

THE COPPER RESOURCES OF NORTHERN MICHIGAN

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by

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The views and conclusions contained in this document are those of the author and should not be interpreted as necessarily representing the official policies or recommendations of the Interior Department's Bureau of Mines or of the U.S. Government.

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The copper resources of northern Michigan are in the Precambrian X rocks of the Marquette Supergroup, and in the Precambrian Y Keweenaw volcanics (native copper) and Nonesuch Shale (mostly sulfides). The available reserves in the volcanics are ≈1 billion pounds of copper. Approximately 80% of the volcanic outcrop belt which includes the Portage Lake Lava Series and the overlying unnamed formation has not been explored by drilling. This area, which extends from Keweenaw Point to the Wisconsin border, could contain one or even two orders of magnitude of the native copper now available. Metal reserves in the Nonesuch Shale are estimated to be ≈12 billion pounds of copper and 90 million ounces of silver. The Marquette region is estimated to contain approximately 6.6 billion pounds of copper. The only producing mine is at White Pine. The native copper and Marquette resources will require new technology or a large increase in copper price before economic exploitation of the deposits is feasible.				
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FOREWORD

This report was prepared by John T. Wilband, Department of Geology, Michigan State University under USBM Contract number J0366067. The contract was initiated under the Minerals Availability System (MAS) Program. It was administered under the technical direction of the Eastern Field Operations Center in Pittsburgh, Pennsylvania with Ted Drescher acting as Technical Project Officer. David Williams was the contract administrator for the Bureau of Mines. This report is a summary of the work recently completed as a part of this contract during the period August, 1976, to April 30, 1978. This report was submitted by the author on May 1, 1978.

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Access to much of the data summarized herein was granted to the Principal Investigator by Mr. Robert Seasor, Chief Geologist of White Pine Copper Company; Mr. James T. Schaefer, President, Universal Oil Products Company; and Mr. Robert Reed of the Michigan Geological Survey. Insofar as some data generously supplied to me were considered proprietary, they have been omitted from this report.

The courtesies shown me during this investigation by Mr. Ross Grunwald and staff of Homestake Copper Company, and by Mr. Robert Seasor and staff of the White Pine Copper Company are acknowledged with gratitude. Without the close liaison with the staff of the Michigan Geological Survey through virtually all phases of the data compilation this investigation could not have been successfully concluded.

Finally, the services of several students, especially Michael Price, William Morris, Susan Crissman, Mary Jane Klesney, Kris Roberts, and Sherry Kreger are acknowledged with thanks.

Gerald Anderson of the Cleveland-Cliffs Iron Company kindly granted permission to use Plate 1.

INTRODUCTION

The native copper deposits of northern Michigan have been mined continuously from 1844 to 1968 and constituted the principal source of domestic copper for 40 years. Unfavorable economic conditions and labor difficulties resulted in a cessation of mining in 1968. The only significant copper production from Michigan now comes from the White Pine mine. There does not appear to be any chance of reopening the native copper mines of Houghton or Keweenaw Counties in the near future, particularly since the underground exploration and mining feasibility program at the Centennial No. 3-6 site as well as surface exploration programs were abandoned in 1977.

Available reserves of the native copper district are estimated to be ≈ 1.0 billion pounds of copper in mineralized zones which range from 6 to 30 pounds of copper per ton. The total reserves of the district are probably in excess of 10 billion pounds of recoverable copper. This copper resource cannot be economically exploited at the present level of technology.

Total copper reserves, largely in the form of sulfides at the White Pine and Presque Isle (Western syncline) deposits in the Nonesuch Shale, are estimated to be approximately 12 billion pounds of copper. Of this total, the extractable reserves as of January, 1977 at White Pine are approximately 4.8 billion pounds of copper based on a cut-off grade of 20 pound copper per ton (in-place). The reserves potential at the Presque Isle deposit await an economically feasible mining method before further development. The mineralization in part underlies the Porcupine Mountain State Park and would,

no doubt, be expected to satisfy rigorous environmental legislation and withstand public opposition to open a mine at the site.

Other potential sources of copper in the State of Michigan are the Kona Dolomite and the Michigamme Slate within the Marquette trough near Marquette, Michigan. Although these occurrences are of low grade, exploration work continues within the Marquette region.

HISTORICAL INFORMATION

Prehistoric Mining

It has been estimated that there are approximately 5,000 known pits (Drier, 1961) up to 30 feet deep, from which native copper was mined prehistorically. Drier (1961) estimates that 500 million pounds to perhaps more than one billion pounds were recovered.

The miners built fires on top of the copper bearing veins, then brushed the fires away and poured water on the surface to fracture it. Charcoal at the bottom of some of the pits has been dated as 4,000 years before present. The pits have been found the length of the copper bearing region, extending 100 miles from Keweenaw Point to the Porcupine Mountains.

So far little is known about the miners. No burial grounds have been found. No carvings, writings, paintings, or art work, except copper tools were left behind to provide clues of their culture or place of origin. The fact remains, however, the centuries of labor were represented by the area. It is assumed that such effort was probably not simply to obtain copper for trinkets and ornaments, but for tools, armament, and for trade.

The significance of this prehistorical mining became apparent in the 19th century copper rush when the extensive pits led prospectors to most of the known deposits.

Early History

The first known account of the Copper Country of Michigan was published at Paris, in 1639 by Lagarde (Drier, 1961). This report was followed by references to copper in reports by Jesuit missionaries. No serious mining was attempted until 1844, after Douglas Houghton's report (1841) was published three years earlier.

This early copper rush, with the attendant boom towns and mushrooming population lasted for little more than a year. At this time the first successful mine, the Cliff mine, was opened in 1845. This and other early producers of copper, such as the Minesota, and Central mines became known as "mass" or "fissure" copper mines, in which the copper was contained in "masses" in steeply dipping veins. The copper produced in the first six years of mining was exclusively from this type of deposit.

The Isle Royale Mining Company began mining the amygdaloidal flow top of what became known as the Isle Royale Amygdaloid or lode. Thereafter, the "fissure" type ore bodies increasingly diminished in importance relative to the other types of copper occurrences.

A third type of host rock was discovered by Edwin Hulbert in 1864 in what became known as the Calumet and Hecla Conglomerate. This became the bonanza lode, which produced approximately forty percent of the native copper in the district.

Hulbert proposed the formation of the Calumet Mining Company of Michigan and the name Hecla for another company to hold lands in the vicinity. In 1866, Alexander Agassiz became treasurer both of Calumet and of Hecla, and in 1867, he became resident superintendent of both mines, after Hulbert was dismissed.

His initial problems of "stamping" the extremely hard conglomerate ore were overcome by ordering the existing milling equipment to be replaced by the steam stamp developed by Thomas Ball. By the end of 1867 the Hecla Mine was the largest producer in the district. It topped Calumet and the best amygdaloid producer, the Pewabic, by more than 50 percent (Benedict, 1952).

The principle producing flow tops and associated mines with the dates of first and last production of the mines are listed in Table 1.

All the principal ore deposits were discovered and being exploited before 1900. The only significant discoveries which were not discovered in, or close to, exposure were the Iroquois Amygdaloid, and the Kingston Conglomerate. Each of these deposits was discovered by drilling.

The copper sulfide deposits in the Nonesuch formation were discovered in outcrop in the 1850's, but not mined until 1865. Only the native copper was recovered; the chalcocite was left in tailings. The old White Pine mine, which originally operated from 1879 to 1881 was purchased by the Calumet and Hecla Mining Company in 1904. Exploration and drilling revealed copper in the vicinity of the present mine site. Three shafts were sunk in 1914, and mining of the native copper continued until 1921. The mine was purchased by Copper Range Company in 1929. Interest in the deposit revived since developments in metallurgy had made the economic recovery of copper from chalcocite feasible in the 1930's and, coupled with the copper shortage created by the outbreak of the Korean War, a development of the ore body was initiated at the request of the Federal Government. The mine was put into a regular production basis in 1953, and by 1965 the one-billionth pound of copper was poured in the smelter.

Post 1950 Developments

The Calumet and Hecla Division drilled the property around the Old Caledonia adit in the Evergreen Series of lava flows in Ontonogan County. This project outlined approximately 11.5 million tons of ore in the Butler and Knowlton lodes. Some underground development was done , and 223,145 tons of hoist were removed through the adit between 1952 and 1958.

In 1965 a major agreement between Copper Range and the Calumet and Hecla Company took place whereby a territorial zone was established. Most of Copper Range's holding northeast of Portage Lake were turned over to C and H, and C and H holdings southwest of Portage Lake were tranferred to Copper Range.

In May 1968, the Calumet and Hecla Mining Company merged with and became the Calumet Division of Universal Oil Products (UOP). The new operator continued to run the mines and took control of all mineral and land holdings. Virtually all copper production from the native copper mines operated by the Calumet Division of Universal Oil Products (UOP) ceased in August 1968 after a labor dispute. Several mines were already operating at a loss and so the decision was made to close down the mines. Most of the surface equipment was liquidated; the smelter and mill at Torch Lake, Michigan, were dismantled; and several of the shafts were sealed. Most mines are filled with water.

TABLE 1
SIGNIFICANT FLOW TOP LODES

Lode	Principle Mines	Production (all mines)
Isle Royale	Isle Royale	1852-1948
Pewabic	Pewabic Franklin Quincy	1856-1945
Ashbed	Atlantic Mine	1872-1906
Kearsarge	North Kearsarge South Kearsarge Ahmeek Mohawk Seneca Peninsula (Allouez) (Calumet and Hecla) (Osceola)	1882-1968
Osceola	Osceola	1891-1968
Baltic	Baltic Champion Trimountain	1898-1968
Evergreen	Adventure Caledonia Mass	1899-1958

SIGNIFICANT CONGLOMERATE LODES

Allouez	Allouez	1869-1968
Calumet and Hecla	Calumet and Hecla	1865-1968
Houghton	Douglass	1929-1968
Kingston	Kingston	1965-1968

SULFIDE PRODUCERS

Nonesuch Shale	White Pine	1945-present
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Mines in parentheses have crosscuts into the Kearsarge Lode.

[TABLE 1: Significant Producers (Lodes and Mines)]

In July 1972, Homestake Copper Company, a subsidiary of Homestake Mining Company, leased the mineral holdings from UOP. Homestake was joined by INCO's subsidiary the American Copper and Nickel Company in 1973 in a joint effort to explore the Centennial property and to investigate new methods of mining and milling. Homestake retained 60% of the venture and acted as manager of the exploration and development program.

