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Utilization of Scrap Preheating and Substitute Slag Conditioners for Electric Arc Furnace Steelmaking

By G. W. Elger, R. H. Nafziger, J. E. Tress, and A. D. Hartman

U. S. Department of the Interior

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cfm	cubic foot per minute	kW•h/st	kilowatt hour per short ton
dscf	dry standard cubic foot	16	pound
°F	degree Fahrenheit	lb/lb mol	pound per pound mole
ft	foot	lb/min	pound per minute
ft ²	square foot	lb mol	· · · · · · · · · · · · · · · · · · ·
ft ³	cubic foot		pound mole
ft/ ³ 1b	cubic foot per pound	lb/st	pound per short ton
ft/s	foot per second	mg	milligram
ft ³ /st	cubic foot per short ton	mi n	minute
g	gram	mL	milliliter
g/mL	gram per milliliter	pct	percent
	-	°R	degree Rankine
gr/dscf	grain per dry standard cubic foot	scf	standard cubic foot
n	hour	scfm	standard cubic foot per minute
in	inch	st	short ton
ln Hg	inch of mercury	st/yr	short ton per year
In H ₂ O	inch of water	vol pct	volume percent
cV•A	kilovolt ampere	_	_
:W	kilowatt	wt pct	weight percent
.₩•h	kilowatt hour	yr	year

UTILIZATION OF SCRAP PREHEATING AND SUBSTITUTE SLAG CONDITIONERS FOR ELECTRIC ARC FURNACE STEELMAKING

By G. W. Elger, R. H. Nafziger, J. E. Tress, and A. D. Hartman

ABSTRACT

The Bureau of Mines investigated preheating of continuously fed, fragmented ferrous scrap charges by furnace offgases and utilizing substitutes for imported fluorspar to condition electric steelmaking slags.

Three types of continuous scrap charging procedures were investigated to determine electrical energy consumption in a 1-st (short ton) electric arc furnace. Cold and preheated scrap charges were continuously fed at rates averaging 37.5 and 43.7 lb/min, respectively. The feed rate varied appreciably from test to test owing to hangup of the scrap in the charge bin. Approximately 7 pct less electrical energy was consumed in melting scrap preheated to 840° to 1,110° F by furnace offgases than in melting cold scrap. Overall energy consumptions were 888 kW*h/st for cold scrap, 829 kW*h/st for preheated scrap, and 637 kW*h/st for conventional backcharged scrap.

Stack gases from scrap preheating averaged 120° F and a flow rate of 1,615 scfm compared with 220° F and 1,302 scfm for cold-charged scrap.

Slag conditioners tested included synthetic fluorspar, boric acid, hydroboracite, used aluminum potlining wastes and aluminum anode tailing wastes, and Sorelflux B. The boron-containing conditioners fluidized the slags the best, followed in order by the conditioners containing fluorine and titanium.

INTRODUCTION

This investigation is part of the Bureau of Mines program to develop technology that emphasizes the reuse of recycled, economically important materials, including iron and steel scrap. In addition, the research helps meet the basic objective of substituting abundant domestic materials for imported strategic commodities such as fluorspar.

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Of the 84.6 million st of raw steel produced in the United States in 1983, 31 pct or 26.2 million st was produced in the electric arc furnace (1).5 Nearly all of this steel was derived from cold ferrous scrap materials using a series of backcharges whereby the furnace roof is opened periodically to admit additional It has been reported scrap. 489 kW·h of energy is lost from a 40-st furnace each time the roof is opened to permit backcharging (2). The overall average energy consumption for steel produced by this process is approximately 535 kW h/st (3). This represents 140 billion kW h used to produce domestic steel by the electric arc furnace process in 1983, the last year for which figures Increasing electric power are available. costs, coupled with the need to improve efficiency and productivity of electric furnace steelmaking operations to compete with cheaper imported steel, provide an incentive to decrease electrical energy consumption.

