter (besonders Bohrarbeiter) und Flotationsarbeiter von einem anderen Grubengebiet wie z.B. Rana, wo von den stillgelegten Gruben: Bosmo, Hoffell und Dunderland viele Facharbeiter arbeitslos geworden sind, nehmen muß.

Es wird wahrscheinlich vorteilhaft sein, die meisten Bauarbeiter für die Anlage nach Angebot an eine Baufirma in Hammerfest oder Tromsø zu übergeben. Den schwierigen Maschinentransport dagegen wird man wahrscheinlich am vorteilhaftesten selbst übernehmen, dasselbe gilt natürlich für Montage und Inbetriebsetzung der Anlage.

BERECHNUNG

Die Aufbereitung für das Hepperfjordkupfererz wird heutzutage als eine einfache Flotationsaufbereitung gebaut werden. (Da nur Cu-Mineralien, nur ein Flotationskonzentrat). Man kann wahrscheinlich, weil es sich hier um kupferreiche Mineralien (Kupferglanz und Bornit) handelt, mit mindestens 35 % Cu im Konzentrat rechnen (siehe Bericht über frühere Aufbereitungsversuche von GOLDKUHL).

Mit einem Ausbringen von 90 % und Roherz mit durchschnittlich 1,6 % Cu braucht man: 24,3 t Roherz per t Konzentrat. Die produzierte Konzentratmenge von 60.000 t Roherz wird also jährlich: 1.470 t, von 120.000 t Roherz 4.970 t jährlich mit 35 % Cu, ergeben.

Die oben genannte Berechnung ist natürlich keine endgültige und eine nur angenommene, reicht aber für die Überlegung der günstigsten Lage der Aufbereitung aus zum Vergleich der zu transportierenden Mengen Kupfermaterials.

Es gibt zwei Alternativen für die Lage der Aufbereitung:

I: Die ganze Aufbereitung wird im Grubengebiet und in diesem Fall so nah wie möglich an Förderstollen aufgestellt, ebenso wie Kompressor, Zerkleinerung und Dieselmotor (falls man nicht genügend elektrische Kraft bekommt).

II: Im Grubengebiet werden nur Kompressor und Buckengrobbrecher aufgestellt (damit nicht zu große Erzstücke von der Grube mit der Seilbahn gefördert werden müssen). Feinzerkleinerung, Flotation und Kraftanlage werden beim Hepperfjord Kll gebaut.

Nach I kann man ohne Hochspannungsstrom mit kurzen Leitungen von der Kraftstation arbeiten. Der Transport von der Grube zum Kll wird von etwa 2.500 t (5.000 t) Kupferkonzentrat jährlich, für welche Mengen eine Seilbahn nicht rentabel sein würde, mit Lastauto, das die andere für Grube und Flotation nötigen Materialien zurücknimmt, vorgenommen.

Nach II werden 60.000 t (resp. 120.000 t) Roherz mit 1,6 % Cu nach der Aufbereitung beim Kll transportiert und für diese Mengen kann nur eine Seilbahn in Frage kommen. Da die 2 Kraftverbraucher Grube und Aufbereitung getrennt liegen, muß eine Hochspannung zwischen Hepperfjord Kll und Grube gebaut werden. Was die Seilbahn betrifft, so kann diese entweder in gerader Linie über die Gebirge wie auf der Karte von 1935 gezeigt wird oder durch das Kristal zur Mündung des Hepperfjordelva in den fjord gebaut werden. Nach vorgenommenen Untersuchungen und nach Erfahrungen von anderen norwegischen Gruben muß von dem Bau

The Copper Ore Deposits

on

U l v e r y g g e n M o u n t a i n ,

near

Ripperfjord, Finnmarken.

Detailed Description of Development Work.

THE JOHN'S MINE.

Underground work

The tunnel T /see sketch 1/ which gives access to the John's Mine is about 8.800 feet S.W. from the shore of the fjord, on the line of the strike of the lode. It is between the Erik's tunnel N.E. and the Ole's tunnel S. W. at a distance of 980 feet from the former and 1.100 feet from the latter. The elevation of the surface above the John's tunnel is 1.270 feet above the sea, while the tunnel itself is 80 feet below the outcrop of the lode. A Shaft /S/ has been sunk from the tunnel level about 60 feet deep. The lowest present ore exposure in this mine is , therefore, 140 feet below the surface. The tunnel has been driven into the mountain side almost perpendicularly to the strike of the lode.

At a distance of 44 feet from the mouth of the tunnel drifts have been run E. E. and S. W. on the strike of the lode, due north and due west respectively, the former 100 feet long, the latter 90 feet. The E. E. drift /D.E. passes through good ore its entire length, with the exception of 18 feet, samples of which returned only 0,33 % copper. The first 60 feet of the N.E. drift from the tunnel contain 3,52 % copper. then follow 18 feet containing 0,33 % copper, and finally 24 feet containing 1,25 % copper. The whole block, 100 feet long, contains therefore, an average of 2,42 % copper.

At the N.E. end of the drift, cross-outs have been driven N.W. and S.E., the former about 13 feet long, the latter

35 feet. At the S.E. end of the latter an upraise /U/ has been made to the surface. A poorer zone commences, evidently at the line of these cross-cuts. Samples from them returned from 0,1 to 0,7 % copper only.

Near the junction of the tunnel and drifts, the above-mentioned shaft /S/ has been sunk in pay ore.

The drift /D.W./ extends S.W. from the tunnel for a distance of 90 feet, but is not sufficiently mineralised to justify ore extraction. The samples taken here average about 0,20 % copper. At the S.W. end of this drift a cross-cut /C2/ runs N. W. for a length of about 90 feet. The first 20 feet of this crosscut from the south W. drift /B.W./ are poorly mineralised, containing about 0,20 % copper. But at this point the mineralisation commences to increase in value until it rises to 0,5, 0,6, 0,8 % copper, reaching 3,15 % at a point 37 feet from the starting point of the cross-cut /C2/, at its junction from the S.W. drift /B.W./. At this point a drift /D1/ has been run S.W. through good ore for a length of 90 feet. The first 30 feet of this drift going S.W. from the cross-cut /C2/ contain 2,14 % copper, the next 30 feet, 2,31 copper, while the remaining 30 feet to the face of the drift returning 1,61 % copper, resulting in an average of 1,95 % for the whole drift, the face still being in good ore, containing 1,7 % copper. The cross-cut /C2/ continues from this drift /D1/ about 50 feet N.W. mineralised to the face, the samples returning an average of 1,62 % copper.

The adit level /T/ leading into the mine has also been extended N. W., but its direction bearing heavily S.W., it passes soon into the area of poorer mineralisation. While the assay returns for the first 25 feet of the tunnel extension showed 2,01 % copper, the average of the remaining 55 feet is only 0,7 % copper, resulting in an average of 1,13 % copper for the whole length of the cross-cut.

Ore being exposed N.E. and S.W. of the tunnel, a bore hole was started near the junction of the tunnel, and drifts in a direction S. 87° W., dip -- 44° from the horizontal. This bore-hole is driven therefore, through and immediately below the zone of poorer mineralisation between the two ore bodies exposed in the tunnel level. The first 14 feet of this bore-hole passed through ore containing

2,50 % copper. The value then decreased gradually, passing through a zone of poorer mineralisation for a distance of 46 feet, when it reached ore again, containing, however, only 0,87 % copper. After penetrating this for a distance of 60 feet it reached a zone of poor mineralisation and was abandoned after reaching a total length of 273 feet.

To ascertain whether this decrease in value was local or permanent, another bore-hole was driven. For a distance of 35 feet from the orifice the cores averaged 1,24 % copper, decreasing thence gradually in value to 0,6 and 0,4 % and, after a total depth of 55 feet had been reached, the hole was abandoned.

SURFACE WORK.

Nine cross-cuts, from 50 to 280 feet long, constitute the development on the surface of the John's Mine. The lode is plainly exposed, and is easily followed. The farthest ore exposed N.W. is 514 feet from the mouth of the John's tunnel in cross-cut II. /about 135 feet long/, in a direction perpendicularly to the strike of the lode. The N.W. limit of the lode, the hanging wall, has, however, not been reached here so far.

The mineralisation in this cross-cut II. is poor, averaging about 0,5 % copper, although for a small distance this cross-cut returns 1,1% copper, indicating, possibly, the starting point of a block of richer mineralisation.

Immediately S.E. from cross-cut II. and at a distance of about 40 feet N.E. on the strike of the lode, a cross-cut /I/ 72 feet long exposes the ore again. For about 30 feet at its N.W. end, practically on a continuation of cross-cut II., the mineralisation is poor, starting with 0,5 % copper, but rising in value gradually in its S.E. direction until it reaches 1 % at a distance of 30 feet, increasing in value until it reaches 3,6 % copper for the last 15 feet.

About 40 feet of the S.E. end of this cross-cut /I/ return an average of 1,5 % copper.

Immediately S.E. from this cross-cut the lode crops out boldly.

About 180 feet N.E. from the N.W. end of cross-cut /I/ and parallel with cross-cut /III./ has been driven, running about

100 feet N.W. The average mineralisation of this cross-cut III, is 0,62 % copper while it averages 1,5 % copper for a distance of about 30 feet in the N.W. end of the cross-cut.

About 173 feet N.E. on the strike of the lode from cross-cut III., or about 400 feet N.E. from cross-cut II., a long cross-cut /IV/ has been driven for a length of 280 feet through a zone of poor mineralisation, averaging about 0,5 % copper. For a length of about 35 feet at its N.W. end the samples returned 1,32 % copper.

About 300 feet S.W. from the John's tunnel, a cross-cut /V./ 165 feet long, has been driven through a zone of poor mineralisation, returning a little over 0,1 % copper.

Parallel with this cross-cut and at a distance S.W. of about 130 feet from the John's tunnel, cross-cut VI., about 65 feet long, exposes ore containing 1,76 % copper for a length of about 30 feet in its N.W. half, or 1,23 % for the whole length of the cross-cut.

Almost immediately above the John's tunnel a cross-cut /VII./ about 55 feet long, exposes good ore in its entire length, containing 2,53 % copper.

Crosscut IX., about 50 feet long, driven above the N.E. part of drift D.N., marks evidently the transition of this richer block into a poor block of ore again. It contains only 0,61 % copper while in its continuation N.W. , cross-cut /VIII./, about 100 feet long, traverses the same poor block of ore, containing not much more than traces of copper.

The result of the sampling of the John's Mine show unmistakably that, while the whole lode is mineralised, blocks of richer mineralisation in this ore zone alternate with those of poorer mineralisation. The assay returns demonstrate that in all blocks the richest ore is about in the centre of the block for some distance, decreasing in value gradually in each direction until the limits of the richer blocks merge into the poorer blocks.

The distance between the furthestmost S.W. ore exposure on the John's Mine and the furthestmost N.E. ore exposure on the adjoining Olle's Mine is about 400 feet. No exploratory work of any description has been done in this space, although the lode crops out boldly and promising.

The distance between the furthestmost N.E. ore exposure on the John's Mine and the furthestmost S.W. ore exposure on the adjoining Erik's Mine is about 550 feet. No exploratory work of any description has been done in this space.

Ore probably in Sight.

Block P.--- This block is exposed underground by the drift /D.N./ for a length of over 100 feet, averaging 2.45 % copper. It is again exposed on the surface in cross-cut VII. The limit of the present ore exposures in the John's Mine at depth is 60 feet below the tunnel level, or 140 feet below the apex of the lode.

To ~~xxx~~ prove the continuation of this ore body, as exposed at the surface and at the tunnel level, a bore hole was driven by a diamond drill near the junction of the tunnel, and drifts pointing N. 48° W. at an angle of 26° from the horizontal. This bore hole is 126 feet long, and it penetrates pay ore for a distance of 33 feet averaging 1.3 % copper. After passing for a further distance of 30 feet, through a zone of poor mineralisation containing 0.2 % copper, it reaches pay ore again containing 1.46 % copper, which it penetrates for a further distance of 26 feet. At this point it reached a zone of poorer mineralisation again, and was abandoned after reaching a total length of 126 feet. Assuming then, that the further development work proves the reasonable assumption correct, that the ore blocks between the present limits of their exposure maintain their characteristics throughout, we should have a block of ore here for extraction 100 feet long, 56 feet wide and 140 feet deep, or 56,000 tons of ore at an average of 2.18 % copper per ton.

Block P 1.--- This block is exposed underground by the S.W. drift /D.I./ for a length of about 100 feet, by a cross-cut C 2 for a length of about 50 feet, and on the surface for a length of about 35 feet. This would represent, under the assumption stated above,

a block of ore 100 feet long, 140 feet deep and 85 feet thick, or 85,000 tons of ore at an average of 1.86 % copper per ton.

Block 3.----- The only ore exposure representing this block we find is cross-cut I., on the surface for a length of about 50 feet, and in cross-cut II. for about 20 feet. It should be noted that the average mineral contents for this 50 feet in cross-cut I. is 1.7 % copper, the value of the mineralisation increasing from 1 % at the N.W. end of the ore exposing part of the cross-cut to 3.6 % at the S.E. end of the same part.