The company started dewatering the mine during April 1973, as part of an initial \$5 million program which also included renovation of the No. 3 and No. 6 shafts and surface equipment. Once the initial project was completed, \$4.3 million more was spent on the building of a pilot plant. This plant included a concentrator to mill ore along with other new equipment the company wanted to try out. Information gathered at the pilot plant would be used to establish ore processing costs in addition to serving as a design model for a larger plant.

As of December 1975, the No. 6 shaft was being deepened at a rate of 100 to 130 feet per month. At that time the shaft was nearing 2,042 meters (approximately 1,265 vertical meters), and the goal was to reach the 55th level (2,347 meters) by September 1976.

In November 1976, mine exploration development and production was halted due, in part, to depressed copper prices and to disappointing exploration results. The pumps were pulled in 1977 and the mine is filling with water. Hopefully, this is not the epitaph of Copper Country.

PREVIOUS WORK

Several investigations have been undertaken in an attempt to evaluate the copper resources in Michigan. Because of the "boom" atmosphere and fiercely competitive nature of the large and small companies which operated the mines in Copper Country in the latter part of the nineteenth century, data concerning regional and even local production are somewhat obscure.

The first regional investigation of "Copper Country" took place in the summer of 1840 in an expedition led by Douglas Houghton, newly appointed state geologist by Steven T. Mason, the first governor of Michigan. His report, filed in December of that same year, touched off a copper rush to the Keweenaw Peninsula.

Two papers by More (1912, 1915) are particularly useful for purposes of obtaining an historical account of the then active mines of the various copper lodes. The first exhaustive compilation of the district by Butler and Burbank (1929) became, and still is, one of the most informative references on Copper Country.

Much of the post 1929 data have not been summarized and remain in company files or annual reports of the Michigan Geological Survey. A renewed interest in non-operating mines was generated by the 1965 land exchange between Calumet and Hecla Copper Company and Copper Range. Copper Range acquired

several old properties, specifically the Indiana and Champion, which were re-evaluated by Joseph Patrick of Copper Range. These reports are in White Pine Copper Company's files for the investigations which were completed in 1968. Feasibility studies on properties acquired by C and H from Copper Range are on file at Universal Oil Product's Calumet office.

Geologists of the United States Geological Survey mapped several quadrangles in Copper Country and reported on their economic potential (Cornwall and Wright, 1956; White, et al., 1953; Cornwall, 1954a, 1954b; Davidson, et al., 1955; White, 1956). A particularly useful summary by Cornwall, et al. (1949) reported on the active copper mines in Houghton and Keweenaw Counties at the time of writing.

Miska and Fris (1971) give a very general summary, up to 1968, of all commodities including copper for the Lake Superior East planning subarea of the Great Lakes Basin. More specifically, Bennet, et al. (1973) reported on Michigan reserves as part of an appraisal of copper supply from domestic sources.

An excellent compilation of nonferrous metal potential in northern Michigan by Bodwell (1972) includes a section on the copper deposits.

GEOLOGY

General

The Michigan copper reserves are found in 3 geologically distinct environments. Two of the three environments, the Nonesuch lode and the lodes within the Portage Lake Lava Series, are of middle to upper Keweenaw age (Precambrian Y¹) and underlie the Keweenaw Peninsula of Northern Michigan. The remaining environment is in the Marquette "trough", particularly within the Kona Dolomite of Precambrian X age. The Kona Dolomite is part of the Marquette Supergroup whose younger members are presently mined for their iron ore.

¹Scheme for the subdivision of Precambrian time, adopted by the U. S. Geological Survey:

Precambrian Y = 800 m.y. to 1,600 m.y.
Precambrian X = 1,600 m.y. to 2,500 m.y.
Precambrian W = older than 2,500 m.y.

Marquette Area

The Precambrian W rocks of Michigan are exposed in an east-west trending belt which extends from Marquette west into Wisconsin. The rocks consist predominantly of granitic intrusives (now mostly gneissic) and greenstone volcanic flows. The greenstone belt has intercalations of metasedimentary and felsic pyroclastic rocks. The flows are mostly of mafic composition and the development of pillow structures which can be found in several exposures indicate subaqueous solidification of the basalt. These greenstones, typified by the Mona Schists

in the Marquette region (Plate 1) have a suggested age of 2.77 billion years before present (Goldich, 1972).

The Precambrian X rocks of Michigan in the Marquette region have been collectively referred to as the Marquette Supergroup (Cannon and Gair, 1970). The Supergroup appears to have been deposited and deformed in a similar tectonic setting to the Animikie of Northern Minnesota and S.W. Ontario. The rocks of the Supergroup have been subdivided into four groups: the Chocoma, Menominee, Baraga and Paint River Groups (see Plate 1).

The lowermost Chocoma Group has characteristics in common with a modern miogeosynclinal or shelf sea facies association. The Kona Formation (maximum thickness 800 m) is a mixture of carbonates, slates and quartzites and contains algal structures similar to those found in present day intertidal zones. The overlying Menominee Group has been suggested to be a transgressive sequence (Tyler and Twenhofel, 1952). The Negaunee Iron Formation is the uppermost unit in this Group. Extensive erosion of this formation serves as a break between the Menominee and overlying rocks of the Baraga Group.

The youngest unit within the Group, the Michigamme Formation, covers a large area of the Upper Peninsula. The formation has been thought of as a deep water sequence containing some turbidites and mafic volcanics. Cannon (1973) has called the formation eugeosynclinal.

The Paint River Group is exposed in the Iron River - Crystal Falls district of Michigan. It is a series of slates, graywackes, an iron formation.

The Michigan Copper District

The native-copper deposits in the middle Keweenaw Portage Lake Lava Series occur in flow tops and Interflow conglomerate beds (Figure 1). More than 96 percent of the copper produced from this formation has come from an area 28 miles long (White, 1968) which extends just southwest of Painesdale to just east of Mohawk (Figure 2, Plates 2, 3, 4&5). Minor production has come from the Mass-Victoria area and from small mines northeast of Mohawk.

Copper sulfides with minor amounts of native copper occur at the base of the Nonesuch Shale, and within the upper few feet of the underlying Copper Harbor Conglomerate. Although traces of copper can be found at this contact over an area at least 125 miles long (White, 1971), only two deposits of significant copper concentration have been found so far. These are the White Pine orebody and the synclinal area west of the Porcupine Mountains known as the Western Syncline or Presque Isle Syncline.

The stratigraphy in the copper district is as follows:

Top: Not exposed
 Freda Sandstone, >12,000'
 Nonesuch Shale, 250-750'
 Copper Harbor Conglomerate, 200-7,000'
 Unnamed Formation, 0-8,000'
 Portage Lake Lava Series, >15,000
 Base: Not exposed

The base of the Nonesuch Shale has been selected as the boundary between upper and middle Keweenaw rocks (White, 1972). The stratigraphic section below the base of the Nonesuch Shale is given in Figure 3.

Copper Deposition in the Marquette Supergroup

Taylor (1972) suggests that the Kona sediments were deposited in a stable shelf, lagoonal environment. The sulfides, mainly chalcocite, chalcopyrite, bornite and pyrite are found disseminated within three beds with a total thickness of 56 feet. The beds are approximately 200 feet above the base of the formation. The three beds, an upper argillite, middle quartzite, and lower argillite are not mineralized in equal amounts; mineralization is most intense at the argillite contacts.

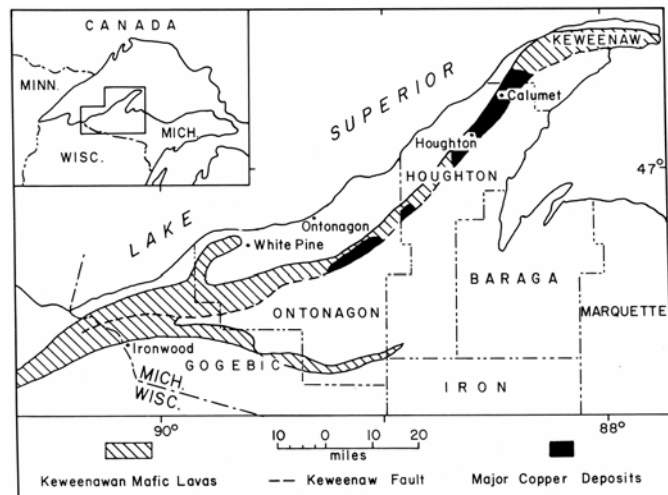


Figure 1

[FIGURE 1: Distribution of Keweenaw Lavas with Major Native Copper Deposits]

The genesis of the sulfides has been summarized by Taylor (1972) and Clark (1974). Clark (p. 31) suggests that copper solutions moved through the middle quartzite bed and into the enclosing argillite and replaced preexisting (syngenetic?) pyrite. Although the source of the solution has not been defined, Clark suggests they may have migrated updip from offshore pelagic sediments.

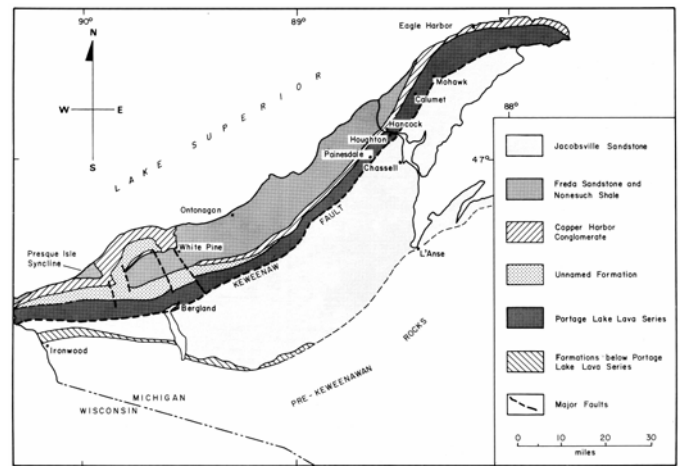


Figure 2

[FIGURE 2: Keweenaw Rocks of Michigan]

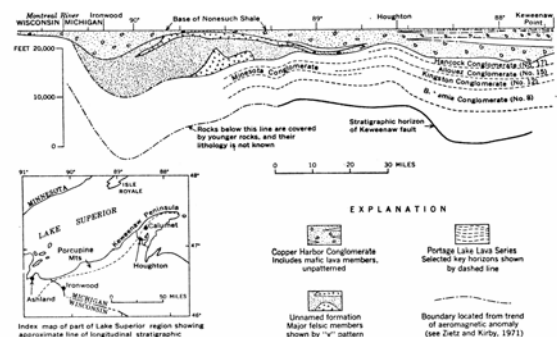


FIGURE 3.—Longitudinal stratigraphic section of Keweenaw rocks below the base of the Nonesuch Shale from Keweenaw Point to the Montreal River. All data projected to an east-west line. Dashed lines within the Portage Lake Lava Series represent individual horizons that can be traced with confidence between the indicated points, largely by correlation of drill-hole sections. Thickness east of long. 89°15' from sources listed in figure 1 and from Butler, Burbank, and others (1929, pl. 15); thickness west of long. 89°15' calculated from unpublished maps by E. R. Brooks, H. A. Hubbard, R. F. Johnson, W. S. White, J. T. Wilband, and J. C. Wright.