In the traditional electric arc furnace process for producing steel, approximately 20 to 22 pct of the total heat generated is lost in the hot gases emitted from the furnace (4). Makeup heat must be added, which uses additional energy. In addition, the hot offgases often require external cooling prior to fume re-Literature estimates show that an approximate 10-pct decrease in electrical power requirements is achievable through scrap preheating (5). Therefore, a technique to preheat the scrap using the hot offgases would offer the dual benefit of conserving energy while cooling the gas.

Developments in preheating scrap charges for electric furnace steelmaking occurred as early as 30 yr ago. However, in nearly every case, some means of external heating was used. The techniques included the use of an oil burner mounted in either a charging bucket (6-10) or the furnace (7), the use of natural gas in a lance connected to a chute leading to the

furnace (11), or the use of a charging bucket with louvers that is placed above hot billets or ingots (12). cases, electrical energy usage was de-However, with few exceptions, creased. no mention was made of increased fuel In one case, electrical energy usage. consumption decreased by 5 to 12 pct at the expense of gas consumptions ranging from 210 to 380 ft3 per net short ton of metal, depending on the scrap density (9). Overall energy costs were decreased by the replacement of electrical energy by preheat fuel (10). Further significant energy cost savings could be attained if no fuel were required for preheating scrap.

Previously developed scrap preheating techniques can be classified into three types. The first utilizes a special vessel for preheating. Typically, external fuel burners supplement the heat from the offgases used. After preheating, the scrap is transferred to a charging bucket prior to introduction into the furnace. Considerable scrap handling, large space requirements, and high costs for the vessel pit are disadvantages of this process (13). Bethlehem Steel used this technique in the past.

Another method involves placing the charge bucket into a preheating vessel. In this technique, the scrap cannot be preheated to a high temperature and there is more dust. This is the practice used by Daido Steel in Japan whereby solid fuel is added to the preheater (14). Electrical energy savings of 45 kW h per short ton of scrap are claimed.

In the Baumco-Hotaka process, hot furnace offgases are directed toward scrap preheating stands where the ladles are heated. Energy savings of 43 kW h per short ton of steel have been reported Toshin Steel Co. and Nippon Kokan K. K. in Japan jointly developed a scrap preheating process whereby hot furnace offgases are directed toward a preheating chamber to heat the scrap in a charging If the scrap does not contain oil and other combustibles, the gas leaving the preheater is sent directly to the Unburned gas from the prebaghouse. heater containing combustibles is afterin a combustion burned chamber

⁵Underlined numbers in parentheses refer to items in the list of references preceding the appendix.

recycled to the preheater. From 36 to 46 kW h per short ton of steel was conserved during preheating operations in 110- and 55-st-capacity furnaces, respectively (15).

In the third technique, a bucket lined with castable refractories serves as a preheating vessel and a charging bucket. Potential bucket distortion, dust losses, and a large weight requiring a larger crane capacity can cause problems with this method.

Both Fujisawa Steel and Mitsubishi Steel employ scrap preheating in a clean-house enclosure. Scrap is preheated in a special chamber, which then is transported to the furnace. This eliminates the transfer of preheated scrap from one bucket to another. More than 36 kW h per short ton of steel was conserved (16).

All of these techniques involve back-charging methods for feeding the furnace. At present, no scrap preheaters are in operation at any major domestic mills. Small preheaters are installed in some minimills. Relatively high capital costs have precluded the adoption of fuel-fired preheaters. Other disadvantages of these systems include (1) distortion and warpage of the charging bucket doors, (2) uneven heat distribution within the charge, and (3) scrap oxidation.

Research focusing on the preheating of prereduced iron ore materials was conducted by the Bureau of Mines using electric furnace offgases (17-18), and by CANMET using an external heater (19). the case of prereduced iron ore brigreater than 10 pct savings in energy consumption was realized when preheated briquettes were charged, compared with cold briquettes (17).Up to 15.6 pct lower energy consumption was achieved when using preheated, prereduced iron ore pellets (18). CANMET obtained a 27-pct decrease in energy consumption by preheating iron ore pellets prereduced by the Stelco-Lurgi, Republic Steel-National Lead (SL-RN) process (19).