It appears, therefore, reasonable to assume, having due regard to the characteristics of the ore blocks which are at present known to a greater extent, that the cross-cut has only exposed one-half of the thickness of the ore block. It would, however, be beyond the limit of conservatism to state that we have a block of ore here 100 feet square and 140 feet deep, representing 100,000 tons of ore, but it is a possibility so strongly indicated by the characteristics of the more developed ore bodies in this property, that a disappointment may be looked for only remotely. With due regard to the limited development, we are justified, however, to assume here the existence of an ore body at least 50 feet by 100 feet in area, representing about 360 tons of ore to each foot in depth, or 50,000 for a depth of 140 feet, the present lowest ore exposure in the John's Mine. This block of ore would average 1.53 % copper, and

Block 4.--- The ore exposures in cross-cut III. and IV., about 50 feet long each, and a distance of 175 feet apart on the strike of the lode, showing their greatest values in their present N. W. face, warrant /under the same reservations as those stated in regard to Block 3 / the assumption that we have here an ore body of an area of at least 50 feet by 175 feet, representing 625 tons of ore for each foot in depth, or 87,000 tons for a depth of 140 feet.

An examination of sketch 1, showing the present development work at the John's Mine, shows that between the ore blocks indicated by the surface cross-cuts and the underground work, a large virgin territory remains, where further prospecting work will undoubtedly uncover additional ore bodies.

The contour of the ground permits of a tunnel being driven

into the mountain side at least 200 feet deeper than the present tunnel, which would tap the lode at about 300 feet below its apex. This tunnel would be nearly 1,000 feet long.

THE ERIK'S MINE.

At a distance of about 1,000 feet N.E. on the strike of the lode from the John's tunnel, is the adit T, which leads into the Erik's Mine. The surface above the tunnel is at an elevation of 1,160 feet above the sea, or 110 feet lower than the surface of the John's tunnel. The Erik's tunnel is about 80 feet below the surface. It is driven into the mountain side at practically right angles to the strike of the lode. A shaft /S./ has been sunk from the tunnel level to a depth of 70 feet. From the bottom of the shaft a cross-cut /C/ runs N.W. about 35 feet long.

The ore has been exposed in this mine, therefore, for a total depth of 150 feet from its apex, or 120 feet lower than the lowest ore exposure in the John's Mine.

The tunnel /T/ cuts the footwall of the lode at a distance of about 30 feet from its mouth. Strike and dip are here practically identical with those measured at the John's Mine, i.e. $N.42^{\circ} E.$, for the strike and $78^{\circ} N.W.$ from horizontal for the dip.

At a distance of 43 feet from the mouth of the tunnel, drifts have been driven N.E. and S.W. /D.N. and D.W. respectively/ about 70 feet long each. The N.E. drift /D.N./ passes for a distance of 50 feet through good ore, averaging 4.10 % copper. Ten feet in about the centre of this 50 feet stretch of ore return an average of 4.53 % copper, where the ore assays almost 5 % copper.

The remaining 20 feet of the N.E. drift /D.N./ to the face contain 1.69 % copper, the value decreasing gradually towards the face of the drift from 3 % to 0.9 copper, the latter at the face of the drift. The whole length of the drift /70 feet/ averages, therefore, 2.7 % copper.

The S.W. drift /D.W./ contains for a length of 30 feet from the tunnel 3. % copper, decreasing in the last 12 feet from 3 % to 1.4 % copper.

The following 25 feet S.W. average 0,69 % copper, decreasing gradually in value in this distance from 1,4 % to 0,5 % copper.

The average copper contents of the ore block exposed in the S.W. drift /D.W./ for an length of 55 feet, is therefore 1,95 % copper.

At the N. E. end of the drift P. W. it meets another tunnel /T1/ about 40 feet long, parallel to the first-mentioned tunnel /T/. From the drift /D.N./ this tunnel /T.1./ runs S.E. in ore ~~shank~~ containing 0,63 % copper, to the footwall of the lode, a distance of about 10 feet. About 20 feet S.W. from the face of the N.E. drift /D.N./ a cross-cut /C 1/ has been driven N.W. 40 feet long, entirely in ore averaging 1,67 % copper. An upraise /U/ has been raised from this cross-cut to the surface at a point about 15 feet from the junction with the N.E. drift /N.D./ This upraise is entirely in ore averaging 2,36 % copper.

This tunnel /T/ has been extended N.W. for a length of about 70 feet.

The samples taken here average 1,24 % copper.

In the S.W. drift /D.W./ at a point about 50 feet S.W. from the tunnel /T/ a cross-cut /C 2/ has been driven N.W. 30 feet long. The samples taken in this cross-cut average only 0,35 % copper.

At the end of the S.W. drift /E.W./, at a distance of 80 feet S.W. from the tunnel, a cross-cut /C 3/, 35 feet long, has been driven N.W., and another cross-cut /C 4/, 33 feet long, runs S.E. At the same point the drift sometimes, turning gently northerly, running almost due west for a further 66 feet. Both cross-cuts and the continuation northerly of the S.W. drift /D.W./ are in a block of poor mineralisation /P.1/ containing about 0,25 % copper.

SURFACE WORK AND ORE PROBABLY IN SIGHT.

By the underground development work at the Erika Mine, a block of ore has been opened up, exposed by two shafts from the surface to a depth of 150 feet, and by several drifts and cross-cuts. The block within the limits of the present ore exposure is 125 feet long, 85 feet thick and 150 feet deep — representing over 110.000 tons of ore containing 1,90 % copper.

At the surface, near this block of ore, S.W., a cross-cut /11./ has been driven about 125 feet long. For a distance of 50 feet this cross-cut contains 1,15 % copper, while 38 feet of this distance can be extracted assaying 1,9 % copper. The remainder of the cross-cut being poorly mineralised, contains from 0,1 % to 0,2 % copper only.

At a distance of about 350 feet S.W. from the Erik's tunnel /T/ and about 170 feet N.W. from its mouth, a cross-cut /1/, 60 feet long exposes good ore in about 35 feet of its length. The ore averages 1,77 % copper, the values commencing in about the centre of the cross-cut with 0,3 %, increasing going N.W. the last 20 feet, averaging 1,87 % copper, the richest ore being found in the N.W. face of the cross-cut, containing , 2,41 % . We find here again the characteristic indications of an ore body, the values increasing to a maximum towards centre and decreasing towards the limits of the ore body.

Under the same assumptions which were explained in the descriptions of similar ore exposures in the John's Mine, we may reasonably expect a considerably body of good ore here, but the limited development does not justify a definite expression as to ore in sight.

Compared with other parts of the property S.W. the surface work at the Erik's Mine is very limited, although the lode crops out boldly along the surface and looks very promising. Only the two cross-cut mentioned have been made here, leaving an unexplored space of over 300 feet between them.

N.E. of the Erik's Mine an attempt at exploration has been made at the Johannes Claim only, leaving a virgin strip of over 7.000 feet between this claim and the Erik's Mine. The lode crops out and can be easily followed over the whole distance.

THE OLLE'S MINE.

Tunnel.--- At a distance of 1.100 feet S. W. on the strike of the lode from the tunnel leading into the John's Mine, the Olle's tunnel has been driven into the mountain side. Five hundred feet of territory in this distance has not been prospected at all. The tunnel is about 600 feet long.

The elevation of the Olle's tunnel is 1.328 feet. It is about 150 feet below the outcrop of the lode, and about 50 feet higher than the John's tunnel and 170 feet higher than the Erik's tunnel.

The Olle's tunnel has been driven across the lode through poorly mineralised ground, cutting, however, at a distance of about 350 feet from its face, an ore body about 30 feet long, in the tunnel, containing 1,28 % copper. It is not worthy that immediately above the tunnel but some distance further S.W., an ore body of considerable extent is exposed /in surface cross-cuts IV and V/. of which the ore exposure in the tunnel represents undoubtedly the N.E. limit.

The object of the owners of the mine of driving the tunnel has evidently been to tap the large ore bodies exposed in cross-cuts IV and V. Running parallel to them, however, it will be necessary to run short drifts S.W. to obtain the same results. It is understood that this has been the original intention.

To prove the existence of the ore body at the depth of the tunnel, a drift was run S.W. below the ore body exposed on the surface by cross-cut V. At the conclusion of the writer's last visit to the property in November 1911, this drift had reached a length of about 5 feet. Samples taken at the face of the drift at that time contained a little over 1 % copper, justifying the assumption that the ore body exposed at the surface in cross-cut V. had been tapped. This drift is now being extended.

A Surface work and Cross-cut IV and V. -- Immediately above the ore exposure in the tunnel, but some distance S.W. from it, surface cross-cut IV., about 135 feet long, passes through ore containing 1,38 % copper, for a distance of 80 feet. In the immediate continuation of this cross-cut N.W., but at a distance of about 30 feet S.W., crosscut V., about 300 feet long, has been driven, exposing good ore for about 70 feet in its S.E. end. The ore exposure in these two cross-cuts belong evidently to the same body, and contain 1,38 % copper. N.W. beyond this ore exposure in cross-cut V., ~~another~~ after passing a zone of poor mineralisation for about 100 feet, another ore body is opened up, showing a width of 120 feet. The samples taken here average 2,06 % copper. This is the ore body referring to above as having probably been tapped by the drift now being

driven below it, S.W. from the tunnel level.

Cross-cut III and Cross-cut II.--- Parallel to the tunnel and to cross-cut V., and fully 200 feet N.W. from them, cross-cut III., 250 feet long, passes at its S.E. end a good ore body for a length of 60 feet. The samples taken here average 2.56 % copper. Beyond this ore body N.W., a poorly mineralised zone is passed, but in the immediate continuation S.E. of the ore body exposed in cross-cut III, the same ore body is opened up in cross-cut II. /100 feet long/ for a length of about 70 feet, averaging 1.6 % copper, although at its N.W. end it contains over 2 % copper, decreasing in value gradually southwesterly.

Cross-cut I.----- About 200 feet east from cross-cut III, an ore body is exposed in cross-cut I. /80 feet long/, for a length of about 20 feet, containing 1.52 % copper. This ore-body represents either the east limit of the ore-body exposed in cross-cut III. and II., or of another ore body further east, which has not been prospected so far. The territory between cross-cut I at Olle's Mine and the nearest opening east at the John's Mine, a distance on the strike of the lode of over 500 feet, has not been explored.

Ore in sight.----- The work at the Olle's Mine has undoubtedly exposed large and valuable ore bodies, the exploitation of which will result most profitably. It is to be regretted that the long tunnel should have been driven without disclosing at depth the character of the ore bodies exposed in cross-cut V.

The drift which is now being driven S.W. from the tunnel level appears, however, to have reached the N.E. boundary of the body exposed in cross-cut V, as intimated above. Further development is required to prove the correctness of this conclusion, although the probability may be assumed with justification. Assuming this ore body to extend to the tunnel level, over 150 feet below the surface, we would find here a block of ore, at least 100 feet thick, 200 feet long and 150 feet deep, representing 210,000 tons, containing /at the surface/ 2.06 % copper.

The tunnel has evidently been cut through the east limit of the ore occasionally, but penetrating a zone of poor mineralization.

body exposed in cross-cut IV.

Block P. ----- The ore exposure in the S.E. end of the cross-cut V., the N.W. end of the cross-cut IV. and in the tunnel, belong evidently to the same ore body.

The Extreme limits of its exposure are, therefore, on the line of the strike between cross-cut V and the tunnel level, 70 feet perpendicular to the strike from its N.W. exposure in cross-cut V., to its S.E. exposure on cross-cut IV., 150 feet, representing a block containing 112,000 tons, averaging 1.36 % copper.

Block P. 1.----- The ore opened up in cross-cut II and III belongs apparently to the same ore body. The extreme exposure are from its N.W. limit in cross-cut III. to its S.E. limit in cross-cut II., 130 feet, and along the strike direction of the lode from cross-cut III to cross-cut II., about 100 feet, representing an area of 13,000 square feet, or over continues at depth of the tunnel level, the present lowest ore exposure in this mine, this block would contain 140,000 tons of ore.

The ore exposure in cross-cut I is not sufficient of size to form the base of speculative calculation, beyond the reference made to it above in the description of the work performed.

Large areas within the present development work at the Olle's Mine have not been prospected. Further ore bodies will undoubtedly be found here, if the lode is thoroughly explored.

TRIANGELN AND SEGMENTED MINES.

Tunnel and cross-cut II.--- About 600 feet N.W. from the Olle's tunnel the Triangeln tunnel has been driven into the mountain side. The tunnel is only about 20 feet long, at an elevation of 1,350 feet above the sea, and about 100 feet below the outcrop of the lode. It is immediately below a surface cross-cut II., 300 feet long. This cross-cut passes through ore the whole distance, averaging 1.7 % copper.

Segmented Shaft.---Near its N.W. end an inclined shaft the Segmented, has been sunk to a depth of about 50 feet, passing ore occasionally, but penetrating a zone of poor mineralization.

Cross-cut I.--- About 170 feet N.E. from Cross-cut II, a similar cross-cut I has been made, paralleling the former, about 500 feet long. Excepting a stretch of about 20 feet near the middle

of the cut, where the samples returned 1,4% copper, the cross-cut passes through poorer mineralised ground, increasing gradually in value to about 100 feet from the N.W. face of the cross-cut where the mineralisation has reached 1% copper, and continues to the N.W. end of the cross-cut, averaging 1,1% copper.