From White, 1972.

[FIGURE 3: Stratigraphic Section of Keweenaw Rocks Below the Base of the Nonesuch Shale]

Distribution and Genesis of Native Copper Ores

The copper mineralization within the Portage Lake Lava Series was contemporaneous with, or later than, much or most of the deformation (White, 1968). This deformation is represented by minor folding, tilting, and high-angle faulting. Where mineralized, these last features formed the so-called fissure or vein lodes. White (1968, op. cit.) notes that the dips in individual horizons of the Lava Series tend to steepen 20 to 30° along strike from Mohawk to Painesdale (Figure 1). This variation, he states, creates a gentle twist in each horizon which may be a significant factor since most of the copper production has come from this area.

The forms of the orebodies have been described in detail by White (1968), Butler and Burbank (1929), Cornwall, et al. (1949), Broderick, et al. (1946), and Weege, et al. (1972). The following summary draws heavily upon their investigations.

The amygdaloidal lodes have irregular copper distributions since several flows stratigraphically above one another can be mineralized up to depths of 40 feet

in fragmental portions. The flow tops are not continuously fragmental so that lateral pinchouts in the ore are very common. This erratic distribution causes great difficulty in assigning a meaningful average thickness to many ore deposits. Stope heights vary from mine to mine; ten to fifteen feet are common.

Conglomerate orebodies often defy the most comprehensive development plans to best follow the ore. In the Kingston ore-body, for example, native copper may occur at any place within the 40-foot-thick conglomerate. In general, it occurs within roughly restricted zones designated as hanging wall, footwall, and intermediate ore (Weege, et al., 1972). The segregation of the ore zones is not as obvious at the Centennial No. 3-6 orebody; however, because the conglomerate averages approximately 8 feet, the entire bed is usually mined. Considerable time and effort have been expended by Homestake geologists to ascertain the controls of copper distribution at the Centennial No. 3-6 mine. The variation of copper between channel sand and conglomerates, tributaries, and silt lenses provide empirical ore guides which appear to be related to permeability of the members and perhaps to differential compaction during burial and deformation.

The ultimate source of the copper and the paucity of sulfur is still not fully explained. Most investigators agree, however, that the native copper is epigenetic. The ore has been attributed to downward percolating meteoric waters, magmatic hydrothermal solutions, and to upward migrating copper solutions, probably as brines, which derived their copper by a dehydrating metamorphism from the lavas themselves. This last hypothesis appears to be the most favored, particularly in the light of recent geochemical investigations by Jolly (1974). The low amount of sulfur is more easily explained by this hypothesis since normal magmatic hydrothermal solutions would be expected to form copper sulfides.

Genesis of the White Pine Orebody

The origin of the copper mineralization at White Pine and the Presque Isle syncline is not fully known. At White Pine the geologists recognize that the characteristics of the orebody can be fitted into several models. In fact, Robert Seasor, Chief Geologist at the mine, asks visiting geologists whether they wish to take the "syngenetic or the epigenetic mine tour".

Ensign, et al. (1968) have summarized the problems of both genetic hypotheses. More recently, Brown (1974) has experimentally duplicated the trace element pattern of Cd, Pb, and Zn which is found at the White Pine orebody. These elements are found in a narrow blanket-like zone which directly overlies the cupriferous zone and transects the bedding at gentle to moderate angles. They are also found in a second zone which is strictly stratabound associated with a pyritic bed some 60 feet above the Nonesuch base. Brown (1971) attributes this first zone to copper chloride brines which ascended

through the Nonesuch strata replacing preexisting pyrite, and sweeping the more soluble Cd, Pb, and Zn before the copper front. The transition between the cupriferous zone below and the unreplaced pyritic zone above is thought to represent the position where the copper solutions were finally spent.

The overlying trace concentrations have been interpreted as syngenetic, however, Brown (1974) interprets this as a temporary but sudden pulsation of the ore solution, perhaps due to some tectonic readjustment. Brown, under laboratory conditions, demonstrated that the Cd, Pb, and Zn could be swept much higher than the defined zone set by the observed "fringe" or "transition" zone.

White (1971) has suggested the Copper Harbor Conglomerate as a potential source of the copper. He presents calculations that show the conglomerate could have contained a sufficient amount of water released by compaction to account for the White Pine ore-body, assuming a copper concentration of 50 ppm. This model requires a load of 3,000 to 5,000 meters of Freda Sandstone above the Nonesuch to compact the Copper Harbor. White alternatively suggests that surface waters could have entered an exposed portion of the Copper Harbor on the northern limb of the syncline and migrated to the White Pine area, so long as a suitable hydraulic gradient existed between the recharge area and the water table at the orebody during the mineralization event.

PRODUCTION

With the closing of the joint experimental mining and milling venture by the American Nickel Company and Homestake Copper Co-pany at the Centennial 3-6 Mine in 1976 there has been no copper production from "Copper Country" since. The price of copper reached all time highs in 1974 just before the rehabilitated mine went into production. When the mine went into production in 1975 the price of copper plummeted, and the grade, which was expected to increase at depth, remained disappointingly low as the shaft was deepened from the old 38th to beyond the 50th level. The financially unsuccessful project was closed, the pumps removed, and the mine is flooding.

The refined copper production from the native copper mines (Figure 4) spans a period of 130 years in which the amount of copper produced rose almost exponentially with time from 1845 to World War I as new mines were brought on line. The "boom and bust" production of two World Wars and the effects of the depression, with a spurt during World War II reflect the economic turmoil of the times. The gradual overall decay in production from the peak production years of 1915 to 1918 reflects the fact that no new native copper lodes were developed that could match or even sustain the production of the older lodes which were slowly being mined out.

The White Pine orebody was brought into production in 1953 (see Figure 5) and has produced over 2.5 billion pounds of copper (including 1978 production). The copper produced at White Pine is already 23 percent of the total amount produced for all the native copper mines. Although the White Pine Company mines copper sulphide and native copper, the production figures (Table 4) for this orebody are listed under "Pounds Sulphide Copper" to differentiate this ore from "Copper Country" native copper ores.

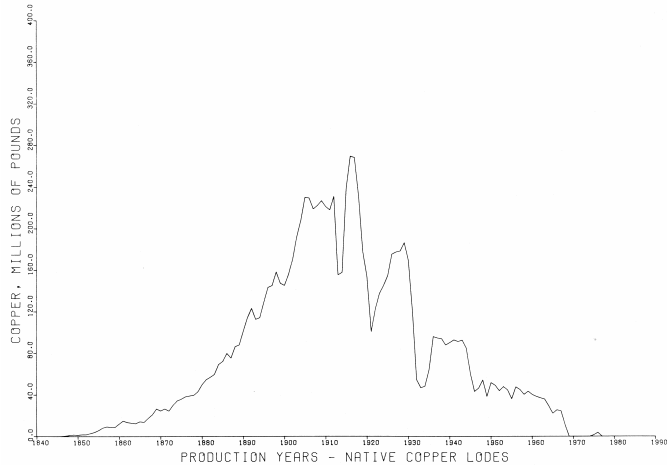


Figure 4

[FIGURE 4: Copper Production - Native Copper Lodes]

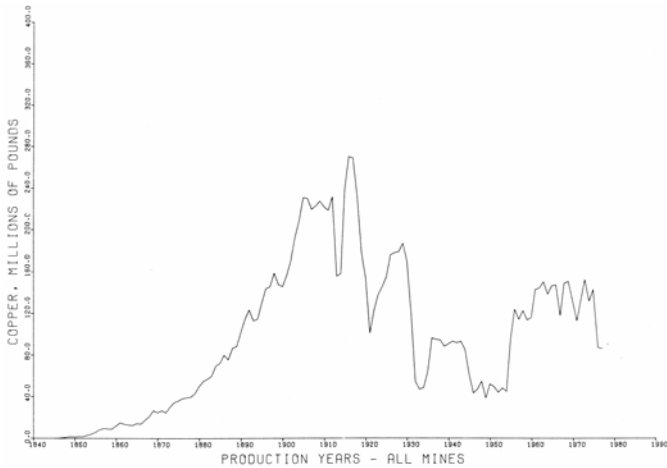


Figure 5

[FIGURE 5: Copper Production - All Mines]

TABLE 2
REFINED COPPER PRODUCTION (POUNDS) FROM ALL LODES

	1845-1925	1926-1977	LODE TOTAL	% of NATIVE COPPER LODES
Algoma amygdaloid	12,467	112,368 ^b	124,835	
Allouez conglomerate	69,521,253	4,454,284 ^b	73,975,537	
Arcadian amygdaloid	4,783,711	22,430	4,806,141	
New Arcadian amygdaloid	164,794	0	164,794	
Ashbed +Atlantic amygdaloid	142,840,804	6,219,555 ^b	149,060,359	1.42
Baltic (+Superior) amygdaloid	873,525,912	427,942,546 ^b	1,301,468,458	12.37
Calico amygdaloid	8,006,420	0	8,006,420	
Calumet and Hecla conglomerate	3,375,361,468 ^b	843,173,416 ^b	4,218,534,884	40.08
Evergreen series	66,508,152	5,565,470	72,073,622	
Houghton cgl. & Iroquois anyg.	0	86,022,420	86,022,420	
Ile Royale amygdaloid	210,839,585	130,550,073	341,389,658	3.24
Kearsarge amygdaloid	1,176,948,378	1,094,869,382 ^b	2,271,817,760	21.58
Kingston conglomerate	0	19,959,554	19,959,554	
Lake amygdaloid	7,326,227	0	7,326,227	
"Nasau" epidote	84,095	0	84,095	
Nonesuch	18,622,725	2,455,666,905 ^b	2,474,289,630	
Oceola amygdaloid	416,398,662	179,713,337 ^b	596,111,999	5.66
Pewabic amygdaloid	873,712,184	206,916,809	1,080,628,993	10.27
Fissures ^a	199,853,050	3,487,180	203,340,230	1.93
Not designated	71,015,685	24,858	71,040,543	
Mines on Ile Royale	978,625	0	978,625	
TOTALS	7,516,504,197	5,464,700,587	12,981,204,784	
		Native Cu	10,525,378,879	96.50

a: includes 230,897 pounds mohavite from Mohawk Mine.
b: includes reclamation of sands and slimes.