Experiments in preheating scrap materials for charging to basic oxygen furnaces (BOF) have been conducted during the past two decades to decrease hot metal requirements. Preheating was accomplished by using oxygen natural gas or

oxygen-fuel oil burners (20-21). British investigators used BOF waste heat as a preheating medium (22). Later Bureau studies using hot waste gases from a 500-1b BOF at the Twin Cities Research Center to save fuel and decrease pig iron requirements showed that thermal energy recovered could account for up to 44 pct of the energy necessary to melt the scrap (23).

It is possible that energy and cost savings could be realized if the need to open the furnace roof was eliminated by continuously charging scrap into the furnace. Research involving the continuous charging of cold shredded scrap into a 1-st electric arc furnace was conducted by the Bureau of Mines (17-24). Preheating continuously charged scrap would help reduce energy requirements further in scrap meltdown.

One objective of the present investigation was to evaluate the potential energy savings while continuously charging preheated and sized scrap into a 1-st electric arc furnace. Use of the sensible heat in the furnace offgases alone should decrease the energy requirements compared with those necessary for charging cold scrap.

SUBSTITUTE SLAG CONDITIONERS

Fluorspar is a critical mineral. proximately 80 pct (453,000 st) of domestic fluorspar consumption was imported in 1983 (25). Approximately 34 pct of the fluorspar consumed domestically is used as a fluidizer in steelmaking. steelmaking operations require a slag fluidizer to promote the required reacto increase productivity, prove tapping operations, and to conserve Nearly 194,000 st of fluorspar was consumed in steelmaking in 1983 (25). The electric furnace consumes approximately 3.2 lb of fluorspar per short ton of raw steel.

A number of wastes and materials more readily available than fluorspar domestically are not being used as extensively as possible. Disposal of several of the waste materials presents environmental problems. If a satisfactory substitute can be found, U.S. dependence on foreign

sources of suitable fluorspar would diminish and import costs could decrease.

The search for substitutes for fluorspar as a slag conditioner began several years ago with work conducted at Stelco (26) and the Bureau of Mines Twin Cities Research Center (27). Colemanite (a calcium borate), fused boric acid, and MnO2 were judged to be suitable substitutes for open hearth and basic oxygen fur-QIT-Fer et Titane Inc. has been promoting its Sorelflux (ilmenite) as a good fluorspar substitute for several Further experiments were conducted at the Bureau's Albany and Twin Cities Research Centers to determine whether synthetic fluorspar prepared from byproduct fluosilicic acid from a fertilizer process could replace natural fluorspar in electric furnace and BOF steelmaking. Results were positive, and no adverse effects on furnace operations were observed (28-29). Experiments also were conducted at the Twin Cities Research Center involving the substitution of used alumina pot cell linings for The linings of alumina refluorspar. duction cells must be replaced regularly. The linings are carbonaceous and become impregnated with fluorine and sodium compounds while in service. The material

accumulates at a rate of approximately 190,000 st/yr and presents a disposal problem for the aluminum industry (30). Fluid slags can be obtained using this material in a cupola during ironmaking operations or in a BOF for steelmaking $(31-33)_{\omega}$ No metal contamination was observed. Most recently, trials were conducted in a 4-st electric arc furnace using spent potlining material. the heats provided steel within specifi-From a qualitative standpoint, cations. the slags with the potlining fluidizer were more fluid than those obtained when fluorspar was used (34).