Cross-cut III and IV.----- Immediately S.W. of the N.W. end of the cross-cut II, at a distance of 100 feet, two cross-cuts III and IV have been driven adjoining each other so closely that they may be regarded as one continuous cross-cut. The combined length of these cross-cuts is 100 feet. They have been driven entirely in ore, assaying 1,67% copper.

Cross-Cut V.---- About 150 feet S.E. from cross-cuts III and IV., and about 150 feet S.W. from cross-cut II., ore is exposed again in cross-cut V., about 35 feet long, averaging 1,35% copper.

Cross-cut VI.---At a distance of about 250 feet S.W. from cross-cut II., parallel in the latter, cross-cut VI, about 200 feet long, passes through ore in the S.E. half of the cut, averaging 2,06% copper for a length of 100 feet, while about 50 feet in the N.W. half, direct continuation, return 1,4% copper. The whole cross-cut contains over 1% copper, while 150 feet of it average 1,8% copper.

To prove the continuation of the ore body as exposed on the surface, a bore hole was driven into the mountain side by a diamond drill, pointing N 40° W. at an angle of 10°. At a distance of 10 feet the ore was encountered, containing 0,5% copper. As the hole proceeded, the grade of the ore increased until at a depth of 130 feet the core assayed 3,12% copper, decreasing thence in value in the next 65 feet, where the core assayed 1,4% copper. Beyond this point a zone of poorer mineralisation was reached, and after reaching a depth of 400 feet, having passed an additional ore body 65 feet thick, containing 1,6% copper, it was discontinued, although the limit of the mineralbearing zone exposing ore bodies at the surface had not been reached.

This bore-hole has exposed two ore-bodies, 120 and 65 feet

thick respectively, averaging over 1 % copper or /leaving out the
lower ores near the orifice of the hole / 80 feet and 35 feet thick
respectively, containing 1,64 % copper.

Ore in sight.----- The openings made suggest that we are
here in the presence of an immense ore body, bounded by cross-cut
II at the Triangeln and cross-cuts I. and III. at the Kvadraten
on the line of the strike of the lode, 400 feet, and by cross-cuts
III. and IV., and the Triangeln tunnel, 300 feet, perpendicular to
the strike, representing an area of 120.000 square feet, which would
furnish for each foot in depth about 8.500 tons of ore.

Assuming, then, that the ore continues at depth to at
least the tunnel level, 100 feet, without change, the block would
furnish 850.000 tons of ore containing 1,6 % copper.

Cross-cut I ----- In the absence of further development
throwing light on the extent of the ore body exposed in cross-cut I.,
a statement as to the probable amount of ore represented by it would
not be justified. Having in view the characteristics of the more fully
developed ore bodies on the property, the comparatively low minerali-
sation of the ore exposure in cross-cut I would suggest that it forms
either the N.E. limit of the big ore body /R/ described above, or that
that it represents the S.W. limit of a new ore body.

KVADRATEN MINE.

Tunnel.----- About 450 feet S.W. from the Triangeln
tunnel the Kvadraten tunnel exposes, at a depth of about 100 feet
below the outcrop of the lode. The tunnel is 230 feet long, at an ele-
vation of 1.335 feet. It is, therefore, 250 feet higher than the lower
ore exposure in the Erik's Mine. The mineralisation commences at the
mouth of the tunnel, and increases until it reaches about 1 %
copper, at a distance of about 30 feet continuing for about 120
feet, where it passes into poorly mineralised ground again. The sam-
ples taken of the ore body return 1,33 % copper.

Assuming that SURFACE WORK. continues at depth without change to
the tunnel level, where it is also exposed, this block would yield

Cross-cut I and II.----- immediately N.E. from the
tunnel and on a line with the ore body exposed therein, surface cross-
cuts I and II have been driven, 75 and 120 feet long respectively

Cross-cut I passes ore /over 1% copper/ in its entire length, while the last 50 feet N.W. contain 1,30 % copper, some samples near the N.W. face returning 2,52 % copper. Cross-cut II. practically a direct continuation of cross-cut I., runs entirely in ore averaging 1,6 % copper.

Cross-cut III. ----- At a distance of 140 feet S.W. from the tunnel, cross-cut III. has been driven, about 40 feet long. Although the samples taken here returned about 1 % for the whole length of the cross-cut, only about 20 feet in the centre averaging 1,25 % copper.

Cross-cut IV. -----About 80 feet S.W. from cross-cut II. cross-cut IV., about 40 feet long, traverses ore at its S.E. half for length of about 20 feet. This ore contains 1,10 % copper, although the average of the whole cross-cut yields about 0,9 %.

Cross-cuts V and VI.-----These cuts, lying in a depression of the ground, could not be sampled at the time of the writer's visit being filled with water, which, on account of snow covering the depression, could not be conveniently cleared. Samples taken here on a former occasion by two well-known engineers, returned 1,40 % copper for about 30 feet at the S.E. end of the cut V., and 1,52 % copper for about 40 feet at the S.E. end of the cut VI.

Crosscuts VII and VIII.-----about 100 feet S.W. from cross-cut VI., cut VII and VIII., about 100 feet long, have been driven practically a direct continuation. The samples taken here returned 1,23 % copper for cut VIII, and 1,66 % for cut VII, averaging 1,1% copper for the whole distance of 100 feet.

Ore in Sight.----- The Kvadraten tunnel and cross-cuts I and II have opened an ore body, which between their present limits measures about 60 feet, along the strike of the lode, and 200 feet at right angles to it, an area of 12.000 square feet, representing about 850 tons of ore to each foot in depth.

Assuming that this ore body continues at depth without change to the tunnel level, where it is also exposed, this block would yield the last 50 feet of the S.W. face return 1,64 % copper.

85.000 tons of ore, averaging 1.44 % copper.

Cross-cuts III to VIII.----- The ore exposures in cross-cuts III to VIII are not of sufficient size to justify a speculation as to an amount of ore they might represent to the depth of the tunnel level. Bearing in mind the characteristics of the more fully developed ore occurrences in this lode, in conjunction with fact that the ore exposures in cuts III to VIII are at the S.E. end of the cuts, and that the mineralisation is comparatively poor where exposed in cuts IV. and VII, situated furthest N.W. and richer in cuts V. and VI, situated further S.E., it may justly be conjectured that further developments S.E. and S.W. will demonstrate the existence of one or more large and well-mineralised ore bodies.

VESTRA FALTET EAST.

Tunnel. ----- Between the last south-westerly surface cross-cut of the Kvadraten Mine and the first North-easterly cross-cut on the Vestra Faltet East, is a virgin territory of over 1.000 square feet. The lode crops out boldly and promising in this space, but no prospecting work of any description has been done here.

An attempt has been made to tap the ore bodies at depth at the Vestra Faltet East Mine by the Heyermann's tunnel. This tunnel has been driven into the mountain slope, at a distance S.E. of about 500 feet from the surface ore exposures. The tunnel is said to have reached a length of 60 feet, and to have passed through good ore. It could not be inspected, its mouth being so inconveniently situated that it was covered with snow and debris at the time of the writer's visit, in such a manner that the task to uncover it would have been most laborious and difficult.

The tunnel mouth is at an elevation of 1.400 feet, and would tap, if extended, the ore bodies exposed in the surface cross-cuts at a depth of about 150 feet.

SURFACE.

Cross-cut I.----- The furthestmost N.E. ore exposure of this time is in cross-cut I, 170 feet long, where for a length of about 50 feet at its N.E. end it averages 1.33 % copper, while the last 30 feet the N.W. face return 1.64 % copper.

Near the N.W. end of this cross-cut, in the section which ret 180
1,64 % copper, a shaft has been sunk about 30 feet deep entirely
in ore.

Cross-cut II.----- At a distance of about 180 feet S.W. from
cross-cut I, a surface cut II, 120 feet long, has evidently been
driven in a zone of poor mineralisation. The samples taken here
average about 0,15 % copper.

Cross-cut III.-----About 130 feet further S.W., cross-cut III,
about 100 feet long, exposes ore throughout. The samples taken here
average 1,55 % copper, although for some distance from 2,5 to
3,2 % copper were recorded.

In the section near the N.W. end of this out, which returned
2,5 % copper, a shaft has also been sunk to a depth of about 30 feet.

Cross-cut IV.----- With the exception of cut A in this group
which contained 1,5 % copper, the small cuts in this group are poor-
ly mineralised, averaging only 0,80 % copper.

Ore in Sight.----- The only ore exposures which justify a
speculation on a certain amount of ore they might represent to the
depth of the tunnel level, are those in cuts III and IV A, which
probably belong to the same ore body. In this case there would be an
ore body here between the limits of the present ore exposures in the
two cuts, 160 feet long, on the strike of the lode, and of a thickness
equal to one half of the total length of the exposure in the cuts
perpendicular to the strike, 60 feet representing an area of 9.600
square feet, which will yield 680 tons of ore to each foot in
depth. Assuming that this ore continues without change in its charac-
teristics to the tunnel level, this block will contain about 100.000
tons of ore, containing 1,55 % copper.

A block of poor mineralisation adjoins this block, again
N.E. and S.W., exposed by cuts II and the four south-easterly cuts
of the group IV respectively.

N.E. from cut II a good block of ore is evidently
exposed again in the N.W. half of the cut I. But this being the only
exposure, a suggestion as to any amount of ore it might represent would

be hazardous.

In the light of the above mentioned and repeatedly referred to characteristics of the ore occurrences in this lode, as indicated by the more fully developed ore bodies, it can not be doubted that cut IV A and cut III are in the S.W. part of an ore-body, which will continue N.W. and that the ore exposure in cut I is also leading into a well mineralised ore body N.W. In cut III, as well as in cut I, the value of the ore increases gradually towards the N.W. reaching the highest value at their N.W. face.

VESTRA FALTET MINE WEST /Sketch VII/.

Cross-cut I.----- About 120 feet S.W. from the group of the cuts IV at the Vestra Faltet East Mine, we find the first ore exposure on the Vestra Faltet West Mine in the surface cross-cut I, about 220 feet long. The ore in this cut is poorly mineralised, containing about 0,5 % copper, with the exception of about 20 feet in about the centre, which returned 1,85 % copper.

Cross-cut II and III.--- Cuts II and III, almost due N.W. from cut I, with a combined length of about 80 feet, are in a poorly mineralised zone, containing 0,3 % copper.

Cross-cut IV.----- Cut IV, about 20 feet long, contains only 0,6 % copper.

Cross-cut V.----- Cut V, about 100 feet long, is also in a zone of poor mineralisation at its S.E. half, but contains 1,4 % copper in the last 40 feet at its N. W. end.

Cross-cut VI.----- Of the group of cuts, VI only, the most north-westerly cut, shows above 1 % copper, returning 1,15 %. The remaining three contain from 0,25 % to 0,6 % copper.

Ore in Sight.----- The development work at this mine is not sufficiently advanced to warrant any suggestion as to a possible amount of ore indicated by the ore exposures. It is probable that the ore exposures in cut V and the most north-westerly cut of group VI belong to the same ore body. Only further development can demonstrate this.

/Signed/ Otto Stalman.

London, January 12 th, 1913.

Norges Geologiske Undersøkelse

Josefinegt. 34 - Oslo

E. E. Ho

Oslo, den 17. oktober 1955.

Norges Geologiske Undersøkelse

Bergarkivet.

Geolog Peter Padget

Norges geologiske undersøkelse

O s l o.

Bestemmelse av Cu i 5 prøver fra Ulveryggen, Repparfjord.

Prøve merket:

Cu

554, Trench VI0,76 %

555, " XI (1).....1,12 "

556, " XI (2).....0,49 "

557, " XI (3).....0,91 "

558, " XI (4).....0,45 "

Norges Geologiske Undersøkelse

Bergarkivet

Rapport nr.: 4430

Bestemmelsen er utført av Erna Christensen.

Erna Christensen

Norges Geologiske Undersøkelse

Josefines gate 34
Sentralbord 59 57 80

EC/EHO

Geolog Peter Padget,
Norges geologiske undersøkelse.
O s l o.

Bestemmelse av Cu i prøve merket Ulveryggen, Repparfjord.
Trench XI.

2,87 % Cu

Bestemmelsen er utført av Anna Christensen.

Brorjolf Brunn

Oslo, 3/11.55

Norges Geologiske Undersøkelse

Bergarkivet.

Signert for en del
lukkert post

Norges Geologiske Undersøkelse

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EC/EHo

Geolog Peter Padget,
Norges geologiske undersøkelse.

O s l o .

Bestemmelse av Cu i prøve fra Repparfjord, Trench XI,
1, 2 , 3, 4. Nr. 582.

0,73 % Cu.

Bestemmelsen er utført av Erna Christensen.

Erna Christensen

Oslo, den 14. novbr. 1955

Norges Geologiske Undersøkelse

Borgarkivet.

Norges Geologiske Undersøkelse

Josefines gate 34
Sentralbord 69 57 80
EC/EHo.

Oslo, den 20. oktober 1955

Geolog Peter Padget,
Norges geologiske undersøkelse.
O S L O.