[TABLE 2: Refined Copper Production from all Lodes]

[TABLE 3: Production of Refined Copper by Lode and Mine]

PRODUCTION OF REFINED COPPER BY LODE AND MINE

TABLE 3A

ALLOUEZ CONGLOMERATE

Mines	Pre 1926	Post 1926	Total Pounds
Albany and Boston (1865-1878)	865,057	0	865,057
Allouez (1969-1892)	25,786,651	2,788,411	28,575,062
Delaware (1882-1885)	1,770,570	0	1,770,570
Franklin Jr. (1901-1919)	34,473,984	0	34,473,984
Peninsula (1883-1892)	6,624,991	0	6,624,991
SUBTOTAL	69,521,253	2,788,411	72,309,664
Miscellaneous			
Allouez (rock pile & sands)	0	1,665,873	
TOTAL	69,521,253	4,454,284	
GRAND TOTAL	73,975,537		

[A: Allouez Conglomerate]

TABLE 3B

BALTIC AMYGDALOID

Mines	Pre 1926	Post 1926	Total Pounds
Baltic (1898-1916; 1917-1925)	236,523,711	20,004,300	283,528,011
Champion (1902-1925)	438,403,286	384,176,397	822,579,683
Trimountain (1902-1925)	141,040,592	7,759,077	148,799,669
Superior (1908-1920)	30,018,271	0	30,018,271
Houghton (1915-1917)	540,052	0	540,052
SUBTOTAL	873,525,912	411,939,774	1,285,465,686
Miscellaneous			
Reclamations (sands)	0	16,002,772	
TOTAL	873,525,912	427,942,546	
GRAND TOTAL	1,301,468,458		

[B: Baltic Amygdaloid]

TABLE 3C
CALUMET AND HECLA CONGLOMERATE

Mines	Pre 1926	Post 1926	Total Pounds
Calumet and Hecla (1865-1925)	2,831,092,153	449,336,193	3,280,428,346
Centennial (1882; 1897-1898)	202,744	24,524,556	24,727,294
Osceola (1874-1880; 1897-1903)	25,056,028	0	25,056,028
Schoolcraft (1869-1874)	1,882,529	0	1,882,529
Tamarack (1882-1883; 1885-1917)	379,978,899	0	379,978,899
Tamarack Jr. (1892-1896)	9,462,191	0	9,462,191
SUBTOTALS	3,247,674,544	473,860,749	3,721,535,287
Miscellaneous			
Centennial #6 ventilation raise		240,642	
Centennial #2 rockpile		39,795	
Reclamation (1915-1925)	127,686,924	368,981,540	
Tamarack Jr. rockpile		40,814	
Laurium rockpile		9,882	
TOTAL	3,375,361,468	843,173,416	
GRAND TOTAL	4,218,534,884		

[C: Calumet and Hecla Conglomerate]

TABLE 3D
KEARSARGE AMYGDALOID

Mines	Pre 1926	Post 1926	Total Pounds
Ahmeek (1904-1923)	258,287,424	433,234,213	691,521,637
Allouez (1905-1923)	82,222,989	0	82,222,989
Calumet and Hecla (1904-1925)	67,351,569	0	67,351,569
Centennial (1900-1923)	33,033,675	127,546,619	160,580,294
Douglass	0	52,393,311	52,393,311
Gratiot (1910-1920)	313,848	0	313,848
Kearsarge (1888-1896)	14,512,073	0	14,512,073
LaSalle (1910-1920)	7,814,730	0	7,814,730
Mohawk (1902-1925)	259,114,348	116,802,708	375,917,056
Ojibway (1911)	49,662	201,203	250,865
Osceola (1897-1920)	261,527,694	0	261,527,694
Seneca (1872-1925)	3,863,385	143,149,578	147,012,963
Tecumseh (1906)	59,874	0	59,874
Wolverine (1882-1925)	188,797,107	0	188,797,107
North Kearsarge		71,294,085	71,294,085
South Kearsarge		1,015,963	1,015,963
Peninsula	0	120,050,222	120,050,222
SUBTOTAL	1,176,948,378	1,065,687,902	2,242,636,280
Miscellaneous			
Seneca rockpiles	0	152,416	
Ahmeek rockpiles	0	2,397,540	
Calumet crosscut	0	129,400	
Kearsarge rockbin	0	4,257	
Reclamation	0	26,497,867	
	0	29,181,480	
TOTAL	1,176,948,378	1,094,869,382	
GRAND TOTAL	2,271,817,760		

[D: Kearsarge Amygdaloid]

TABLE 3E
ISLE ROYALE AMYGDALOID

Mines	Pre 1926	Post 1926	Total Pounds
Grand Portage (1863-1884)	3,529,822	0	3,529,822
Huron (1855-1893)	25,309,270	0	25,309,270
Isle Royale (1853-1885; 1901-1925)	180,537,926	130,550,073	311,087,999
Shelden & Columbia (1864-1884)	1,463,336	0	1,463,336
SUBTOTAL	210,839,585	130,550,073	341,389,658
GRAND TOTAL	341,389,658		

[E: Isle Royale Amygdaloid]

TABLE 3F
IROQUOIS AND HOUGHTON LODES

Mines	Pre 1926	Post 1926	Total Pounds
Allouez #3	0	42,960,653	42,960,653
Iroquois	0	43,061,767	43,061,767
TOTAL		86,022,420	86,022,420
GRAND TOTAL	86,022,420		

[F: Iroquois and Houghton Lodes]

TABLE 3G
OSCEOLA AMYGDALOID

Mine	Pre 1926	Post 1926	Total Pounds
Calumet and Hecla (1904-1924)	213,817,584	0	213,817,584
Centennial (1891-1900)	2,556,685	0	2,556,685
Osceola (1879-1920)	190,787,393	114,507,974	305,295,367
Tamarack (1885-1917)	9,237,000	0	9,237,000
SUBTOTAL	416,398,662	114,507,974	530,906,636
Miscellaneous			
Allouez #3 exploration	0	7,862	
Smelter slag	0	3,063	
Reclamation	0	65,194,438	
TOTAL	416,398,662	179,713,337	
GRAND TOTAL	596,111,999		

[G: Osceola Amygdaloid]

TABLE 3H
PEWABIC AMYGDALOID

Mines	Pre 1926	Post 1926	Total Pounds
Albany and Boston (1862-1864)	14,982	0	14,982
Franklin (1857-1919)	119,842,467	0	119,842,467
Pewabic (1855-1884)	27,823,416	0	27,823,416
Quincy (1856-1925)	726,000,319	109,790,547	835,790,866
Rhode Island (1903;1906)	31,000	0	31,000
SUBTOTAL	873,712,184	109,790,547	983,502,731
Miscellaneous			
Quincy reclamation	0	96,635,312	
Franklin sands	0	490,950	
TOTAL	873,712,184	206,916,809	
GRAND TOTAL	1,080,628,993		

[H: Pewabic Amygdaloid]

TABLE 4

MICHIGAN COPPER PRODUCTION

Year	Pounds Native Copper	Pounds Sulphide Copper	Pounds Total Copper	Average Value Per Pound	Value Of Copper
1845	26,880			\$.20096	\$5,400
1846	58,240			.17170	10,000
1847	477,120			.18511	88,320
1848	1,032,640			.19455	200,900
1849	1,505,280			.22321	336,000
1850	1,281,280			.22321	286,000
1851	1,744,960			.16591	289,500
1852	1,774,080			.22321	396,000
1853	2,905,280			.22321	648,500
1854	4,074,560			.22321	909,500
1855	5,808,320			.27304	1,585,904
1856	8,211,840			.26995	2,216,786
1857	9,531,200			.24998	2,382,609
1858	9,157,120			.23245	2,128,573
1859	8,926,400			.21821	1,947,830
1860	12,069,120			.21999	2,655,086
1861	15,037,120			.22973	3,454,478
1862	13,585,600			.26749	3,634,012
1863	12,985,280			.34004	4,415,515
1864	12,490,240			.46993	5,869,538
1865	14,358,400			.39248	5,635,385
1866	13,749,120			.33668	4,629,054
1867	17,525,760			.25365	4,445,409
1868	20,935,040			.23600	4,940,669
1869	26,624,640			.23399	6,229,900
1870	24,622,080			.20699	5,096,524
1871	26,750,080			.22250	5,951,893
1872	24,552,640			.32498	7,979,117
1873	30,089,920			.28807	8,668,003
1874	34,332,480			.23328	8,009,081
1875	36,039,360			.22899	8,180,574
1876	38,270,400			.20899	7,998,131
1877	39,025,280			.18777	7,327,777
1878	39,690,560			.16601	6,589,030
1879	42,848,960			.17172	7,358,023
1880	49,736,960			.20008	9,951,371
1881	54,573,120			.18280	9,975,966
1882	56,982,765			.18410	10,490,527
1883	59,702,404			.15842	9,458,055
1884	69,353,202			.13690	9,494,453
1885	72,148,172			.11809	7,942,792
1886	79,890,798			.10861	8,676,940
1887	75,471,890			.11220	8,467,046
1888	86,472,034			.16780	14,510,007
1889	88,175,675			.13490	11,894,899

Michigan Copper Production Contd.