All of the previous work demonstrated that substituting wastes or readily available materials for fluorspar in various iron and steelmaking operations was feasible on a qualitative basis; however, systematic evaluation has been made a wide variety of materials in the electric arc furnace. Accordingly, second objective of this research was to evaluate the use of substitute materials for their ability to condition the slag during melting operations in a 1-st electric arc furnace. Substitute materials were compared with fluorspar on the basis of visual observations made of the bath before and after the fluidizer addition.

EXPERIMENTAL EQUIPMENT AND PROCEDURE

MATERIALS

Ferrous Scrap

The shredded scrap used for these tests was purchased from a local scrap processor and consisted of three separate each purchased at a different batches, time and with a different composition. The metal stampings used in the charges were purchased from the same source. The chemical analyses of these materials are shown in table 1. The analyses were obtained by a direct-reading vacuum spectrograph from cast samples of scrap melted in separate 800-1b wash heats without the addition of quartz and lime to provide a slag cover. Even though the melted material may have a composition different from that of the unmelted

scrap, this is standard practice for scrap sampling. The analyses of the stampings (thin-gage sheet metal) were obtained without melting.

Because the inlet to the preheater was smaller than some of the pieces of scrap, it was necessary to size the shredded scrap that was to be fed continuously. Only pieces with largest dimensions of less than 4 in were selected. third batch purchased, the smaller size fraction contained substantial quantities of fragmented cast materials, parts, and nonferrous metals. Analyses of most elements in the small-size scrap were different from those of the larger pieces, as shown in table 1. The large pieces of scrap in the third batch consisted of a preponderance of body steel, bumpers, and suspension parts.

TABLE 1. Chemical composition of shredded automotive scrap 1 and metal stampings used in three types of scrap charging tests, weight percent

Batch	1	2		3	Stamp-
			Large	Small	ings
A1	0.46	0.83	NA	2.15	0.046
C	.41	•56	0.051	1.66	.67
Cr	.16	.40	.050	• 47	.49
Cu	.18	.81	• 20	1.18	.087
Fe	98.5	95.7	99.5	91.5	97.2
Mn	.12	.43	.036	•69	•41
Ni	.17	.30	.11	.31	.70
P	.008	.034	.011	.070	.013
S	.030	.045	.037	.048	.008
Si	<.01	.85	.011	1.86	.26
Sn	NA	.019	.009	.037	.005
Ti	NA	<.01	<.01	<.01	.003

NA not analyzed.

¹Lead and zinc contents of the scrap samples were not analyzed since these constitutents volatilized from the furnace during charge meltdown and were recovered in the dust product.

Slag Fluidizers

Three types of slag conditioners were used: fluorine-, titanium-, and boron-containing compounds or materials. The

fluorine group included natural fluorspar, synthetic fluorspar, used potlining, and butt tailings. The natural fluorspar was purchased from a commercial supplier and was the type normally used in steelmaking operations. The synthetic fluorspar was made at the Albany Research Center by reacting limestone with waste fluosilicic acid obtained from a phosphate plant. The used potlining and butt tailings were obtained from Kaiser Aluminum and Chemical Co., Spokane, WA. were from the aluminum reduction cells. The butt tailings were the ends of the carbon anodes.

The titantium-containing material was Sorelflux B. This was furnished by QIT-Fer et Titane Inc. and was mostly ilmenite and feldspar with a small amount of hematite.

The boron-containing conditioners included boric oxide and Girstley borate. Boric oxide was prepared by fusing and grinding boric acid. The Girstley borate, which consisted of hydroboracite (CaMgB₆O₁₁·6H₂O), is used as a glaze in the ceramics industry. The Girstley borate was heated to 1,475° F to drive off the contained water. The chemical analyses of all the slag conditioners are shown in table 2.