Norges Geologiske Undersøkelse
Bergarkivet.

Bestemmelse av Cu i 3 prøver fra Lepparfjord.

Prøve merket	Cu
560 Trench I (1)	<0,03%
561 " (2)	<0,03%
562 " XI kopperrik sone..	1,41%

Bestemmelsene er utført av Erna Christensen.

Erna Christensen

Utklag av rapport av C. W. Carstén

Norges Geologiske Undersøkelser

Bergarkivet.

REPPARFJORD KOBBERMALMFÖREKOMSTER (i Finnmark).

.... Forekomstene optræder i en eiendommelig sparagmitisk bergart vest for Repparfjordbotn på nordsiden av Ariselven i høider 3-400 m.o.h.

Den optrædende sparagmitskifer, som tilhører de såkalte Finnmarksformasjoner, fører overveiende kvarts, plagioklas, klorit og muscovit. Biotit, pyroxen og enkelte andre mineraler inngår leilighetsvis i mindre mengder. I veksellagring med denne bergart optræder hist og her mindre mektige fyllitbenker. En rekke mikroskopiske preparater synes å antyde at den gjennomsnittlige SiO_2 -gehalt i almindelighet ikke ligger stort over 75%. (Kvartsgehalten er i almindelighet 60-65%). Samtidig inngår forholdsvis betydelige kvanta alkalier.

På Ulveryggen, hvor hovedforekomsten finnes, er den forhåndenværende sparagmitskifer, som her optræder med vesentlig NE-lig strøkretning og steilt fall, over en lengde av 1.5-2 km og over en bredde av op til over 100 m (tildels litt forskjøvet i forhold til strøkets retning) mere eller mindre sterkt impregneret med kobberglans og brøgetkobber. Et meget stort antall røskegrøfter (i alt ca 50) er her opskutt lodrett strøkretningen. Samtidig er der fra søndre dalside ansatt flere stoller i forbindelse med forskjellige undersøkelsesdrifter, orter og synker. Samtlige stollmunnar lå ved mitt besøk 13.juli helt tildekket av sne og gjorde stollene helt utilgjengelige.

En gjennomsnittsprøve fra Ulveryggens malmsone, representerende stoffprøver og uttatt i 1929 viser følgende kjemiske sammensetning:

2.16 % Cu
81.83 % SiO_2
8.39 % Fe_2O_3
6.97 % Al_2O_3
0.60 % MgO
0.28 % CaO

Norges Geologiske Undersøkelser

Bergarkivet

Rapport nr.: 4431

Da stufprøver ved forekomster av ovennevnte type kun gir et lite pålidelig holdepunkt for verdesettelsen av den forhåndenværende gang, bør ovenstående tall korrigeres ganske betraktlig. Man kommer da sandsynligvis til en kobbergehalt av gjennomsnittlig 1 & 2 %, kanskje tildels derunder, samtidig som SiO₂-gehalten likeledes vil synke ganske vesentlig.

Innenfor malmsonen viser bergarten på en rekke steder på sletter og sprekker malakitbelegg (i almindelighet regnet som indikator på impregnasjon av kobberglans og broget-kobber). I en rekke av de av de opsprengte røsker er imidlertid malakitgehalten sterkt vekslende, hvilket samtidig antyder en meget varierende kobber-gehalt. I større partier av malmsonen kan således kobbermineraler ikke påvises med det blotte øie eller luppe.

Forekomstenes utstrekning og utholdenhet mot dypet er forholdsvis ukjent, da samtlige drifter under dagen ligger temmelig nær dagoverflaten. De optredende kobbersulfidens stabilitet mot større dyp er imidlertid i de fleste tilfeller temmelig tvilsom.

Utenfor Ulveryggen optrær sulfidimpregnerte sparagmitssulfider av samme type på flere steder uten at disse dog synes å spille nogen betydelig rolle.

.....

N.D. 797-1333 I.

F. B. ⁶⁶₃₃

N.D. 1359-1333 I.

Ripperfjord Copper Mines

Report

by

Professor Johan H. L. Vogt.

24/8/07.

Norges Geologiske Undersøkelse

Bergarkivet

Rapport nr. 4477

Translation.

The Ripperfjord Copper Ore Field.

at Ripperfjord in Kvalsund, south of Hammerfest, Finnmarken Amt, Norway

Regarding location, transportation and communications the pamphlet issued by "Nordiska Grufaktiebolaget"/Ripperfjord Mines/ Östersund, Sweden, gives ample information. ~~Here will only be compared.~~

The copper ore field is located about 40 km. S.E. of Hammerfest. The "sparagmite" ore deposits lie generally at a height of about 250/300 or 350 m. above the sea. Close by Ripperfjord, near the foreman's dwelling house, is good space for the erection of Works. A projected ropeway from the Eric's mine to this place will be 1½ km. long and from the Johns mine and the Olles mine the distance is in a straight line about 1650 and 1800 m. respectively.

There is in the neighbourhood of the place for the proposed works available waterpower /by damming &c./.

The mining at the main deposits /the "sparagmite ore"/ can for a long time be carried on by adit driving.

Details as regards altitude, distance, water fall district &c. can be had from a map which the Company at present is preparing in scale 1-4000, but this map, however, was not completed at the time of my visit.

Geological Review.- Near the shore of the inner part of Ripperfjord and from here to Ulveryggen with surroundings and quite a distance to S.E. appears immense beds of quartz sandstone with conglomerate and "sparagmite".

At the Brathammeren mine and "Skiffergangen" /slate vein/ appears a green, at times very slaty, mineral which most likely is a transformed argillaceous schist, about which I will, however, not express my positive opinion. These beds belong to the formation which T. Dahl at the time called "Raipas" and are now supposed by many to be of devonian age. At other places in the contiguous district for instance at Kvalsund, in the neighbourhood of Næverfjord and near Altenfjord at Porsa /N.B. not at the mines/ appears very large deposits of dolomite.

At a distance of, to judge by the eye, a little more than 1 km. S.E. from Ulveryggen occurs a field of Serpentine.

According to what I know from earlier examinations there is

on the peninsula between Fipperrfjord and the Ripperrfjord valley in the East and Altenfjord in the West; also large fields of gabbro and granite.

The quartz sandstone with appertaining layers of conglomerate and "sparagmite" is a hard, compact, somewhat transformed rock, which as a whole is of rather a monotonous character. It is as a rule comparatively small grained /i.e. with small grains of sand/. At several places are, however, layers of conglomerate or conglomer-breccia, seen, with only small edge worn stones, at Ulveryggen often of a size of hazelnut /seldom larger/, and of sparagmite /i.e. layers with both grains of quartz and feldspar/. The ore at Ulveryggen is in everyday-vocabulary called "sparagmite" ore. This name has most likely its source from the fact that I, during my visit in 1903, saw a small working of ore in sparagmite, and for this reason I did at that time call the ore "sparagmite" ore, but "sparagmite" is not the dominating rock here. The ore could just as well be called sandstone ore. At Ulveryggen and the nearest surroundings the strike is very regular, about N.E. to S.W. nearest 35 or 40° - $S35$ or 40° and the dip is at Ulveryggen steep towards N.W. I measured the dip at Erika mine and Johna mine to about 75° , at Olles Mine to $80-85^{\circ}$. The small variations in the observations here I do not find worthy of any consideration. The result is that in the group of Erika mine - Johna mine - Olles Mine and the cuts S.W. of Olles Mine, strike and dip are very regular. At other places, consequently in profile, near the Ripperrfjord, are on the other hand seen quite strong foldings with dip in various directions.

The width of the quartz sandstone section with appurtenant layers of conglomerate and sparagmite is quite large, at least 250 m. probably considerably more. The time did not allow me /topographical basis also being wanting/ to make up a geographical profile, which, however, at the present time would not be of any technical interest.

The boundary between the sand stone and the green, often schisty rock at "Skiffjergangen" and the Brathammiren mine, runs about 50 meters North West of the Brathammiren mine.

Geological comments re. the copper ore deposits.-

On the peninsula between the Rippertjord, Rippertjord valley in the East, Altenfjord in the West and Kvalsund in the North there are a whole series of copper ore deposits of various geological /or morphological/ characters. The occurrence of the ore is to an essential degree dependent upon the mechanical character of the surrounding rock.

There are accordingly,-

a) Typical ore veins, namely calcareous spar - quartz - veins with copper pyrites and sulphur pyrites. For instance, the Brathammeren mine near Rippertjord, amongst others the Forsa veins.)

b) So called schist veins, namely impregnations /infiltrations/ in slaty rock or slaty zones within certain rocks.

c) So called sparagmite ore in sandstone - conglomerate-sparagmite rock species.

It is easy to decide that the copper ore everywhere here is in a geological /genetic/ respect of a younger character and that it neither in the schist ore /skiffermalin/ nor in the "sparagmite" ore was formed simultaneously with the rock in question. Especially as regards the "sparagmite" ore one finds fine streaks of ore veins intersecting each other all over in the rock indicating that the ore is infiltrated through the rock after its formation.

Geologically analogous ore deposits infiltrated or infiltrated in sand stone are found at various places abroad. Larger deposits of this character are on the contrary, as far as I can remember, not to be found in Norway or Sweden. In the sand stone, which originally must have been a more or less porous rock, the infiltrations quite often distribute themselves in the great number of pores through the whole rock. I want to distinctly point out that in dealing with the sparagmite ore there is not dealt with primary sedimentation of copper ore but with a secondary infiltration. This is richest in certain lenticular limited parts, as far as known by the direction in field of about the same strike and dip as the country rock, but it is not to be understood so that there are certain levels or sections of the country rock with the same copper percentage.

The impregnation in the sand stone and so forth at Ulverysen can be followed for a length /mainly but not accurately reckoned along the strike of the country rock/ of 1 km. or more and there is occasionally -when the poor ore is taken in- impregnation of a width /vertically on the strike/ of about 100 m. At the small lakes and about S.W. of the cuttings in the continuation of Ulverysen near Olles Mine there is also to be found impregnation of sparagmite ore at quite a great distance from the dominating strike at the Eriks, Johns and Olles Mines /Ulverysen/.

X We are consequently here dealing with an impregnation of sparagmite ore of large dimensions as to length and width. (This impregnation took place, geologically speaking, at a very early period, when the present surface was several kilometers lower than the surface at that time. At that time overlying masses of rock of several kilometers thickness are now eroded away. (When the impregnation of sparagmite ore has such considerable dimensions in length and width and as the impregnation took place at a depth, independently of the now existing surface, then the impregnation cannot cease near under the present surface but will continue to greater depths.

The particularly rich impregnation lenses which are found close to the surface, for instance in the adits, will probably disappear locally or wedge themselves out at a depth, but the impregnation, as such, will continue with changing sections of rich and poor impregnations. A horizontal sections, for instance 100 meters or 200 meters below the present surface at Ulverysen will thus have the same general character as the present surface /or level for instance at the Eriks adit/. At the Olles adit ore has been proved at a depth of 60 meter below the surface and one must therefore come to the conclusion that the ore continues at a depth. It would of course be well to have actual proof of this, but personally I attach great importance to the general conclusion arrived at through examination of the occurrences and general geology of same.

The Sparagmite Ore.- It is evident both through examination of the Mines and study of the assays that the immense sand stone deposits at Ulverysen /principal deposit, are of a width

measured at a 100 m. or more, all over, or nearly all over, impregnated with ore, to wit, Bornite which occasionally is accompanied with some chalcocite and copper pyrite.

The impregnation is often poor, but there are richer local impregnations and it is of course, only those richer impregnations that have a technicoeconomic interest. In order to give an idea about the poorer impregnations there are copied here some assays of drill-core from the diamond drill holes at Eric's mine, the drill-hole was 80 m. in depth and the drill-core from every fourth m. to wit for 3- to 4 m. 7 - 8m. and so forth to 70-80m. / was assayed. The assays given in succession, /for 3 - 4m. 7 - 8m. 11 - 12 m. etc/ showed, 0.09, 0.35. 0.11. 0.02. 0.85. 0.81. 0.94. 0.32 0.13 0.13
0.20 0.10. 0.10. 0.13. 0.18. 0.31. 0.08. 0.16 0.10 0.28

of copper. Further is a serie of assays copied from Klems drift, also in Erik's mine which drift essentially runs through poor impregnations
0.10 0.12 0.10 0.08 0.27 0.09 0.43 0.19 0.13 0.10.
0.23 0.19 0.14 0.18 0.18 0.52 0.33 0.48 0.12 0.28.
0.95 0.89 0.32 1.37 1.68 0.13 0.58 0.23 0.14 $\frac{1}{2}$ copper.

All these assays arranged according to copper percentage.