Year	Pounds Native Copper	Pounds Sulphide Copper	Pounds Total Copper	Average Value Per Pound	Value Of Copper
1940	90,402,594	--	--	\$.11296	\$10,211,877
1941	92,864,678	--	--	.11797	10,886,823
1942	91,365,660	--	--	.11775	10,758,306
1943	92,736,686	--	--	.11775	10,833,000
1944	84,845,078	--	--	.11775	9,104,913
1945	60,801,716	--	--	.11775	7,159,402
1946	43,203,563	2,741	43,206,304	.13820	5,971,111
1947	46,581,000	473,473	47,054,473	.20958	9,861,676
1948	54,309,661	124,381	54,434,042	.22038	11,990,174
1949	38,498,167	--	38,498,167	.19202	7,392,418
1950	51,788,017	--	51,788,017	.21235	10,997,185
1951	49,297,405	459	49,297,864	.24200	11,930,083
1952	43,939,304	--	43,939,304	.24200	10,633,312
1953	47,999,690	--	47,999,690	.28798	13,822,951
1954	44,796,824	--	44,796,824	.29694	13,298,993
1955	36,265,073	58,643,499	95,208,572	.37491	35,690,374
1956	47,649,651	75,699,096	123,348,747	.41818	51,553,329
1957	68,562,019	68,562,019	137,124,038	.29576	36,612,279
1958	40,402,115	81,656,908	122,059,023	.25764	31,447,287
1959	43,434,301	69,647,232	113,081,533	.31182	35,261,084
1960	40,314,264	75,120,243	115,434,507	.32053	37,001,516
1961	38,517,943	103,756,178	142,274,121	.29921	42,572,434
1962	37,147,227	106,613,451	143,760,678	.30600	43,990,767
1963	35,880,333	113,900,191	149,780,524	.30600	45,832,840
1964	29,326,043	108,458,751	137,784,794	.31960	44,036,020
1965	22,174,047	123,743,392	145,917,439	.35017	51,095,910
1966	25,562,323	121,181,717	146,744,040	.36170	53,077,319
1967	24,458,983	92,731,257	117,190,240	.38226	44,797,141
1968	11,288,306	137,092,516	148,380,822	.42247	62,686,446
1969	--	149,952,283	149,952,283	.47534	71,278,318
1970	--	131,373,570	131,373,570	.57700	75,802,550
1971	--	112,155,147	112,155,147	.52000	58,320,676
1972	--	131,937,378	131,937,378	.50617	66,782,743
1973	--	151,678,939	151,678,939	.62100	94,192,621
1974	--	131,037,602	131,037,602	.83170	108,983,973
1975	1,455,651	140,819,850	142,275,501	.58000	82,519,791
1976	3,866,218	83,044,081	86,910,299	.68624	59,641,323
1977	--	85,960,551	85,960,551	.68770	59,115,071
TOTAL	10,529,198,135	2,455,666,905	12,984,865,040		

Data in italics are corrections from original file reported in unpublished updates by Hardenburg (see text).

[TABLE 4: Annual Copper Production]

Michigan Copper Production Contd.

Year	Pounds Native Copper	Pounds Sulphide Copper	Pounds Total Copper	Average Value Per Pound	Value Of Copper
1890	101,410,277			\$.15609	\$15,820,003
1891	114,222,709			.12760	14,574,818
1892	123,198,460			.10091	12,431,957
1893	112,605,078			.10750	12,105,046
1894	114,308,870			.09494	10,852,484
1895	129,330,749			.10730	13,877,189
1896	143,524,069			.11093	15,921,125
1897	145,282,059			.11584	16,829,474
1898	158,491,703			.12050	19,090,250
1899	147,400,338			.17760	26,178,300
1900	145,461,498			.16667	24,046,068
1901	156,289,481			.16722	26,134,727
1902	170,609,228			.12160	20,746,082
1903	192,400,577			.13720	26,397,359
1904	208,309,130			.13010	27,101,018
1905	230,287,992			.15890	36,592,762
1906	229,695,730			.18745	43,056,465
1907	219,131,503			.20045	43,924,910
1908	222,289,584			.13580	30,186,926
1909	227,005,923			.13480	30,600,398
1910	221,462,984			.13125	29,067,017
1911	218,185,236			.12376	27,002,605
1912	231,112,228			.16341	37,766,049
1913	155,715,286			.15269	23,776,167
1914	158,009,748			.13602	21,492,486
1915	238,956,411			.17275	41,279,720
1916	269,794,531			.27202	73,389,508
1917	268,508,098			.27180	72,980,501
1918	231,096,150			.24628	56,914,360
1919	177,594,135			.18691	33,194,120
1920	153,483,952			.17456	26,792,159
1921	100,918,001			.12502	12,616,768
1922	122,545,126			.13382	16,398,989
1923	137,691,306			.14421	19,856,463
1924	145,333,227			.13024	18,928,199
1925	154,799,247	(TOTAL 7,521,064,924)		.14042	21,736,910
1926	175,441,565			.13795	24,202,164
1927	177,537,175			.12920	22,937,881
1928	178,442,704			.14570	25,999,102
1929	186,393,974			.18107	33,750,357
1930	169,297,775			.12982	21,978,237
1931	118,495,055			.08116	9,617,059
1932	54,396,108			.05555	3,021,704
1933	46,853,130			.07025	3,291,432
1934	48,215,859			.08428	4,063,633
1935	64,107,889			.08649	5,544,691
1936	95,968,019			.09174	9,092,010
1937	94,649,805			.13167	12,466,540
1938	94,075,588			.10000	9,407,559
1939	87,969,153			.10965	9,639,897

Copper

This section presents copper production data by lode (Table 2), lodes and mine (Table 3), and by year (Table 4). Annual production by mines and reclamations are presented only for those which operated from 1926 to 1977 (see Tables In Appendix A) to avoid redundancy since Butler and Burbank (1929) have presented a detailed account of Michigan copper district production up to and including 1925. Post 1925 data are drawn heavily from unpublished files (tax files, at the Michigan Geological Survey annual production sheets, etc.) and especially from the unpublished files of Harold Hardenburg (geologist - retired, Michigan Geological Survey). Production data for 1844 to 1925 are largely drawn from Butler and Burbank (1929).

Total refined copper production by lode is presented In Table 2. The grand total for the district is subdivided into 1845-1925 and 1926-1977 production periods for each lode. The 1845-1925 sum of 7,516,504,197 pounds reported by Butler and Burbank (1929, p. 69) is used to maintain continuity with their report. This value differs from that reported in their summary table (7,504,265,411 pounds; 1929, p. 64) and from the high value (7,521,064,024 pounds) reported in the Michigan Geological Survey files (Table 4, "Pounds Native Copper" column). It is not clear why these discrepancies exist.

The original data in Table 4 have been modified for 1939, 1941, 1943, 1944, 1954-1957 inclusive, and 1960-1961. The updated figures (Italicized) were obtained from Hardenburg's data which I interpret from his notes to be more desirable than those of the compilation.

Nevertheless, total amount of copper produced during the 1926-1977 period is 900,471 pounds higher than the corresponding summation in Table 1. The 1926-1977 production data in Table 2 represent summations of production by mines in Tables presented in Appendix A.

Assignment of copper recovered from tailings or "stamp sands" is based on Michigan Geological Survey records. These data are listed in Table 5.

TABLE 5

COPPER RECOVERED FROM TAILINGS AND SLIMES		
Allouez	C & H (1955, 1957, 1960-1962)	1,056,250
Calumet and Hecla		
Lake Linden sands	C & H (1915-1925)	121,007,924
Tamarack sands	(1926-1930, 1935-1943)	261,474,764
	(1925)	6,679,000
	(1926-1930, 1937-1943)	17,405,497
Kearsarge		
Lake Linden sands	C & H (1944-1947)	26,497,867
Tamarack sands	(1947-1967)	65,194,337
Atlantic	Copper Range (1963-1967)	6,116,719
Baltic	Copper Range (1937-1947, 1956-1964)	16,002,772
Pewabic	Quincy (1944-1967)	<u>96,635,312</u>
		618,070,432

[TABLE 5: Copper Recovered from Tailings and Slimes]

TABLE 6
MICHIGAN SILVER PRODUCTION

	<u>Troy Oz.</u>	<u>Value</u>
1883		\$ 173
1884		2,000
1885		2,879
1886		4,654
1887		
1888		5,331
1889		9,655
1890		
1891	73,000	72,270
1892	65,600	57,072
1893	43,500	33,930
1894		
1895		1,373
1896-1909	--	--
1910	262,200	141,600
1911	507,700	274,100
1912	528,453	324,999
1913	295,173	178,284
1914	413,500	228,665
1915	585,933	297,068
1916	--	247,485
1917	759,068	499,300
1918	473,342	504,480
1919	425,610	477,054
1920	510,601	518,326
1921	310,727	310,727
1922	361,912	361,912
1923	253,705	208,038
1924	153,201	102,951
1925	139,499	96,812
1926	107,094	66,827
1927	97,193	55,109
1928	17,152	10,035
1929	17,899	11,084
1930	7,820	3,011
1931	2,024	587
1932	71,408	20,137
1933	125,926	44,074
1934	529	342
1935	4,219	3,032
1936	--	--
1937	25,454	19,689
1938	93,634	60,531
1939	101,878	69,154
1940	88,657	63,045
1941	60,796	43,233
1942	61,674	43,857

TABLE 6 (continued)

	<u>Troy Oz.</u>	<u>Value</u>
1943	48,479	34,474
1944	54,218	38,555
1945	21,863	15,547
1946	--	--
1947	3,089	2,796
1948	--	--
1949	--	--
1950	--	--
1951	--	--
1952	--	--
1953	--	--
1954	--	--
1955	--	--
1956	365,960	--
1957	430,114	--
1958	--	--
1959	--	--
1960	--	--
1961	--	--
1962	401,491	435,618
1963	244,773	317,945
1964	349,195	451,509
1965	457,851	592,000
1966	473,266	611,933
1967	301,992	468,088
1968	472,813	1,013,995
1969	744,290	1,332,778
1970	891,579	1,578,826
1971	629,162	989,136
1972	699,160	1,177,770
1973	654,175	2,175,000
1974	642,944	
1975	628,270	
1976	461,971	
1977	446,808	

[TABLE 6: Silver Production]

Silver Production

Silver production has not been carefully monitored for Keweenaw Copper mines because most of the silver eventually became part of the refined product. This product, known as "Lake Copper" is characteristic of the copper refined from the Nonesuch ores at White Pine and the native copper ores. Silver production data (Table 6) are from data supplied by the Michigan Geological Survey.

Present production of silver is limited to the White Pine orebody. The silver is only partially recovered as a by-product and from 12 to 14 ounces of silver per refined ton of "Lake Copper" go to White Pine's customers.

The silver is scavenged by a special silver circuit in the White Pine mill which serves the dual purpose of: (1) removing excess silver for which White Pine is not compensated for, or which may render the copper unsuitable to a customer's specifications, and (2) providing a silver concentrate which can be added to the copper concentrate in the smelter for special orders which require a higher than normal "Lake Copper" silver content. Present policy at White Pine Copper Company is to ship the excess silver concentrate to an offsite smelter.

RESERVE ESTIMATIONS

The reserve estimations presented in Table 7 were derived from several sources. Most useful were the files of the Calumet Division of Universal Oil Products in Calumet, and information and company reports filed with the Michigan Geological Survey. Without permission to peruse certain files from Industry and Survey personnel, the figures presented in this section would be highly speculative. The White Pine reserves, when released to the writer, were the latest figures available (December, 1976).