TABLE 2. - Chemical composition of slag conditioners used in steelmaking tests, weight percent

	Natural	Synthetic	Used	Butt	Sore1-	Boric	Girstley
	fluorspar	fluorspar	potlining	tailings	flux B	oxide	borate
A1	2.11	0.09	11.3	2.02	1.97	ND	0.50
B	ND	ND	ND	ND	ND	22.2	9.72
C	1.27	.71	1.86	63.8	ND	ND	ND
Ca	40.6	45.3	2.10	.53	.66	ND	12.7
F	34.8	40.5	21.5	7.5	ND	ND	ND
Fe	.28	ND	ND	ND	38.8	ND	.23
Mg	.27	•25	.10	.04	1.79	ND	2.01
Mn	ŅD	ND	ND	ND	•12	ND	ND
Na	ND	ND	25.0	4.1	ND	ND	3.75
P	ND	ND	ND	ND	.021	ND	ND
Рь	1.0	.34	ND	ND	ND	ND	ND
S	<.39	.19	.27	2.85	ND	ND	ND
Si	3.70	1.77	.81	.31	2.65	ND	4.10
Ti	ND	ND	ND	ND	20.3	ND	ND
V	ND	ND	ND	ND	•25	ND	ND
	dotorminad	The second secon					

ND Not determined.

EQUIPMENT

All tests were conducted in a conventional electric arc steelmaking furnace (Lectromelt Corp., model ST)6 lined with a basic refractory brick and covered with a rammed-alumina roof. The inside diameter of the furnace was 39 in, and it had a nominal capacity of 1 st of molten me-The electrical energy was provided by a 1,200-kV·A transformer through three 4-in-diam graphite electrodes. Four voltage taps were available. The furnace roof could be swung aside for top charging, and a slag door was located on the side opposite the taphole for access to The entire furnace the molten bath. could be tilted for tapping. Figure 1 shows a layout of the furnace, scrap feeder-preheater, and furnace offgas ductwork including gas sampling location and dust removal units. This furnace has been described previously in more detail (18).

An enclosed chute fabricated from stainless steel was attached to an opening in the furance roof for feeding charge materials and for exhausting hot At the opposite end of the chute was the rotating feeder-preheater, which consisted of six 12-in-diam sections of stainless steel tubing connected at 90° to each other in a zigzag fashion. horizontal section fitted at the feed end was mated to the exhaust ductwork. preheater was insulated with a 2-in-thick Fiberfrax refractory blanket covered with aluminum foil. In use, the charge was fed through a chute into the horizontal section, where two helices attached to the inside surface moved the charge into the zigzag portion. The inlet to this chute was open to the air, which permitted infiltration of air into ductwork at the scrap feed end. The rotation of the preheater caused the charge to move through this section to the discharge hot exhaust gases from the furnace passed through the preheater countercurrent to the direction of the

charge, thus heating the charge and cooling the gases. Thermocouples located at each end of the preheater measured the temperatures of the inlet and exit gases.

A stainless steel cyclone was placed in the ductwork downstream from the preheater to remove as much dust as possible The gas from the exhaust gas stream. exited directly into a long vertical section of duct that assisted in the removal of those smaller sized dust particles escaping the cyclone. The gases then were directed horizontally to a second cyclone and an adjacent baghouse for final cleanbefore being exhausted to the ing