0.02	0.08	0.08	0.09	0.09	$\frac{1}{2}$ copper				
0.10	0.10	0.10	0.10	0.10	0.10	0.11	0.12	0.12	0.12
0.13	0.13	0.13	0.13	0.14	0.14	0.16	0.18	0.18	0.18 xxxx
0.19	0.19	$\frac{1}{2}$ copper							
0.20.	0.23	0.25	0.27	0.28	$\frac{1}{2}$				
0.31	0.32	0.32	0.33	0.33	0.38 $\frac{1}{2}$				
0.43	0.48								
0.52	0.58								
0.81	0.85	0.94	0.95	0.99	$\frac{1}{2}$				
1.37	1.68	$\frac{1}{2}$							

These figures seem to show that the copper-percentage in the poor impregnation- which as a matter of course, will practically not be mined- only exceptionally sinks below 0.10 $\frac{1}{2}$ copper, the poor impregnation seems ~~xxxx~~ oftener to hold a copper percentage between 0.10 & 0.50 $\frac{1}{2}$ with an average of about 0.2 $\frac{1}{2}$ in copper. That is to say, that if in working it should happen that some of the poor impregnation

should be mixed with the goods going into the Concentrator, this poor impregnation will then, however, pay its own concentration expenses. The economical interests will, as previously stated, be concentrated on the richer impregnation sections. These appear lentiform within the poor impregnations. There are sections of so large dimensions that they can be worked separately by themselves and which hold over 3% in copper. /average sample for several meters length/ and have at Erika and John's mine shown 3.10 3.20, 3.24, 3.25, 3.45, 3.60, 3.73, 4.10, 4.13, 4.17% in copper. Outside of these rich sections come zones with lower copper percentage. 2-3% and thereafter 1-2% and lower.

The working should be arranged according to the following plan,-

Through preparatory work prove the richest sections where the richest, central parts, with an average of 3% copper or more, can be worked by themselves. Also the fairly rich impregnations can be worked by themselves. There is thus every opportunity to mine ore of different percentage of copper. How low the copper percentage should be allowed to run is of course dependent upon concentration expenses &c. I believe it would pay to handle ore as low as 1% in copper, or possibly a little below. The limit, however, cannot be decided as yet.

The apatite ore has been exploited at 3 mines, partly through adits and partly through cross-cuts /in John's mine also by a shaft/ in Ulveryggen, namely Erik's, John's and Olle's mines. /See map/. The distances between these mines are -

Between Erik's and John's mine	336 m.
" John's and Olle's mine	330 m.

The adit level in John's mine is about 30 m. and in Olle's mine about 20 m. above the level of the adit in Erik's mine.

The 3 mines are situated after each other, nearly, but not exactly on the strike. In each of these mines, besides the poor impregnation, are certain streaks of richer impregnation, between 1 & 3% copper. Hereof can any one easily be convinced by going into

into the mine and take samples at various places.

As the work with the drifts proceeded there were taken a series of average samples, those are already assayed. A few months ago there were in addition to these very comprehensive series of average samples taken, representing nearly every meter in the drifts. These average samples are not as yet assayed, though, when these samples are completed and laid in on the map, it will give a very good knowledge of the extension of the rich parts in the three mines "Eriks' John's & Olle's". When the assays are completed and delivered to me I will give a more specified report. I want in the meantime only to say that in Ulværyggen there are several deposits worth working, holding over 3% in copper and also other large deposits carrying between 1 & 2% in copper which would pay to work. A calculation has been laid before me showing that in the "asperagite-field" ore areas have already been ascertained,

with at least 3% copper	295 m ²
" " " 2% "	3130 "
" " " 1% "	1585 "

By studying the mine maps, the up to this time made assays, and by walking round in the mines, I got the decided opinion that the above figures were not too high. But as in the near future detailed assays will be had, I wish to express myself upon the ore area and its productiveness.

Besides the mines there are a whole suit of excavations made across the strike of the deposits further S.W. in Ulværyggen. Hereby it is ascertained that the deposits are at least 1. Km in length. Erik's mine is situated furthest towards N.E. in Ulværyggen & seems not to belong to the richest parts of the deposits. In the more central parts, at John's & Olle's mines and at the cuts S.W. of Olle's mine appears the copper-percentage, as an average, to be higher than at Erik's mine.

It is here dealt with a large ore-field of which I have the best hope.

The ore can be delivered cheap.

The mining-work will certainly be burdened with a considerable length of drifts but the mining of the ore will become cheap. To what

price a ton of ore should be figured at will principally depend on how low a copper-grade one can go. if one can go, as I believe clear down to 1% copper or possibly still lower will the mining expenses come very low. I wish, however, to postpone expressing an opinion about the estimates until the assays are ready, as they will give better information about the ore areas and mining conditions.

Preliminary remarks about the treatment of the ore. The mining work can be so arranged, that the richest goods with an average of 3% copper /or probably a little higher/ and the poorer goods, with an average of from 3% copper ^{and} down to 1% in copper / & probably a little below/ are kept separately. I count the rich goods to an average of 1 1/2% copper. There will be undoubtedly more of the poor goods than of the rich. I will first refer to the different hither to obtained estimates and tests performed.

CHLOROUS ROASTING & EXTRACTION. A test performed at Falun, Sweden, under the management of K.A. Ikerblom /for many years superintendent of Faluⁿ extraction works/ showed.

The ingoing ore held	2.17% copper.
• waste held	0.17% •
• loss in the waste was consequently 8.00%	

With a supposed yearly production of 100,000 tons of ingoing goods K. A. Ikerblom has calculated the construction-expenses for extraction plant at 1. Million Norwegian Kroner. and the running expenses / including amortisation, engineer wages etc. etc./ at 10 Kr. p. ton

With a yearly production of 10,000 tons ingoing goods Ikerblom figured the construction expenses at 17 Kr. per ton. /including amortization etc./ The supposition for this estimate is that the copper is delivered ready as refined copper. An eventual small silver percentage without any economical consequence worth mentioning, is not taken into consideration in the income-calculations.

THE ELLMORE PROCESS. A test /proportionately in a small scale/ with sparagite-ore showed.

The ingoing ore held	2.47% copper
• waste held.	0.40 % •
• Concentrate	23.56% •

This concentrate will later have to be treated for the production of metallic copper. Another test, not with the sparagmite ore, but with slate ore showed,-

The ingoing ore held,	0.89% copper
The waste held maximum,	0.19% "
The concentrate held,	17.29% "

The loss by the Ellmore process can in accordance with these tests be put at 15-20% of the copper percentage in the crude ore. To this core the loss in the treatment of the concentrate for producing metallic copper. The total loss of metal can probably be put at 25% - $\frac{1}{2}$ of the copper contents in the ingoing goods and possibly one may have to calculate the loss a little higher still. An estimate on a yearly production of 100,000 tons of raw ore gives /without considering mining/ a construction cost of Kr. 325,000, and running expenses per ton at most Kr. 4.50. To this will of course have to be added expenses for treatment of the concentrate into metallic copper. If a smelting plant be built at the place then I estimate this expense at about Kr. 150 per ton of copper /the expense is otherwise to a great extent dependent upon the percentage of copper in the concentrate. If in order to reduce the loss of copper by the Ellmore process one is satisfied with a concentrate with a lower copper percentage the expenses in later treatment will be considerably higher. For ore with 6-7% copper the smelting expenses, under conditions as at Rippertjord, may be estimated at Kr. 300.- per ton copper. Experiments with Siemens-Halske electrolytic copper process are now going on, but any results are as yet not known.

The chloroous roasting with subsequent extraction /which we will call the Falun process/ works, according to experience with a low loss of copper /about 8-10% of the ingoing goods/, but the expense per ton is high. The estimate is, - a yearly production of 100,000 tons crude ore Kr. 10.-, and a yearly production of 10,000 tons crude ore Kr. 17.- per ton. The cost by the Siemens-Halske process will probably also come up to these figures. As the sparagmite ore principally contains quartz /silica/ and as the

copper percentage is comparatively low then the smelting of the ore to copper matte is out of the question. It is useless to make any trials in this respect.

I consider it impossible to utilise a metallurgical process, which always will be comparatively expensive calculated per ton ingoing goods, on such a low grade ore /although cheap in mining/ as the sparagmite ore. The bulk of the ore will hold a comparatively low percentage, below 2% probably nearer 1%, and such an ore requires for preparatory treatment a cheap mechanical process for preliminary production of concentrate. The cost as regards the old concentration methods is a couple of kroner per ingoing goods, but gives with such finely impregnated ore as the one in question a very considerable loss in the waste. The oil processes are somewhat dearer /I have seen an estimate at Kr. 4½ per ton crude ore/, but they have the advantage that the work with a less loss of metal. I have confidence in the oil process and this would be the most rational at Ripperrigsd. There are at present several oil processes, but which of these should be selected it is difficult for me to say as I have not had any experience in the different methods. This question should be carefully investigated. It may be that it also would be the most rational to use an oil process for the rich ore, with 3% copper and higher. However, I consider it more probable that it will pay best, so as to avoid the not so inconsiderable loss connected with the oil process, to utilise a metallurgical process upon the rich ore, although this is more expensive per ton crude ore. There may be a question of the Falun process, the working of which is well known, or Siemens-Halskes electrolytic process, which up to the present time has been little used. There can be no question of direct smelting of the sparagmite ore. So as to thoroughly go into this question trials should be made with,-

Oil process with crude ore containing about 1½% copper
Falun process /or Siemens-Halskes process/ with crude ore containing 3% copper.

Further tests with the Falun process is hardly necessary.

The Brathammaren mine and "Skilfjergangen".- The Brathammaren mine is situated about 5-6 kilometers from the fjord by the road

or about 6-7 km. from the proposed smelting works. The vein cuts through the stratification of the country rock. This is nearly vertical, about 85 deg. towards S.E., whilst the vein has a dip of about 30 deg. towards S.E. The vein has been exploited for a length of about 50-60 meters and about 40 meters along the dip. The thickness of the vein is in the middle parts 0.6 - 0.8 m. but gets narrower towards the ends /the horizontal line/ and is here 0.1 - 0.2 m. The vein is a felspar - quartz vein carrying some copper pyrites. Ore has been picked containing 7% in copper but the vein is comparatively short in length /strike/, not very wide and does not carry much copper ore. The adjoining lode, the so called "Skiffergangen", contains variegated copper, occurring partly in quartz kidneys and partly impregnated in schist. The vein has been uncovered in several places for a length of about 450 m. but the different impregnations are not found on the same level. (There are at the different places impregnations in different slate zones that will complicate the working considerably. In addition, the impregnation is irregular, and in my opinion, taken as a whole, this vein has little or no technical importance. From the Drathammeren mine an adit is now being driven to the slate-vein "Skiffergangen" /Witt's shaft, which is now 12 meters deep/. According to the plan of the mines this adit will in a straight line be 155.2 meters long and will come in at a depth of 34.7 m. below the surface /22.7 m. below the bottom of Witts shaft/. (At my visit the adit was, including bends, about 130 meters long. There thus remains about 25-30 meters before the schist vein is reached. The cost per meter adit is about Kr. 80.- including cartage, rails &c. The remaining costs will thus amount to Kr. 2,000.- to Kr. 2,500.-. There is in my opinion no reason why this adit should be continued. When one already has a large and promising deposit then work and capital should not be split up in simultaneously prospecting deposits of small promise.

Christiania, 24 Aug. 1907.

Sgd. Johan H. D. Vogt.

Professor.

[illegible]

170141

Form - 9-60 (Rev. 1-18-60)

1. The first part of the document is a letter from the President of the United States to the Congress, dated January 1, 1861. It is a formal communication, and it is written in a very formal and dignified style. The President begins by addressing the Congress, and then he proceeds to discuss the state of the Union. He mentions the progress of the country, and he also mentions the difficulties that the country is facing. He then goes on to discuss the policy of the administration, and he ends the letter by expressing his confidence in the Congress.

2. The second part of the document is a report from the Secretary of the Treasury, dated January 1, 1861. It is a formal report, and it is written in a very formal and dignified style. The Secretary begins by addressing the President, and then he proceeds to discuss the state of the Treasury. He mentions the progress of the Treasury, and he also mentions the difficulties that the Treasury is facing. He then goes on to discuss the policy of the administration, and he ends the report by expressing his confidence in the President.

3. The third part of the document is a report from the Secretary of the Interior, dated January 1, 1861. It is a formal report, and it is written in a very formal and dignified style. The Secretary begins by addressing the President, and then he proceeds to discuss the state of the Interior. He mentions the progress of the Interior, and he also mentions the difficulties that the Interior is facing. He then goes on to discuss the policy of the administration, and he ends the report by expressing his confidence in the President.

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10

1. INTRODUCTION
 - STATE OF TEXAS
 - COUNTY OF DALLAS
 - OFFICE OF THE DISTRICT ATTORNEY
 - STATE OF TEXAS
 - COUNTY OF DALLAS
 - OFFICE OF THE DISTRICT ATTORNEY

Norges Geologiske Undersøkelser
 Bergarkivet
 Rapport nr.: 4482

④

Via Ulvryggen, Friks stoll, i berga
 " Brattkallstegen, i " och i berga
 " Johan Adolfsstr., i Jorngallan

Under byggnaden är:

Via fjorden i byggnaden till kontor och stigarbrytning.

Projektorer och Längd att byggnaden och stigarbrytning:

Via Ulvryggen, den lilla stollan, i by berga.

STUFVINGEN

Till nordiska stufvingsbänk, ett stort ca. 100 st. in. tånger,
 af hvilka 47 st. är omlagade och stoll.