Reserve estimates for the strata bound Nonesuch lode are considered to be highly reliable. The reserve values and grades reported for native copper orebodies are not, however, reported with a high degree of confidence. During the course of this investigation it has become very apparent through discussions with mine geologists, mining engineers, and miners of the native copper ores that grade estimation in an operating mine is an extremely difficult task because of the erratic distribution of the copper mineralization.

Attempts to delimit the ore by drilling are frequently frustrated because the drill may sample a lean portion of the lode less than a meter from highly enriched ore. In several mines exploration was conducted, not by drilling, but by drifting through the lode. Although several experienced geologists and mine captains could make good grade approximations, it was more common to find the grade of ore after it had been stamped. As mining progressed, the pattern of ore distribution in developing

areas was then established by the bulk assays from the mined out stopes.

The tonnage and grade estimates for the deposits are rated alphabetically for convenience, rather than attempt to classify their accuracy according to methods established for rating data from other mining camps. Data rated A are considered reliable on the basis of drilling or other methods of sampling where ore has been blocked out, remains as pillar, or was left in stopes. In most cases the values are probably conservative estimates of the tonnage.

The estimates rated B are viewed with less confidence. For example, the Presque Isle deposit which was drilled by Amax cannot, be considered to be outlined in grade and tonnage with the same level of confidence as its operational counterpart at White Pine.

TABLE 7
IN SITU COPPER RESERVES (THOUSANDS OF TONNES) AND GRADE

NATIVE COPPER LODES (Reserves=Grade)	MINE	RESERVES	GRADE (%)	RATING ¹	THOUSANDS OF POUNDS OF COPPER
ALLOUEZ (1,000-1.0)	Allouez	1,000.0	1.0	C	22,046
BALTIC (1,506.5-0.80)	Baltic Champion Trimountain	100.0 1,402.6 3.9	0.5 0.82 1.89	B A A	1,102 25,356 163
CALUMET and HECLA (2,976.5-1.18)	C & H Centennial #2 Centennial #3-6	1,000.0 345.1 1,651.4	1.0 0.95 1.34	C A A	22,046 7,228 48,195
EVERGREEN SERIES (11,421.5-1.11)	Caledonia	11,421.5	1.11	A	279,499
HOUGHTON-IROQUOIS (655.2-0.74)	Allouez #3 Iroquois	576.2 74.3	0.76 0.61	A A	9,654 1,070
ISLE ROYALE (10,000-1.0)	Isle Royale	453.6	1.0	A	10,000
KEARSARGE (8,785.5-0.69)	Seneca & Allouez & Peninsula Mohawk North Kearsarge	6,336.7 429.6 2,019.2	0.625 1.4 0.76	A A A	87,313 13,259 33,832
KINGSTON (4,372.1-1.25)	Kingston	4,372.1	1.25	A	120,485
PEWABIC (5,000-1.0)	Quincy	5,000.0	1.0	C	110,231
OSCEOLA (10,297.8-0.95)	Osceola #6-13	10,297.8	0.95	A	215,676
OTHERS (1,500-0.63)	Indiana Winona	1,000.0 500.0	0.5 0.9	C C	11,023 9,921
	TOTAL	47,969.0			1,028,101

TABLE 7 (CONTINUED)

SULFIDE LODES	MINE	RESERVES	GRADE (%)	RATING ¹	THOUSANDS OF POUNDS OF COPPER
NONESUCH (433,075.5-1.25)	White Pine Presque Isle	337,821.1 95,254.4	1.24 1.27	A B	9,235,124 2,667,000
	TOTAL	433,075.5			11,902,124
KONA DOLOMITE (1,000,000-0.3)	Kona	1,000,000.0	0.3	C	6,613,867
	GRAND TOTAL	1,481,044.5			19,544,092

¹ A: Data believed to be reliable and conservative.

B: Speculative.

C: Highly speculative.

[TABLE 7: In Situ Copper Reserves]

The C-rated deposits are those for which only minimal tonnage and/or grade data were available. In most cases the geologic setting of these deposits are well-described and it is possible to make extrapolations down and along dip using "district norms" (if, indeed, the district has a norm). Some are little more than explored prospects, such as the Indiana mine. Others, such as the Quincy mine, one of the main producers in the district have become legends without a tonnage. Virtually every person who knows "Copper Country" will attest to the fact that there was still a lot of ore left underground when the Quincy closed. How much is left? The answer, if available, has not been found in the copious amount of literature written about the mine. Therefore, the values presented for the Quincy in Table 7 are very speculative. Table 8 summarizes the reserve estimates in each category.

The Kona Dolomite in Marquette County has been vigorously explored. The low grade and poor grindability has made it somewhat less attractive at present copper prices and marketability. The tonnage reported was derived by planimetric analysis of maps of the mineralized zones in theses by Taylor (1972) and Clark (1974), and from other sources.

The percentage of extractable ore cannot be ascertained from the data available except for the White Pine ore body where the insitu reserves (372,384,000 tons at 24.89 lbs/ton) and extractable reserves (212,260,000 tons at 22.66 lbs/ton) are quoted. The present extractable reserve model assumes 57% of the ore remains as pillars. This value (57%) may change as the mine progressively deepens.

Approximately 12 billion pounds of copper are projected for the Nonesuch lode (Table 7). These reserves surpass the 130 year production figure for all the native copper mines. On the other hand, the native copper district reserves are a conservative 1 billion pounds of copper. Stated otherwise, this is the approximate equivalent of what was mined from the Pewabic Lode (Table 2). Most of the reserves are from the Kingston, Osceola, and Caledonia mines, and from the Kearsarge lode.

Silver reserves cannot be estimated for the native copper deposits because of lack of data. The Nonesuch reserves (Table 8) were obtained from the silver values reported by Robert Seasor (personal communication, 1977) for the White Pine orebody. These range from 0.16 to 0.2 troy oz. per short ton of ore. The extractable silver reserves for the White Pine mine are assumed to contain 0.16 oz. per short ton, or approximately 33.9 million oz. In situ reserves, estimated to contain up to 0.2 oz. per ton, would be approximately 74.5 million oz. This total could be increased by 16.8 million oz. to 91.3 million oz. If the 105 million tons of copper bearing shale at the Presque Isle syncline contained .16 oz. of silver per ton. Silver values for the Presque Isle are not available.

TABLE 8
SUMMARY OF INSITU RESERVES (ALL WEIGHTS IN THOUSANDS OF UNITS)
Values in parentheses are exclusive of Kona

Rating	Tonnes	Grade (\$Cu)
A	377,190.1	1.21
B	95,345.4	1.27
C	1,008,500.0	0.30
	(8,500.0)	(0.91)
TOTAL	1,481,044.4	0.60

	Tonnes	Grade	Percent	Pounds	Percent	Oz. Ag.
NATIVE COPPER LODS	47,969.0	0.97	3.2 (= 10)	1,028,099	5.3 (8)	unknown
NONESUCH	433,075.5	1.25	29.2 (= 90)	11,902,124	60.9 (92)	76381.4
KONA	1,000,000.0	0.3	67.5	6,613,867	33.8	unknown
TOTAL	1,481,044.5	0.6		19,544,090		
	(481,044.5)					

[TABLE 8: Summary of Reserves by Ratings]

COSTS

The absolute costing of an operating native copper mine can be approximated from past cost data. The figures listed in Table 9 are given as percentages of the total costs for five native copper mines (two conglomerate lode mines and three amygdaloid lodes mines). Although all mines were not operating simultaneously, the production rates were approximately similar. The low standard deviations for each unit cost average indicate a relative uniformity of cost percentages between 1959 and 1968 when the various mines were in production. The lumping together of both types of native copper producers does not introduce a significant difference in the variance for the size of the populations sampled. For simplicity the costing is considered to be the same for both types.

Approximately 50 percent of the total cost was attributable to fixed mining costs, and 25 to treatment costs. The last column in Table 9 should be considered an *order-of-magnitude estimate* as defined by Zimmerman (1968). These are ratio estimations which are based on previous cost data and have a probable error over ± 30 percent.

Direct costs for the Kingston mine, which mined the Kingston Conglomerate from 1965 to 1968, have been estimated by Karkkainen (1977). The direct mining costs are estimated to be \$9-50/ton for a 1000 TPD mining rate. Direct milling cost estimates ranged from \$2.216 to \$3.80 per ton mined for a 1715 TPD and 1000 TPD mining rate, respectively.

If the mining cost estimated by Karkkainen is assumed to be relatively constant across the district for native copper producers, as the relatively low standard deviation from the five mines sampled would seem to indicate, then the total cost of mining can be derived from the percentages. A projected total costs would be \$19.48/ton based on a calculation which assumes that the mining cost is 48.756 percent of the total cost.

Based on these estimates a breakeven operation for a mine with an extractable grade of 1% copper (20 lbs/ton) requires a return of \$0.974 per pound of copper.

Mining costs at the White Pine mine are lower than those estimated for the native copper producers. Like all underground mines it is a relatively high cost operation in which expenses increase as the orebody is developed further from the shafts. It is current practice to control production so that only 9,000 TPD are hoisted rather than the 25,000 tons designed capacity so long as the price of copper is suppressed. Unfortunately, this production cutback is also reflected in a cutback in the workforce.

TABLE 9
ESTIMATED COSTS FOR A 1000 TPD NATIVE COPPER MINE

	Percentage of Cost	%	Estimated Cost/Ton
General Overhead			
Depreciation	5.466	1.931	\$ 1.06
General Officers	1.124	0.211	0.22
General Supplies	6.804	0.780	1.33
General Insurance	0.626	0.300	0.12
Employee Benefits	7.554	1.183	1.47
Legal Expense	0.090	0.018	0.02
	21.664		\$ 4.22
Taxes (except Federal)			
Property	0.734	0.130	0.14
State Corporation	0.552	0.140	0.11
Unemployment	0.494	0.240	0.09
Old Age	0.978	0.144	0.19
	2.758		\$ 0.54
Costs of Mining (Direct)			
Labor, Superintendence	26.156	0.995	5.10
Supplies, Maintenance, materials	22.600	2.687	4.40
	48.756		\$ 9.50
Treatment Costs			
Transportation, Mine to mill	4.024	0.445	0.78
Milling, concentration (Direct)	14.528	2.656	2.83
Transportation, mill to smelter	0.094	0.012	0.02
Smelting and Refining	6.798	2.071	1.32
	25.444		\$ 4.96
Costs Beyond Smelter	1.378	0.653	0.26
TOTALS	100.00		\$19.48

[TABLE 9: Estimate Costs for a 1000 TPD Native Copper Mine]

The mine and mill capital costs listed by Karkkainen (1977, p. 24) for the Kingston mine are offered as a model which may be applied to the district.