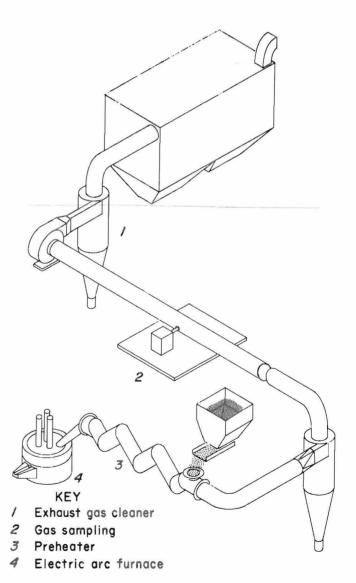


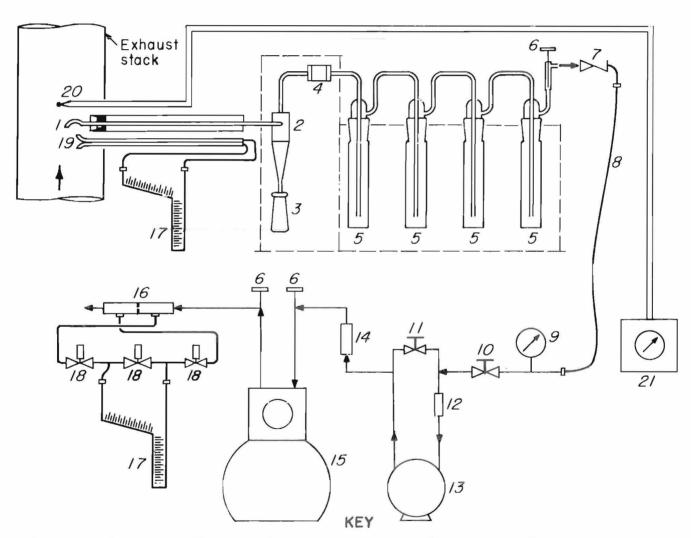
FIGURE 1.—Experimental equipment used in scrap preheating and meltdown tests.

⁶Reference to specific manufacturers or products does not imply endorsement by the Bureau of Mines.

atmosphere. Gas and particulate samples were taken in this horizontal section.

The stack gas sampling and measuring equipment used was a portable Rac Train Stak sampler, manufactured by Research Appliance Co. The components in the sampling train for particulate stack sampling, shown in figure 2, included the probe with attached nozzle, a particulate collector, a cooling and/or gas collector with four impingers, flow-measurement devices, and a vacuum source.

A 1/4-in-diam nozzle was attached to a 4-ft probe (1). The particulate, which was drawn through the probe by a vacuum pump (13), was separated from the gas by a cyclone (2) and the fiberglass filter (4) in a heated compartment. The probe also was heated to avoid condensation of moisture before reaching the ice bath. To condense the moisture out of the gas sample, ice was used around the impingers (5). The condenser section (5) consisted of four glass impingers connected in



- / Probe 2 Cyclone 3 Flask 4 Particulate filter 5 Impingers 6 Thermometers 7 Check valve 8 Connecting cord 9 Vacuum gage 10 Coarse adjust valve
- // Fine adjust valve 12 Oiler 13 Vacuum pump 14 Filter 15 Dry gas meter
- 16 Orifice tube 17 Incline manometer 18 Solenoid valves 19 Pitot 20 Thermocouple 21 Temperature recorder

FIGURE 2.-- Apparatus used for stack gas measurements and sampling.

series by ball-and-socket joints. The first and second impingers contained 100 mL of distilled water, the third impinger was empty, and the fourth contained 200 g of silica gel. Greenburg-Smith impingers were used with the tips modified on the first, third, and fourth impingers. The modifications consisted of the standard tip being replaced with a 1/2-in-ID glass tube extending to 1/2 in from the bottom of the impinger. The dried gas from the impingers was drawn through a dry gas meter (15) and an orifice meter (16).

A type-S pitot tube (19) was attached to the sampling probe (1) for continuous measurements of velocity so that adjustments in sample flow could be more rapidly applied to meet changes in stack flow conditions.

PROCEDURES

Electric Arc Furnace Tests

For all tests, the furnace was preheated by inserting a propane burner through the slag door and preheating for approximately 12 h. The initial charge of shredded auto scrap, reductant, and slag formers was top charged to the furnace by means of a charging bucket. The amount of scrap in this initial charge was 450 lb. The slag formers consisted of 62 lb of pebble lime and 40 lb of quartz, for a calculated basicity [(CaO + MgO)/SiO₂] of 1.5. The reductant used was 25 lb of metallurgical coke.