Stollan är:

1/.	Perveigensstr. på svartfjället	5 stoll
2/.	Stellan denna och N. Ariseng	1 "
3/.	Stre: Ariseng via N. Ariseng	5 "
4/.	Via Ariseng	1 "
5/.	Bergsberg	3 "
6/.	Johan Adolfsstr.	7 "
7/.	Skiftningen	1 "
8/.	Horn grös	1 "
9/.	Luttis gata	4 "
10/.	På en parallel med berga.	1 "
11/.	Brattkallstegen	5 "
12/.	På en parallel med berga.	1 "
13/.	Ulvryggen	7 "
14/.	På en parallel med berga.	2 "
15/.	" " " " "	1 "

Stoll 47 stoll.

Å den geologiska kartan är stollan inlagd med flera utsträckta
 linier och några inläggningar som stollan och linier.

Å inläggningar är stollan i Ulvryggen, Stenbjället och
 Bergbjället.

GEOLOGI.

Förkomsterna är delade i nästan af en kontaktzon mellan
 en yngre eruptivbergart, grönsten /gabbro, diorit/ och äldre bergar-
 ter, sandstenar, skifferar och gnejsar, som af Dr. J. Daniel anfördes

till dels halmsystem af neoniska färg och dels jais-system, med
deras dräp, konfundera sig till infus-ö byar, eller kolformationer.

Till halms-systemet hörer och i korta stråkta bandstrecken.
Dessa är vanligen till färgen ljusare eller mörkare grö, som ibland
övergår till svart röd eller brun. Halmsystemets beständelsten är
avarts och bländar; ibland sträcker sig till i större mängd
och ibland färgen olivin och blått. Strukturen är vanligen finkorn-
ig till medelkornig, men ofta blir den arten porfyrisk med strökor
af avarts; och ofta taga dessa kvartskorn utgöras afverander, på
ett sätt som kan betecknas som en avart, som ofta blir konglo-
meratartad.

Dessa sandsten, konglomerat till arten och de kontaktar af en
serie gradvisväxande af en subvolkanisk granit (sällsamt syderit)
af ljus, som färg och mörkare kornstämningar väl utmärkes i mas-
sen.

Sedan man passerat kontaktens har man för sig bergarter dels
lika som diorit och dels dioritiska skiffer, som utgöras utaf en
omvandlingsprodukt af den förra. Dessa dioritiska skiffer skiljer of-
ta ganska mycket till och blir de mera såliga i brottet och för
mindre tydlig skifring. På andra ställen, där skifferna är mera tet
och svart blir skifferfärgen mera regelbunden och skifferna äfven
lättsärförhålligt.

Den af ovanstående graniten är här en tet diorit till färgen
mörk grön, nästan svart.

De följande kopparsulfiderna kunna sammanfattas i
följande af deras relativa läge på följande sätt:

- I) Yngst är de mörka, som bestå af kopparkis och svavelkis,
dels i körtlar af kvarts och kalk och dels i mängder af kvarts
och kalk, liggande i tet graniten eller skiffer. Det kan emell-
lunda att i kalken ligger kiselna mest i bander och i kvarten lig-
ga de mest i klumpar (1, 2, 3, 4, 5, 6, 7, 8, 11, 12).
- II) Kopparkis i kontakt mellan graniten och kvartsa sandsten. Mit
här västra Uverypen (af utskifningen).
- III) Kopparkis och brokis kopparsulfid, som impregneration i gran skiffer
impregnerationen är ganska tunn och finkornig. (7, 8, 9, 10).
- IV) Kopparkis, brokis kopparsulfid och kopparsulfid, som mycket finkorn-

(1) The first of these is the fact that the...
(2) The second is the fact that the...
(3) The third is the fact that the...
(4) The fourth is the fact that the...
(5) The fifth is the fact that the...
(6) The sixth is the fact that the...
(7) The seventh is the fact that the...
(8) The eighth is the fact that the...
(9) The ninth is the fact that the...
(10) The tenth is the fact that the...
(11) The eleventh is the fact that the...
(12) The twelfth is the fact that the...
(13) The thirteenth is the fact that the...
(14) The fourteenth is the fact that the...
(15) The fifteenth is the fact that the...
(16) The sixteenth is the fact that the...
(17) The seventeenth is the fact that the...
(18) The eighteenth is the fact that the...
(19) The nineteenth is the fact that the...
(20) The twentieth is the fact that the...

hög indragning i meridianen (13, 14, 15).

FÄRHAFNS BESKRIFNING AF FYNDIGHETERNA.

Jag vill först behandla de mest obetydliga fyndigheterna, vilka endast några förökningstenar äro utförda, antagligen därför att förökningstenarna äro så obetydliga; alla tillhörande kategorier I) och II).

Grönstensen. Ca. 2 km. från fjorden. Här finnas stora kvarts körtlar och vita förökade koppararkis.

St. N. utvald kottar förök. och A. Arisviken. Förökade koppararkis i kvartskörtlar. Enge de förökade fyndigheterna äro så bearbetade och så ej heller vidare löfvande ut.

A. Arisviken för koppararkis i kvarts och kalk. En mindre senkning finnes ca. 5 m. djup, där dock kvartsförekomst utskilnde i botten.

Vid Arisviken koppararkis i kvarts; ej utbetadt.

A. Ulvåsen i kontakt med kottar (grönsten och kvartsig sandsten, vilka äro till och med utvecklingar af koppararkis i grönsten. Obetydlig förökning.

JOFAN ADOLFS GÅRD. På en red herr kvartskörtlar förökade koppararkis äro 7 utvald tagna och på det mest löfvande stället är förökningsten undersökt med en senkning och tillhörande orter. (se vidstående skissa).

Styrningsriktningen tyckes vara SV till NO och stupningen vertikal. Största maktigheten är ca. 2 m. och då jag 1904 besökte grufvorna var det mycket vacker arkis i dagen, men denna aftog nästint. något närmare undersökande var mig omöjligt, då senkningen var vattenfylld. Fyndigheterna torde ej vara af någon betydelse.

BRÄNNINGEN. Här är en vanlig fängsling i grönstenen. Senkningen utgöres af 2 väl skilda, men tätt till hvarandra liggande skikt, som tillhålls bildas senken. Det öfre skiktet af kvarts, förökade koppararkis och något svarvelskis i klumpar. Det nedre skiktet af kalk förökade koppararkis och tillhålls mycket svarvelskis i renhet. Styrningen är N 60° O och stupningen 15° - 16° mot S. Jänglansens största maktighet är 0,6 m., för öfrigt tyckes denna vara ganska va-

riabel och en tydlig tillslutning af VSV och förslaget.

Sedan en mindre sankning utförts (se kartan) inreafte en stoll från bogen och fyndigheten blev uppridan i fält med orter till en längd af ca. 55 meter. Det visade sig då att det VSV blev hängen till slut förslutna på några sterila kalkföremål spricker och det som blev den en resten ca. 1 cm. tjock.

En mindre sankning gjordes och därifrån gräfvos orter 6 m. under stollens nivå. Vid min beräkning var emellertid allt detta under vatten.

Från öfre fältorten är förberedd en stoll sådan att grifvas till genomslag med vitte myrk på skifferhången. Denna stoll storslolen är nu inrifven ca. 75 m. och har genomgått en steril kvartstrand och för resten blott grönsten.

Under antagande af en längd af 55 meter och en bredd af 0,3 meter får vi en kalkare af ca. 16 m³ eller vid en v. af 2,8 ton pr. m³.

45 ton kalkare pr. rikskat meter.

Att genomsnitt af denna kalkare har gifvit 50 cu. (Enligt Arkhus uppgift 4 p. 50.). Enligt Hattuskontorets uppgifter ca. 3,16 p. cu..

För att utvinna denna kalkkvantitet måste ca. 55 m³ berg uttagas hvarann kalkproben toras bli ca. 30 m. Någon skräddning af kalkmassan kommer ej ifråga utan endast utskräddning af grönstenen. Priset per ton berg kommer antagligen att falla sig omkring Kr. 5:-, hvilket ger oss ett pris per ton kalk af Kr. 15:- fritt kaj vid Hattusfjorden. Nu kunna dessa siffror möjligen ställa sig gynnsammare, då grufvan brytes i staden med en annan större staden.

Med de nuvarande höga priserna på koppar kan ju detta bara sig, och om priserna skulle falla, blir det förtjänsten ingen, såvida ej hängen skulle öka mot slutet, något som kan ej har rättighet att tro.

HATTIS OCH HÄNS JÄGAR. Sedan föra koppar och brokis koppar-kalk i skiffer och är underkasta på 2 ställen med mindre sankningar ca. 3 meter djupa, hvarann kan således har ringa kalkare och dessa kalkar. Hattis sig ut; Hattis bättre, men blott ett smalt parti

THE FOLLOWING IS A SUMMARY OF THE RESULTS OF THE SURVEY OF THE
- 1 -

1.00 DOLLAR VOLUNTARY CONTRIBUTION TO THE UNITED STATES GOVERNMENT
- NO MORE THAN 10% OF THE TOTAL CONTRIBUTION -
(CONTRIBUTION TO BE MADE BY THE DONOR)

1. The first part of the document is a letter from the President of the United States to the Congress, dated January 1, 1861. It is a formal communication, and it is written in a very dignified and official style. The President expresses his regret that he cannot deliver the message in person, and he explains the reasons for this. He then proceeds to discuss the state of the Union, and he mentions the recent election of Abraham Lincoln as President. He also mentions the secession of the Southern States, and he expresses his hope that the Union will be preserved.

THE UNITED STATES OF AMERICA
DO hereby certify that
the following is a true and correct copy
of the original as the same appears
in the records of the Department of State.
IN WITNESS WHEREOF, I have hereunto set my hand
and the seal of the Department at Washington,
this _____ day of _____, 19____.

[Signature]
[Seal]

[illegible]

ALL INFORMATION CONTAINED HEREIN IS UNCLASSIFIED
DATE 08-19-2006 BY 60322 UCBAW

ULVRYGGEN.

är otvivelaktigt och blanda af alla närvarande förekomst. Med föreligger nämligen en impregnationszon eller flera linieformiga sådana i en sandsten, som sträcker sig 45° ö. och stupar ibland 90° åt NO ibland 60° åt SV eller generellt sett hela tiden står vertikalt. Riktigheten af denna impregnationszon har sitt maximum, såvitt man kan se i sektn., vid A, där den är ca. 40 meter; dock är det att märkas att någon exakt gräns till den livande bergart sandstena finnes ej, åt NV, där dermot åt SO är den mycket tydlig.

Den impregnationsförmåga sandstena är en ibland med uträkning, åt rätt ibland svagt grönskig, dels troligen beroende på kopparkarbonsat och dels på närvaron af olivin och silicet, dessa svagt grönskiga mineral. I sektn. är den mycket förvittrad och visar ofta stora af malakit grönsfärgade ytor, som gifvit anledning till en tillägnad. Strukturen är fin- till medelkornig, men blir ibland grovare på t.o.m. porfyrisk med strökor af kvarts.

Kopparn förekommer som kopparaktig bronit, kopparmal och kopparplans, oftast i mycket små korn samt förefinnas i strökartens massa; blott de denna är porfyrisk synas kopparmineralen anordnade vid strökorna.

De rikare partierna, påträffade på ställen med stoll, tycks vara linieformigt förefinnas i impregnationszonen. Den hittills vunna erfarenheten tycks på vid handen att dessa partier ligga närmast zonen SO kontakt. De rikare delarna hålla ca. 3,2 % Cu och några från 0,24 % till 1,5 % Cu..

Största riktigheten är som sagt 40 meter dock ej sammankommande. Angående fyndigheterna djup, så borde detta ju vara stor om malmineralet är en ingående beståndsdel i bergarten och således torde vara bildadt på samma gång som denna; men i andra ställen har man en storliggande erfarenhet af kopparplans och bronit kopparmal att dessa malmer ej brukas så djupt, hvareför jag tror att man ej får anse detta djupet så stort.

På Ulvryggen äro 7 uttal lagar, deraf ett enligt uppgift i lufsen, hvaraf följer att fyndigheten får en längd af ca. 1600 meter, som hela tiden kan följans i sektn. och den ligger på sina ställen ser mycket fattig ut.

de utförda arbetena inskränkt sig till 2 stollar (accuton en nu (börjad) och diverse förutseende i dagen. Där sedan följer en öfversikt af denna stoll.

Briks stoll (salen = ca. 240 meter o.m.) de kartorna, är drifven med två tvärlag till salen ca. 17 meter. Sedan drifves klassa och kortens korter, och förut med 2 tvärlag mot NV och en senare med 1 st SV och ett st NV, från hvilket korter en stiftort är drifven ca. 6 meter nedt. Af korten framgår blott loppet af den rikare zonen af impregnationen. Följande analyser äro utförda:

Kortens ort. Tvärlag No. 1.

Generalprof af de 4, 5 första meterna	3.24 % Cu
" " " 4, 5-7, 8 meterna	1.51 % Cu
" " " 7, 8-12, 0 meterna	0.42 % Cu

Tvärlag till korten.

" " " 4, 5 första meterna	0.24 % Cu
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Klassa ort. Tvärlag No. 2.

" " " 4, 5 första meterna	1.50 % Cu
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Briks stoll. Rätt generalprof.