Mine Capital Costs

Preproduction Development	\$12,245,000
Underground Plant	2,300,000
Surface Plant	1,800,000
Working Capital	850,000
TOTAL	\$17,195,000

Mill Costs - \$3.80/ton mined
Milling Capital Cost - \$5,000,000
Working Capital - \$325,000
Net Copper Recovery - 90%

These figures are for a conventional simple crusher - flotation mill and concentration plant with a throughput of 13000 TPD. The cost of a larger mill can be estimated by the current "rule-of-thumb" used by industrial cost analysts which is \$5,000 x daily throughput (tons).

Mining capital cost would necessarily vary with the size of the operation; no attempt is made to cost a mine with a hoist greater than 1,000 tons. Preproduction

rehabilitation of existing mines would be extremely costly due to the amount of dewatering required. In many cases the shaft pillars were robbed in the final phases of mining and the shafts have collapsed.

FUTURE POTENTIAL

Exploration

It is unlikely that surface basic prospecting will reveal significant or mineable concentrations of copper. The area has been traversed many times and most outcrops already bear the marks of prospectors hammers. Probably the best chances of discovering more deposits will be from a combination of geochemical, geophysical, and exploration drilling.

A basic factor which thwarts geologists and prospectors is the glacial overburden which may cover up to 98 percent of a 7 1/2 minute quadrangle. The exposures of basalt which dip towards Lake Superior at angles of 70° near the base to 25° near the top of the volcanic pile weather nonuniformly. The massive, coarser-grained interiors form ridges with the intervening flow tops and conglomerates in parallel depressions or in valleys. It is possible that several large deposits underlie these low areas.

Copper Range Company has conducted geochemical soil surveys and geobotanical surveys on the volcanic rocks over areas known to be mineralized and have confirmed the presence of copper in the underlying bedrock. Several geochemical anomalies were outlined in covered areas where no previous deposits were known except for visible traces of copper (which is almost ubiquitous) in nearby outcrops. It is difficult to evaluate the magnitude of the anomalies in the drift covered areas because of the variable depth to bedrock and the fact that downslope movement of copper solutions would tend to collect in these areas.

Conventional geophysical methods which have proven to be successful in outlining sulfide mineralization cannot clearly delimit native copper mineralization. Recent refinements such as VLF-EM have also proven to be disappointing.

It is clear that the area still warrants considerable exploration. At present, the economic incentives to conduct a rigorous, long range exploration program have been dampened by low copper prices which resulted in closure of all the native copper producing mines, and a 60% cutback in the mining rate at White Pine. Nevertheless, the Copper Range Company has reinitiated an exploration program of its mineral holdings which could last for a period of 5 years.

Bodwell (1972) has investigated the exploration potential of the Portage Lake Lava Series by subdividing the pile into divisions based on the following marker horizons (see Figure 3):

- a) Base of lower division - Keweenaw fault.
- b) Base of middle division - Kearsarge amygdaloid.
- c) Base of upper division - Pewabic West conglomerate to Sec. 3, T 57 N, R 32 W, four miles north of Mohawk where unit pinches out. From there to Keweenaw Point, Ashbed amygdaloid.
- d) Top of upper division - lower contact of Copper Harbor conglomerate.

The data presented in Table 10 are from Bodwell (op. cit., Table 5, p. 23), and serves as a basis for his discussion of the past exploratory drilling efforts. His discussion (p. 26-27) is presented below:

"..... it can be seen that the highly productive middle division has received substantial exploration effort (40.7%) for the area north of Portage Lake. South of Portage Lake, the middle division has received considerably less attention, while the lower division has been more extensively explored. This reflects the empirical result of early discoveries of ore deposits in the lower division.

Portage Lake marks a break in the predominant stratigraphic position of ore deposits within the Portage Lake Lava Series. North of Portage Lake, 99% of production from the entire Portage Lake Series came from the middle division, whereas south of Portage Lake, 95% of all production came from the lower division.

Probably the most attractive opportunities for additional exploration lie in the lower division. It constitutes about 48% of the total section of which about 15-20% has been explored according to the above criteria. The middle division has not been extensively explored south of Portage Lake, but it thins rapidly and becomes progressively more difficult to define westward. The upper division shows 11% explored ground from Keweenaw Point to Norwich Exploration.

West of Norwich Exploration, no significant production has been recorded, although there are some old shafts and surface explorations. The character of the total section changes considerably westward from this point. The succession of predominantly basaltic lava flows (Portage Lake Series) continues southwesterly into Wisconsin. However, stratigraphically above these the succession changes to flows of predominantly andesitic composition with substantial sections of included felsitic flows and intrusives. In recent mapping (Johnson and White, 1969) the U.S. Geological Survey has

separated this higher series as a distinct formation, tentatively referred to as "unnamed formation."

The lavas west of Norwich resemble those of the principal district and copper occurs in similar manner. There is no obvious reason to believe that the lavas are any less favorable host rocks in this section than elsewhere. About 170 square miles of the lava series has been only superficially explored and there is recorded diamond drilling only at the Onondaga Exploration."

The controversy of whether the native copper was concentrated in the area of the principal mining district or whether it is a regional phenomenon within the volcanic conglomerate pile has not been settled. Bodwell (1972, op. cit.) notes that approximately 80% of the ground between Keweenaw Point to the Norwich section are so far unexplored if mineralization is considered to be of regional extent. If, however, the mineralization is limited to the area which has produced 96% of native copper production, 67% of that area remains untested and the likelihood of finding a large deposit, such as the Kearsarge Amygdaloid, are considerably reduced.

The basement rocks, the greenstone volcanic sequences which lie to the south and east of the Keweenaw Peninsula are potential hosts for massive sulfides. This area will probably be receiving more attention since the recent discovery of large amounts of sulfide mineralization near Crandon, Wisconsin.

TABLE 10
EXPLORATION DIAMOND DRILL HOLE DATA, PORTAGE LAKE LAVA SERIES

	Segment of Portage Lake Lava Series										Norwich to Wisc.	Totals Portage Lake Series
	Keweenaw Point to Portage Lake			Portage Lake to Norwich Expl.			Total-Keweenaw Pt. to Norwich Expl.					
Surface area, sq. miles	33.6	89.8	92.3	24.1	34.4	72.7	57.7	124.2	165.0	170.0	517.0	
Area drilled, sq. miles	5.3	36.5	10.6	1.3	2.8	17.1	6.6	39.2	27.7	negligible	73.4	
% of area drilled	15.6	40.7	11.5	5.4	8.0	23.5	11.3	31.6	16.8	"	14.2	

From Bodwell, 1972

[TABLE 10: Exploration Diamond Drill Data]

Reopening of Existing Mines and New Deposits

The existing mining technology and beneficiation techniques cannot be used to exploit the copper resources at present copper prices unless the deposits are selectively "Highgraded". The Presque Isle deposit in the Nonesuch Formation awaits the development of a suitable mining method because of the caving problems encountered in a test mining venture. The Caledonia project has not been considered because of the suppressed copper price. The reserves in the Kearsarge amygdaloid, a six mile long interconnected network of crosscuts, drifts, and shafts, require the removal and the ecologically acceptable disposal of

several billions of gallons of calcium chloride rich water before an equally expensive rehabilitation of the mines could reach the ore, proven at the time of closure, and to develop new ore. This essentially summarizes a somewhat bleak picture.

The aforementioned factors have not deterred a more or less continuous effort of research sponsored by Federal agencies or by the mining companies.

Most recently, a promising contribution to the extractive technology of native copper was reported on by Karkkainen (1977). The investigation dealt with a pilot sorting machine whereby crushed copper ore is sorted from barren rock by means of a voltage differential which activates a separatory mechanism. Although the present design has a low throughput, and economic analysis (assuming a greater capacity) shows that sorting could greatly reduce the costs of copper production. For example, his calculations show for certain cases that sorting could reduce the break even price by approximately \$0.05/lb. Furthermore, the reduction in price required to produce an acceptable 15% return on investment after taxes would be approximately \$0.15/lb.

SUMMARY

Based on data available for this investigation the Michigan copper resource is estimated to 19.5 billion pounds of copper, of this total 12.9 billion pounds are from Gogebic, Ontonagon, Houghton, and Keweenaw Counties in Michigan's Keweenaw Peninsula mining district. The remaining resource is from Marquette County. Of this total, most of the reserves are within the Nonesuch Formation. This does not necessarily reflect the true distribution of copper; it simply reflects the relative ease of estimating reserves within the Nonesuch lode as compared to the native copper lodes. The 1 billion pounds of copper estimated for these lodes is considered conservative.

It is unlikely that the native copper production would ever approach the production levels of bygone years. Nevertheless, the existing properties such as the Kingston mine, portions of the Kearsarge and Osceola amygdaloid mines, the Champion mine at Painesdale, the Caledonia property, and possibly the Quincy mine could provide a modest amount of copper to the domestic requirements with production rates of 500 to 2,000 TPD per 260 day year. Without new discoveries most of these native copper mine resources would be depleted within 5 years; some could remain in production up to 20 years.

Should the present technological and economic environment improve the existing native copper mines which are deemed re-habitable would require a common mill and smelter facility to be reconstructed. The mill should have a capacity of 10,000 TPD in the eventuality that at least five or six mines would be brought on line. At current prices, the facility would cost approximately 50

million dollars. Smelter costs may be avoided if the ore could be shipped to White Pine's smelter.