Each test began when the furnace was first energized to melt the initial charge. The highest voltage and amperage available were used. In the continuous feeding tests, the initial charge was melted completely. The continuous feeding then was started; a total of 1,350 lb

of ferrous scrap (1,080 1b of shredded scrap and 270 1b of stampings) was fed during each test. The difference between tests using a preheated charge and those using a continuously fed charge at ambient temperature was the feed location. The scrap in the ambient temperature tests was fed into an opening in the side of the feed chute between the furnace and the preheater, as shown in figure 3. The scrap in the preheated tests was fed directly into the preheater shown in figure 4.

Samples of preheated scrap were collected during the continuous-feeding period by catching several scrap pieces in a wooden scoop as they dropped into the furnace. They were immediately immersed in water of known quantity and temperature without creating any appreciable emission of steam. The temperature increase of the water and the weight of the scrap were recorded. Using these data, it was possible to calculate an approximate temperature of the preheated scrap.

After the feed had entered the furnace and the bath was molten, the slag conditioner (if any) was added. After about min of heating, the slag was removed from the bath. After the addition of 10 1b of silicomanganese, the bath was adjusted to the proper temperature, and the furnace was tapped. Both metal and slag samples were taken at various times during the tests. These samples usually were taken after initial meltdown, midway through and at the end of the continuous feeding period, after slag removal, and at the tap. Bath temperature measurements were made as necessary using Pt expendable versus Pt-10 pct thermocouples.

During the backcharging tests, only a portion of the initial 450-lb charge was

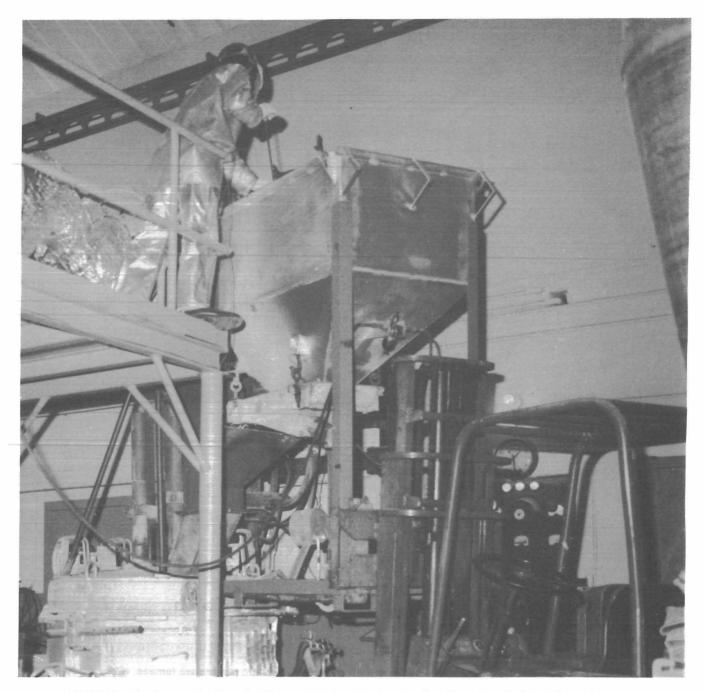


FIGURE 3.—Continuous feeding of cold scrap into feed chute extending through roof of electric arc furnace.