På hela saluberoen i stollan.	1.13 % Cu.
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Mycket nytt rikt parti har af den klassa ort påträffats, och det denna är inskrifven ca. 17 meter i den fattiga salen. Alltså se vi att afstannet mellan de rikare partierna kan vara ganska långt.

Jonas stoll. Salen ca. 200 m. SV om Briks stoll, och den en riva af ca. 40 m. högre än förutnämnda stoll. Jonas stoll är inskrifven ca. 22 m. med två tvärlag till salen. Först genomgicks

5 m. god salubero	3.25 % Cu.
sedan 4 m. " " "	1.4 % Cu.
och så 3 m. " " "	0.36 % Cu.

Alltså en sammanfattning af 17 meter ca i genomsnitt 2.04 % Cu. Detta är ett vackrare resultat än vid Briks stoll, men feltutredningen af det goda partiet är ännu okänd. Fält orter äro dock förbjuda att begå sig i den bästa salen. Några resultat af dessa torde man ej kunna få på en tid.

På mindre strömmen finnes utifrån ca. 15 m. V. om framsen mellan utskott 2 och 3 vidnande med malakit och oxvandringsprodukter af kopparians och brockig kopparsulfid.

Ved A ca. 80 m. SV om Grønsholm Løkken strækken 2 og 3, ligger flod i botten, ingen dagstrosser, tilvandede og et godt vand nærtliggende af ca. 40 meter. Generalprofil af denne strosser var 1,04 p. 00. Ca. 30 meter under dette stiller er en lille flod, ca. 10 meter efter ca. 80 meter vand træffes Løkken.

I en stross ca. 100 m. efter om Grønsholm Løkken strækken 4 og 5 ligger det i den sønderste og brede kopperdel tillige nærtliggende af ca. 6 meter.

Ca. 30 meter efter om Løkken strækken ligger det mest kopperdel i en forstyrret bergart. Strossen var ca. 10 meter bred og 100 m. afvæjning af ca. 110 meter o.k.. Ca. 100 meter SV om denne stross er nærtliggende i blødt og blødt ca. 110 meter o.k..

Ved strækken Løkken 5 og 6 findes en liten stross ca. 7 meter høj og ca. 10 meter bred og 5 meter bred. Et generalprofil var 2,12 p. 00.

Ca. 100 meter SV om strækken stross er nærtliggende i den sønderste del og brede kopperdel i en forstyrret luge søndersten. Strossen var nærtliggende og nærtliggende. Styrning i 45° 0 og strækken ca. 80° af 10.

Ca. 70 meter SV om denne plads ligger Lille-Elvemyr (og udsat på kartet) og på denne plads af dette vand ligger søndersten, men der var intet nærtliggende.

At nærtliggende, nærtliggende vises tydeligt, at der har været et godt vand en kopperdel af den evnede strækning og nærtliggende i den. Den totale længde er, som nærtliggende, ca. 1650 meter på de 6 stræk, men nærtliggende nærtliggende nærtliggende er på denne nærtliggende. At kartet nærtliggende at de stræk 3 og 4 er et stykke om ca. 550 meters længde, på hvilket intet arbejde er nærtliggende. Denne bergart findes på, og der og der nærtliggende i den, men om kopperdel nærtliggende sig selv tiden er nærtliggende. I stræk 1 nærtliggende at 10 har man via forstyrrelsen og via nærtliggende nærtliggende nærtliggende nærtliggende i den, det nærtliggende på være en nærtliggende i felt af Elvemyr og i en lille nærtliggende på længden nærtliggende nærtliggende, nærtliggende till flere nærtliggende meter.

Da jeg er i nærtliggende af nærtliggende kart og ligger har at

flörens analys- och beräkningsrapporter och jag ej ha upgjort några
närmare planer och beräkningar, men jag vill dock angiva några siff-
ror som kunna vara af intresse. Jag står af från början från att
den blifvande anläggningen skall bli ett bly för Elmsjö's system.
Med eller utan detta och att sålunda transporteras ned till verket via
sjön och linorna. Elmsjösystemet känner jag tyvärr så litet till att
jag ej kan göra några utsläpningar etc.

Via Elversjön tror jag att man kan komma ned en sträcka på
ca. 12 meter, så höjdhöjden af koppar får sättas till
1-2 p. ca. och fördelen till 1600 meter; vilket belämnas 18.000 kr.

Enom profileringen så vi att salar på fjerns stoll liknar ca.
40 m. högt en bricka stoll, som ca. 40 meter högt en rymlighetens
höjsta punkt. Om vi nu rekna objekt till bricka stollens nivå så få vi
en genomsnittlig siffra af 40 meter, d.v.s. en salavolym af
2.540.000 ton. I min första lilla rapport antog jag minsta siffr., som
högre kopparhalt, hvilket ej torde vara riktigt.

Liten och antaga en salar af 20 och total brytning pr. år af
125.000 ton, så få vi alligen 100.000 ton sal och skulle då tydlig-
heten råka i ca. 25 år. Under förutsättning af beräkningsanläggning
torde det salmaltiga bergat kunna levereras via verket via sjön för
kr. 2:- pr. ton. Innesknekt alla kostnader, äfven amorteringar.
Man kan väl det visst bli att allt går för sig billigare om berglös-
ningen kan bli bättre än jag beräknat, hvilket är antagligt om
bergarten såsom varandes sandsten är såsom inneskogen.

Erfrorderliga anläggningar skulle vara:

Driftcentral för vatten eller petroleum (storleken beroende på
vidare anläggning).

Beräkningsanläggning kr. 18.000:-

Linor (7 km.) " 150.000:-

Kar och magasin " 30.000:-

Bostäder till en bergman " 50.000:-

Alla siffror äro naturligtvis endast förslagsvis uppgatta, så
allting skall bli så obestämt. Likaså kan ej heller någon värdering
givas.

Stavelen den 10 Sept. 1906.

Aug. Siljeström.

Schürf-Ergebnisse (Bericht Goldkuhl)

östl. Ulveryagen. Am Tage zwei Erslinsen von zusammen 2530 qm mit 1,5 % Cu.

Olles-Erze.

Am Tage acht Linsen mit zusammen 789 qm und 1,36 % Cu oder bei Hinzunahme Ernerer Randpartien 10510 qm mit 1,66 % Cu. Linsen zeigen in allen Horizontalen plötzliches Absetzen. Der 12 m u.O. angeschlagene Quadratstollen erbrachte zwar ein Erz von 1,4 % Cu aber augenscheinlich keinen Zusammenhang mit den zutage anstehenden Erslinsen. Eine an seinem Feldort niedergebrachte Bohrung von 176 m zeigte nur Spuren von Cu.

Das gleiche Schicksal fanden drei an dem weiter östlich angesetzten Stollen niedergebrachten Bohrungen, die die Zonen der Tageslinsen ergebnislos durchfuhren. Der 30 - 40 m u.O. angesetzte Olle-Stollen unterfuhr ergebnislos 3 reiche Tageslinsen.

ohne-Erze.

In diesem Gebiet wurden 3 Linsen mit zusammen 1669 qm und 1,91 % Cu angesetzt. Im flach angesetzten Stollen wurden 3 Erslinsen festgestellt, die aber mit den Tageserzen keinen Zusammenhang hatten. Ein in ihn getriebenes 14 m Gesenk fand ein reiches Erzgestein mit 2,5 % Cu.

Eriks-Stollen.

Im Eriks-Stollen wurde ein 500 qm grosses Erzareal mit 2,04 % Cu gefunden. Ein im besten Erz angesetztes 20 m Gesenk zeigte als Durchschnitt 0,66 % Cu und ein kurzes Gesenkort 1,0 % Cu. Ein 34 m tiefes Bohrloch verlief im Tauben. Dagegen wurden in einem Gesenk mit Gegenort gute Aufschlüsse nachgewiesen.

Hans-Stollen.

Weder über Tage noch in Stollen irgendwie nennenswerte Erzfund.

Kopparförekomsten vid Lopperfjord.

Inledning: Kopparen uppträder i form af bornit och kis samt andra förvittringsprodukter, såsom finkornig impregnation i sandsten. Den sträcker sig från sydvest till nordost af sträckande sandstenszonon af flere kilometer längd och 100 till 200 m. bred samt klassificeras såsom sparagmit. Synbarligen är hela sparagmiten färsedd med spår af koppar, - på sina ställen är detta fallet i hög grad, så att linsformiga förekomster uppträda. En afgränsning mellan den torra, fattiga och hopade sparagmiten är icke förhållande, och endast för det ofvade öget är olikheter i bergartens kvalitet förnimbara. Synbarligen har icke heller något system konstaterats i linsernas lagring och storlek.

Öppningar: De hittills verkställda uppdagningsarbetena bestå i ett antal gånger, omkring 25, hvilka på en längd af 2 km. äro fördelade i sparagmiten och anlagda lodrätta för att bestryka densamma. Mellan gångarna, hvilka rita 10 till 35 m. och hafva ett djup af omkring 1 m., äro i många fall kopparspår synliga, så att man kan sluta till sammanhängande malmassor, det kan likväl ej konstateras, om vi hafva att göra med några stora eller många små linser.

Vidare hafva 4 stollar slagits på tvären i bergväggen, medelst hvilka några linser, hvilkas tillvaro var bekant genom gången, framförts och delvis utförts (undersökts).

I Olles stoll blef på 60:de metern en lins af 14 m. mäktighet framförd. Man har ej vidare undersökt den, utan driver tvärslaget vidare mot norr för att gå under en andra 2dra utholkning.

I Johnstollen framfördes en lins. Densamma innehåller vid en bearbetningshöjd af 20 meter omkring 20000 tons ralm å 1,5% Cu. (koppar). (undersöktes)

I Erikestollen framfördes en lins, hvilken vid en bearbetningshöjd af 24 m. innehåller omkring 14000 tons malm med en halt af 1,6% Cu.

I Hansstollen framfördes på 5te metern en lins af 4 m. mäktighet. Stollen bearbetas vidare.

* *afgränsningar*

Kopparhalt: Enligt uppgift har Nordiska ej låtit analysera några prof från urholkningarna.

De på uppdrag af Skandiakobber anställda undersökningarna hafva givit följande resultat:

Den västliga gruppen af urholkningarna gaf:
0,7 0,8 0,5¹² 1,0 1,5 1,1 i medeltal 0,8%

Den mellersta gruppen vid Ollesgrufvan:
1,4 1,1 1,1 2,4 1,4 1,8 1,7 i medeltal 1,5%

Den östliga gruppen vid Eriks & John gaf:
1,7 0,4 0,4 0,2 i medeltal 0,3%

I Ollesgrufvan:
1,1 0,8 1,3 i medeltal 1,0%

I Johnsgrufvan ur linsen:
3,6 1,3 1,1 1,6 0,3 0,8 i medeltal 1,5%

I Johnsgrufvan den angränsande bergarten:
0,3 0,3 0,7 0,2 0,0 0,1 0,6 i medeltal 0,3%

Eriksgrufvan ur linsen:
1,8 2,4 1,3 1,1 1,3 i medeltal 1,6%

Eriksgrufvan det angränsande berget:
0,26 0,13 0,1 0,3 0,23 0,39 i medeltal 0,2%

I Hansgrufvan:
0,2 0,1 0,8 i medeltal 0,4%

Af dessa prover, hvilka i allmänhet öfverensstämna med Nordiskas analyser, och de bifogade planerna framgår, att den vida största delen af gruffältet innehåller mindre än 1% Cu och följaktligen icke kommer i betraktande, under det att den mellersta delen vid grufdrift torde komma att gifva 1,5% Cu. Ett skiljande för hand af det större "Hauptwerk" synes outförbart, däremot måste driften upphörligt kontrolleras analytiskt.

Det föreligger inga bevis för fortfarandet resp. för upphörandet af malrhalten på djupet, och den omständigheten att i ett borrhål i Eriksstollen och en försänkning i Johnsstollen ett aftagande af halten

Af Gernet: Riffersfjord.

visande sig, för ej väga för tungt i vägsken, då två resultaten i öka öro
nos för att därpå uppfylla en teori.

Metallurgi: Malmen i och för sig särskild i de öfre lagren många för-
vittringsprodukter af koppar, vilka lätt äro lösliga i fortuna-
syror. Genom försök med ett malmsprof, som var sammansatt af 50 olika
prof af alla slag, fastställdes, att i en 12-tig svafvelsyrelösning i
håld på 5 minuter löstes 20% af den i malmen innehållna kopparen.
Af ett rostadt malmsprof löstes genom en 5.-tig lösning af SO_4H_2 på
24 timmar 33% af den i malmen innehållna kopparen.

Af dessa försök framgår, att vid användning af Elmores förfarandet äro
betydliga kopparförluster oundvikliga, om ej den valda malmen förut är
tvättad med fortuna SO_4H_2 .

En annan förbättringsmöjlighet är utvaskningen af den rostade mala-
en, - i båda fallen kan urskiljandet af kopparen af ur lösningen följa
sivell genom cementation med järn, som äfven genom elektrolys enligt
Siemens & Halske.

Vattenkräfter äro, åtminstone under so i ären, i tillräcklig mängd för
handen.