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APPENDIX A

[Mine Production Tables by Year]

[Allouez Conglomerate]

ALLOUEZ CONGLOMERATE

	<u>Allouez</u>
1925-1954	No production
55	31,207
56	115,109
57	222,508
58	---
59	137,327
1960	283,156
61	313,554
62	792,493
63	657,614
64	32,583
65	---
66	32,166
67	170,694
68	---

TOTAL = 2,788,411 pounds

[Baltic Lode]

BALTIC LODE

	<u>Baltic</u>	<u>Champion</u>	<u>Trimountain</u>
1926	3,690,041	17,658,820	2,177,416
27	4,207,819	16,098,926	2,354,808
28	3,415,082	19,563,535	1,275,515
29	2,127,926	20,660,701	1,408,589
1930	3,251,705	19,999,564	542,649
31	3,311,727	14,409,543	-
32		12,188,578	
33		12,167,130	
34		13,929,859	
35		16,759,889	
36		17,486,019	
37		15,790,600	
38		18,010,937	
39		17,856,237	
1940		18,207,973	
41		15,948,202	
42		17,772,116	
43		16,886,013	
44		13,989,318	
45		6,912,669	
46		60,482	
47		945,264	
48		1,889,603	
49		2,306,993	
		2,999,095	
1950		1,748,563	
51		2,630,204	
52		4,365,077	
53		4,157,588	
54		4,295,585	
55		4,435,469	
56		4,222,936	
57		2,951,192	
58		2,414,141	
59		2,369,843	
1960		2,573,688	
61		4,917,903	
62		3,700,767	
63		2,078,880	
64		1,950,979	
65		2,005,928	
66		2,919,542	
67			
68			
	20,004,300	384,176,397	7,759,077

TOTAL = 411,939,774 pounds

[Calumet and Hecla Conglomerate]

CALUMET AND HECLA CONGLOMERATE		
	<u>C & H</u>	<u>Centennial 3-6</u>
1926	26,422,924	
27	28,226,000	
28	34,933,000	
29	35,377,000	
1930	40,982,000	
31	40,349,000	
32	27,861,000	
33	34,686,000	
34	34,286,000	
35	38,230,000	
36	34,352,000	
37	28,724,000	
38	29,820,000	
39	8,436,000	
1956		2,995
57		428,289
58		844,174
59	470,329	533,519
1960	2,393,761	232,366
61	2,008,248	1,114,436
62	1,609,577	2,029,074
63	169,354	1,064,709
64	--	162,654
65		543,801
66		782,853
67		6,072,936
68		5,390,875
75		1,455,657
76		<u>3,886,218</u>
	<u>449,336,193</u>	<u>24,524,556</u>
TOTAL = 473,860,749 pounds		

[Kearsarge Lode]

KEARSARGE LODE				
	<u>Almeek</u>	<u>Peninsula</u>	<u>Centennial</u>	<u>Douglass</u>
1926	35,284,000			
27	34,065,000			
28	37,231,000			
29	35,105,000			
1930	26,538,000		1,900,000	
31	22,656,000		2,950,000	
32	4,493,000			
33	--			
34	--			
35	--			
36	24,963,000			
37	25,022,600			
38	18,680,000			
39	20,479,000	543,000		
1940	20,728,000	4,342,000		
41	22,624,000	5,466,000		580,000
42	15,948,292	4,908,125		3,025,041
43	10,423,493	8,971,344		5,090,150
44	12,630,348	9,872,217	3,176,258	1,566,239
45	5,423,722	6,360,643	5,146,617	2,738,541
46	2,122,259	4,512,473	5,009,413	4,518,126
47	2,791,315	4,528,737	5,627,680	4,110,942
48	2,400,231	5,306,886	5,068,224	4,832,466
49	1,393,535	3,093,260	3,577,711	3,638,998
1950	1,860,736	5,195,542	5,064,016	5,636,550
51	2,786,045	5,482,068	5,084,991	5,239,702
52	3,040,769	3,354,506	4,146,325	4,385,906
53	3,483,122	4,383,893	5,032,781	3,736,428
54	3,644,218	4,121,025	5,359,756	1,928,040
55	3,320,933	3,548,250	3,210,369	220,819
56	7,349,964	4,156,327	5,364,729	209,093
57	6,291,661	3,889,457	5,812,445	276,903
58	7,536,106	4,621,008	7,072,543	659,394
59	7,128,893	4,318,680	8,400,947	
1960	3,557,018	3,739,851	8,217,138	
61	2,212,953	4,232,563	6,247,634	
62	--	3,687,697	6,811,340	
63	--	3,874,030	5,602,033	
64	--	2,890,409	5,603,982	
65	--	859,281	3,466,309	
66	--	--	2,862,298	
67	--	--	725,089	
68	--	--	--	
	<u>433,234,213</u>	<u>120,050,222</u>	<u>127,546,618</u>	<u>52,393,311</u>
TOTAL = 1,065,687,902 pounds				

[Houghton-Iroquois Lodes]

HOUGHTON-IROQUOIS LODES		
	<u>Allouez #3</u>	<u>Iroquois</u>
1943	43,167	596,220
44	672,480	1,387,386
45	1,700,058	1,882,375
46	1,593,748	2,074,866
47	2,271,963	3,399,829
48	2,762,721	3,989,673
49	1,959,516	2,736,241
1950	3,080,115	4,023,452
51	2,417,174	3,870,396
52	2,861,589	3,441,272
53	2,898,133	4,390,572
54	2,618,501	3,742,502
55	1,868,069	2,394,916
56	2,399,419	3,196,693
57	2,082,953	1,933,033
58	2,064,185	--
59	1,701,499	2,341
1960	1,721,918	--
61	2,093,656	
62	1,700,517	
63	1,357,024	
64	214,547	
65	176,595	
66	675,438	
67	25,668	
68	--	
	<u>42,960,653</u>	<u>43,061,767</u>
TOTAL = 86,022,420 Pounds		

KEARSARGE LODE (continued)				
	<u>Mohawk</u>	<u>North Kearsarge</u>	<u>Seneca</u>	
1926	18,340,712	2,064,000	4,981,524	
27	22,124,830	1,400,000	2,881,382	
28	21,243,265	826,000	--	
29	20,043,265	1,600,000	2,999,882	
1930	13,306,762	2,427,000	4,857,895	
31	13,100,000	1,260,000		
32	8,450,388			
1932-1941	NO PRODUCTION			
1942		4,255,503		
43		7,731,503		
44		7,061,874		
45		4,732,440		
46		4,501,521		
47		4,478,937		
48		4,676,707		4,275,405
49		3,283,163		4,608,597
1950		4,280,069		7,762,538
51		3,820,827		8,128,994
52		3,035,618		6,437,911
53		3,356,535		8,437,187
54		3,809,179		7,840,330
55		2,385,674		6,113,436
56		307,535		7,587,311
57				8,131,404
58				9,006,237
59				9,032,436
1960				8,718,754
61				8,412,595
62				6,469,719
63				6,005,883
64				5,463,463
65				3,409,064
66				1,587,631
67				--
68				
	<u>116,802,708</u>	<u>11,294,085</u>		<u>143,149,578</u>
TOTAL = 1,065,687,902 pounds				

[Osceola Lode]

OSCEOLA LODE			
	<u>Osceola</u>	<u>Osceola #13</u>	<u>Osceola #6</u>
1926	9,527,000		
27	16,362,000		
28	18,318,000		
29	18,237,000		
1930	16,051,000		
31	5,152,000		
1954		45,762	
55		457,593	
56		2,249,247	14,943
57		2,197,687	95,875
58		---	376,714
59		5,379	---
1960		262,797	---
61		860,843	---
62		2,316,648	---
63		3,735,040	---
64		4,169,837	---
65		3,534,433	175,352
66		5,139,499	113,336
67		3,662,063	263,746
68		1,198,180	---
	83,637,000	29,831,008	1,039,966

TOTAL = 114,507,974 pounds

[Pewabic Lode]

PEWABIC LODE			
	<u>Quincy</u>	<u>Quincy Reclamation</u>	
1926	13,290,052		
27	9,718,662		
28	1,220,536		
29	4,459,426		
1930	10,939,787		
31	7,575,367		
32	---		
33	---		
34	---		
35	---		
36	---		
37	4,224,528		
38	8,426,054		
39	8,852,856		
1940	8,999,635		
41	8,032,175		
42	7,683,600		
43	6,544,621		
44	6,846,772	5,581,745	
45	2,956,470	6,066,700	
46	CLOSED	5,052,519	
47		4,750,895	
48		3,468,825	
49		4,621,896	
1950		5,946,545	
51		4,724,495	
52		5,549,064	
53		4,498,462	
54		4,272,144	
55		5,317,103	
56		4,760,000	
57		4,364,900	
58		780,640	
59		4,042,545	
1960		3,351,208	
61		3,486,265	
62		2,478,858	
63		3,840,510	
64		3,837,407	
65		2,320,273	
66		3,093,606	
67		336,707	
	109,790,541	96,635,312	

<u>Baltic Lode Sands</u>	<u>Atlantic Lode Sands</u>
1937	340,677
38	55,954
39	253,172
1940	661,727
41	729,102
42	669,491
43	571,674
44	554,909
45	209,305
46	2,741
47	338,390
48	---
49	---
1950	---
51	---
52	---
53	---
54	---
55	---
56	571,935
57	1,701,824
58	1,572,392
59	216,239
1960	2,086,398
61	1,226,081
62	1,217,664
63	591,680
64	941,408
65	487,158
66	---
67	---
	16,002,772
	Lake Linden Reclamation
	(121,007,924)
	Tamarack Reclamation (C & H)
	(6,679,000)

1926	22,970,076	8,018,000
27	20,192,000	8,525,000
28	22,437,000	7,459,000
29	25,991,000	7,520,000
1930	11,965,000	5,877,000
31		
32		
33		
34		
35	9,118,000	
36	19,167,000	
37	15,210,000	5,188,000
38	11,175,000	5,444,000
39	16,865,000	6,535,000
1940	22,488,000	7,194,000
41	24,647,000	8,119,000
42	20,672,454	10,078,562
43	18,577,234	10,232,727
44	11,400,419	3,337,744
45	6,981,339	3,279,833
46	4,269,473	2,986,012
47	3,846,636	3,239,039
48	7,217,728	3,268,976
49	3,697,944	2,599,736
1950	2,552,998	3,669,372
51	2,479,331	2,921,021
52	1,457,486	2,393,936
53		2,890,980
54		2,280,126
55		1,820,255
56		3,692,993
57		2,855,090
58		2,819,375
59		2,651,447
1960		2,576,069
61		2,658,215
62		2,459,020
63		2,432,115
64		2,530,911
65		1,971,964
66		2,319,957
67		1,451,252
	305,378,118	155,295,727

[Plate 2-5: Geological Maps - Isle Royal to Seneca]



APPENDIX C

MINING METHODS

The mining method used in the native copper mines was more or less uniform throughout the district. In general, tonnages were obtained from development down the dip of the lode. The shafts usually followed the lode down dip, in some cases within the mineable horizon. Calumet and Hecla Mining Company used a retreat open-stope mining method which proved quite satisfactory.

Levels which did not connect with the shafts directly were, in some cases, connected by crosscuts which served as skip compartments. These skip loading pockets were installed at alternate levels and connected to the levels above by a storage raise. Alternatively, ore was drawn directly to the skips in areas where pocket loading was impractical.

Usually the lode was mined by stoping to the level above. As stoping proceeded a sufficient number of round pillars were left in the stope to support the hanging wall.

The room-and-pillar method is practiced almost exclusively at White Pine. This mining method would not be satisfactory at the Presque Isle (Western) Syncline deposit because of roof caving problems. Possibly a modification of a longwall mining method could be developed which would allow the economic exploitation of the deposit.