Konklusioner: Undersökningarna i besvarandet
af frågan, om Skandia, motsvarande optionen, skulle köpa Riffersfjord-
grufvorna för ett pris af 1.300.000 kronor. Dessa fråga måste besvaras
nekande, då de uppgifne resultaten ej motsvara priset. Äfven är det ej
möjligt, att detta till den 1 November skall blifva fallet. Öfverhufvud
representera grufvorna i sitt nuvarande skick intet faktiskt, utan en-
dast ett spekulativt värde, om man kan förskaffa sig en klar
bild af malmsärvorna och rentabiliteten, äro utomordentliga arbeten
erfordrliga, vilka minst taga ett år i anspråk och skola kosta 100000
kronor.

Men en förlikning af optionen med den ofvannämnda tiden skulle
endast i det fallet vara att rekommendera, om det köpris, som skall be-
talas för grufvorna, underginge en betydlig nedsättning, ty sannolikheten
talar för, att de bearbetningsvärda malmassorna utgöra endast en liten
del af hela impregnationszonen. Genom urkalkningarna har en

Afskrift.

(4)

af ca 5000 kv. l. 2 malm p. visats. For beregnet af de i udsigt
stillede malmreserverna dela for tilvarende stød, at beskaffenheden af
malmbærekraften p. djupet endda er for litet bekant.

Underteeknadt A. von Gernet.

Geokjemiske undersøkelser

Repparfjord 1959.

I. Innledning.

Sommeren og høsten 1959 ble det gjort en del geokjemiske undersøkelser i området rundt kobberforekomstene ved Ulveryggen, Repparfjord.

Initiativtager til undersøkelsene var statsgeolog Paul H. Reitan ved Norges geologiske undersøkelse. Reitan bestemte hvilke arealer som skulle dekkes, og utførte med sine folk fra NGU prøvetakingen. Forøvrig ble undersøkelsen utført etter Statens råstofflaboratoriums direktiver, og SR var ved sin geokjemiker ansvarlig for den praktiske gjennomføring av oppgaven.

Denne rapport skal gi de viktigste data om de undersøkelser som er gjort.

II. Geografi.

Det undersøkte område ligger i Kvalsund herred i Finnmark fylke, og således mellom 70. og 71. breddegrad. Høyden over havet varierer mellom 0 og ca. 450 m. Middelttemperaturen er $+1.5^{\circ}\text{C}$ og middel nedbør 676 mm pr. år. Terrengtet er småkupert, tildels meget bratt og har tallrike bekker og småvann. Spesielt de høyere liggende deler av feltet er praktisk talt uten løsavleiringer. Av denne grunn er vegetasjonen også ytterst sparsom.

III. Geologi.

Geologisk hører feltet til Raipas formasjonen, som er et grunnfjellsvindu i overskjövne kaledonske bergarter. Malmmineralene er hovedsakelig bornit og kobberkis. Kuprit, kobberglans og malakitt forekommer ofte nær dagoverflaten. Malmmineralene opptrer finfordelt i en arkosisk sandstein. De enkelte malmmineralkorn er oftest så små at de er meget vanskelige å se med det blotte øye. (Se Vokes, NGU, nr. 200/Årbok 1956, s. 74-111, Reitan, NGU, nr. 203/Årbok 1957, s. 76-79).

IV. Hensikt med undersøkelsene og plan for arbeidet.

Kobbermineralenes finkornethet gjør det vanskelig for geologen i felten å skille mellom mineralisert og umineralisert sandstein. Det er dette forhold som - til tross for at svært meget fjell er blottet - gjør det ønskelig med en indirekte malmetingsmetode i feltet.

Hensikten med de geokjemiske undersøkelser var derfor:

1. Studere brukbarheten av geokjemiske malmetingsmetoder i det foreliggende felts spesielle geologiske og geografiske forhold.
2. Forsøke å finne eventuelle nye kobbermineraliserte soner.

I overensstemmelse med dette ble det planlagt, først å konsentrere prøvetakningen rundt de kjente kobbermineraliseringer, og senere eventuelt dekke de omliggende mindre kjente områder.

V. Prøvetatt område, antall prøver, tidsrom.

Det prøvetatte område er et vel 10 km² stort areal av rektangulær form, og hvis grenser mot SØ og NØ dannes av henholdsvis Vestre Ariselv og kystlinjen fra Ariselvens utløp til Repparfjord poståpneri. Området ligger i arkositten, og Ulveryggen danner det sentrale parti. Som kontroll er det også tatt en del prøver i Østre Ariselv som renner i grønnstein.

Det totale antall prøver er 44, herav er 46 tatt i fra Ø Ariselv.

Prøvetakningen er utført av 2 mann på 12 dager i tidsrommet mellom 15. juli og 25. august 1959.

VI. Prøvetakning.

I det dekkede område ble det gjort systematisk prøvetakning av sedimentene fra alle bekker.

Bekk defineres da som: Rennende vann og tørt bekkefar.

Bekkesediment defineres scm: Lös avleiring, uorganisk eller organisk i eller nær bekkeleiene.

En uorganisk prøve tatt under vann og med silt sammensetning ble foretrukket, men i mangel av slik prøve ble det tatt prøve av sand, grus eller humus, i eller utenfor bekkeleiene, alt etter som forholdene på hvert enkelt prøvested var.

Avstanden langs bekken mellom prøvepunktene varierte mellom 100 og 400 meter med ca. 250 m som det vanligste tall.

Prøvene ble puttet i nummererte prøveposer. Posene ble pakket i små Al-bokser, som igjen ble satt i tre-kasser. Prøvene ble i denne emballasje sendt til SR's feltlaboratorium i Kautokeino (de siste prøvene direkte til Trondheim).

Alle data om prøvene ble notert ned på SR's standardformular for dette bruk.

VII. Tørrking og sikting.

Ved laboratoriet ble prøvene tørket ved ca. 80°C. De tørre prøvene ble sikket gjennom 80 mesh sikt av rustfritt stål. Den fineste fraksjon ble puttet i ny papirpose og tatt vare på for analysering og arkivering.

VIII. Analysering.

Prøvene ble analysert etter Holmans metode. Dette er en halvkvantitativ metode med følgende prinsipp:

Et kjent volum av sedimentprøven rystes med et 2 fase-system bestående av a) Vannfase: Ammoniumcitrat-puffer av pH 2, og b) Organisk fase: En oppløsning av dithizon i white spirite. Dithizon er et organisk farvestoff, som gir sterkt fargede forbindelser med de fleste tungmetaller bl.a. kobber. Ammoniumcitrat-pufferen sørger for ekstraksjon av de ønskede metaller fra sedimentene, og gir samtidig den rette pH for selektiv kobberbestemmelse med dithizon.

Man får ved Holmans metode således løst ut bare

en bestemt, mindre fast bundet fraksjon av det totale kobberinnhold. (For nærmere detaljer henvises til Bull. Inst. Min. Met. oktober 1956, side 7-16).

Metoden er meget hurtig, idet 1 person vanligvis gjør 3-400 enkeltbestemmelser pr. dag. I dette tilfelle gikk analyseringen betraktelig langsommere på grunn av de høye kobberkonsentrasjonene.

Gjennomsnittlig analysefeil er under 30 relativ % i området fra "0" (d.v.s. < 1) til 100 ppm (parts per million).

Analysefeilen ble hele tiden kontrollert ved analysering av standardprøver puttet inn på tilfeldig plass i hvert 20. antall prøver.

IX. Resultater.

Resultatene av Holmans prøve finnes på de 2 medfølgende plansjer.

Plansje 1. M = 1:20 000 viser tallene for bekkesedimentenes kobberkonsentrasjoner i parts per million (vekt/volum).

Plansje 2. M = 1:20 000 er de samme resultater fremstilt mer oversiktlig, idet forskjellige konsentrasjonsintervaller er gitt tegn av forskjellig størrelse og farve.

Av plansjene fremgår det at man her er i et område med høy kobberbakgrunn, idet denne i arkositten er 8-10 ppm.

De kjente gruver og skjerp gir seg til kjenne med anomale kobberkonsentrasjoner i de nærmeste bekker (opp til 600 ppm).

Mot nord-vest i det dekkede felt er det også et lite område med høye kobberkonsentrasjoner.

X. Diskusjon av resultatene.

Man har andre steder i Finnmark høstet den erfaring at geokjemiske metoder er meget følsomme i områder med sparsomt overdekke. Dette understrekes av de her foreliggende resultater. Det er enkelt å forestille seg at sporelementene

forholdsvist lett løses ut av det faste fjell i godt blottede områder på grunn av vannets og luftens uhindrede virkning. Det må dog her gjøres helt klart at den høye bakgrunnsverdi for kobber ikke bare kan skyldes manglende overdekke. Man har godt blottede områder andre steder i Finnmark, hvor bakgrunns-konsentrasjonen for kobber er adskillig lavere. Grunn for den høye bakgrunn i områdets bekkessedimenter er derfor etter alt å dømme, også et generelt, relativt høyt kobberinnhold i det faste fjell.

De høye tall rundt de kjente gruvene kan skyldes store fjelloverflater på grunn av sprengningene. Mot at dette er hele forklaringen taler det faktum, at man også utenom de kjente forekomster har et område med høye kobber-konsentrasjoner (N - V i det dekkede felt).

7. Konklusjon.

Det dekkede område er så lite, at man bør være forsiktig med å trekke altfor vidtgående konklusjoner. Jeg finner det likevel riktig å si at resultatene oppmuntrer til videre undersøkelser. En mer eksakt vurdering kan vanskelig gis, før man har dekket et større areal, og anomaliene er fulgt opp med detaljerte geologiske undersøkelser.

Trondheim, 27. november 1959.
Statens råstofflaboratorium

Björn Bölviken
Björn Bölviken

Repparfjord

Case History

Metal:

Copper.

Dispersion Type:

Secondary, stream sediments.

General Locality:

Northern Norway.

Source:

Internal report by Björn Bölviken,
Statens råstofflaboratorium, Trondheim, Norway.
Vokes, F.M., 1957, N.G.U. Nr. 200, p. 74-11.
Reitan, P.H., 196-, N.G.U. Nr. 2-- (in preparation).

1. Location: Repparfjord, Kvalsund, area of Finnmark district, Norway.
Lat: $70^{\circ}30'$ N. Long 24° E.

2. Physical Features:

a) Topography and drainage:

Hilly country above ca. 200 m. Below 200 m. steep slopes to the S. Altitude variation 0-450 m in 10 km². Numerous small streams.

b) Climate:

Seasonal rainfall 676 mm, monthly variation from 29 to 95 mm.
Mean temperature range $+1.5^{\circ}$ C, extremes: Jan. -5.2° C, July $+8.9^{\circ}$ C.

c) Vegetation:

Very little. Moss and heather. Birch forest up to ca. 200 m a.s.l.

3. Geology:

a) Geological setting (App. SR 108):

Arkosic sandstone, greenstone and green schist of a Precambrian window in overthrust Caledonides.

b) Mineralization:

Mainly bornite and chalcopyrite in the arkosic sandstone. The copper minerals are often so finely disseminated that they can hardly be seen in the specimen. Covellite occurs as a supergene replacement of bornite; thin incrustations containing malachite, azurite, and chrysocolla occur on the surface of mineralized sandstone.

4. Weathering and Overburden (App. SR 108):

Depth of weathering of bedrock and mineralization negligible. Above ca. 350 m the overburden is minimal. Between ca. 350 and 200 m ca. 50 % cover by soil < 10 cm thick; a side moraine up to 500 m broad crosses part of the area. Below ca. 200 m generally birch forest and considerable thin soil cover.

5. Geochemical Dispersion Patterns in the sediments (App. SR 109):

a) Background concentrations:

Background concentration of Holman test in - 80 mesh fraction of sediments is 10 ppm Cu.

b) Magnitude and form of anomalies:

Cu-concentrations are up to 600 ppm in streams draining area around old prospects. Anomalies can be traced for about 2 km downstreams. Anomalies in virgin part of covered area are maximum 60 ppm.

c) Special features:

The highest anomalies are supposed partly to be due to large rock surface areas, caused by old mining. This is probably not the whole explanation, due to the fact that a new anomalous area is found 2.5 km further N-W, where there has been no mining activity.

6. Field Technique:

a) Sampling:

Stream sediment samples, big enough to yield ca. 5 g dried and sieved material, were collected systematically along all streams at intervals of approximately 250 m. One sample was collected from each side of the stream at each sample site.

b) Analytical:

The samples were dried in air current about 80° C, and sieved through 80 mesh sieve of stainless steel. Analytical method: Holman test, lab. method. (Dull. Inst. Min. Met. 1956, 7-16).

7. Geochemical Field Organization:

The aim of the investigation was to try out the stream sediment method in the area. Sampling was done by two geologist's assistants in 12 days during summer 1959. Total number of sample sites 223 (446 samples). Covered area 10 km². No other indirect prospecting methods used.

8. Results of Investigation. (App. SR 110):

Significant anomalies obtained near known mineralization. One new anomalous area found, but anomalies there are of less magnitude than those near known mineralization. The cause of the anomalies which are not related to known areas of mineralization has not yet been investigated.

Trondheim, Norway, 18th December 1959.

Statens råstofflaboratorium

Björn Brinken
Björn Bølviker.