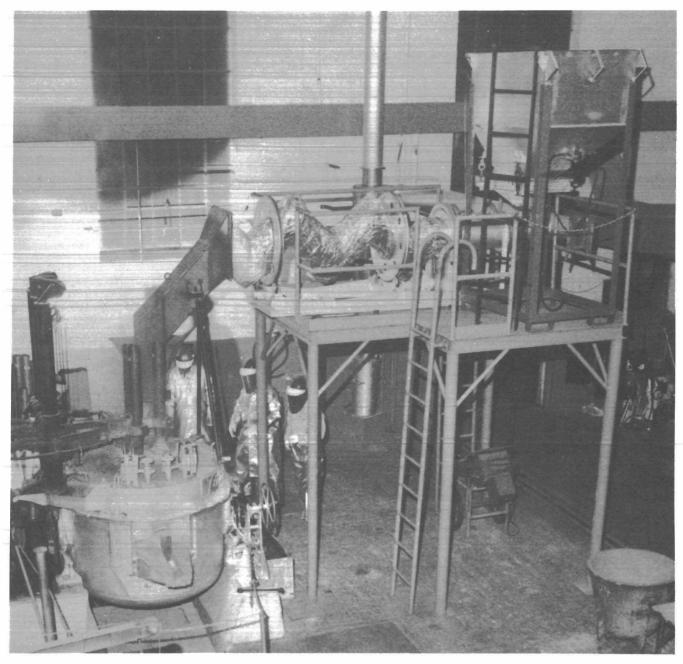


FIGURE 4.—Continuous charging of preheated scrap into electric arc furnace.

melted. At that point, the furnace roof was swung aside, and the first back-charge, consisting of 675 lb of shredded scrap, was loaded as shown in figure 5. When this was melted sufficiently to allow another 675 lb of shredded scrap to be added, the same procedure was followed. The remainder of the test was identical to the continuously charged tests.

Furnace Offgas and Particulate Sampling

Furnace offgases were measured and sampled at a location in the horizontal section of the 14-in-diam duct approximately 80 ft from the furnace. This location met the requirements of no disturbances within eight duct diameters downstream and two duct diameters upstream from the sampling location. Four sampling points

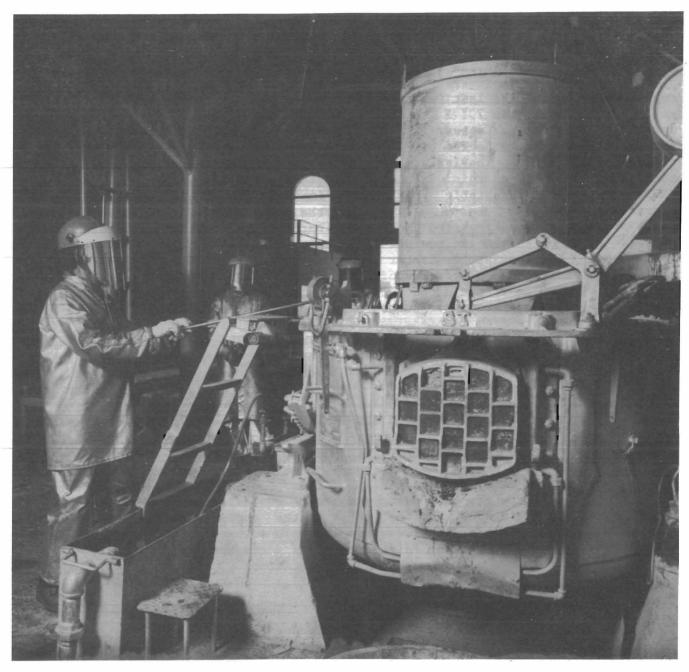


FIGURE 5.—Backcharging of scrap into electric arc furnace.

were located on a horizontal traverse at distances of 1.0, 3.5, 10.5, and 13.0 in from the inside wall of the duct. Because of size limitations of the sampling area and the short sampling times available, only one traverse was made. Particulate concentrations in the offgas were determined according to standard U.S. Environmental Protection Agency (EPA) method 5 (35-36).

Sample Train Preparation

A glass-fiber filter was first dessicated, weighed to the nearest 0.5 mg, and then placed in a glass-filter holder. The condenser section consisted of four glass impingers connected in a series by ball-and-socket joints. The first and second impingers contained 100 mL of distilled water, the third impinger was

empty, and the fourth contained 200 g of All impingers were weighed silica gel. to the nearest 0.01 g. To condense the moisture out of the gas sample, ice was placed around the impingers. The gas temperature leaving the last impinger was maintained at less than 70° F to prevent damage to the dry gas meter. The sampling train was leak-checked by using a vacuum of 15 in Hg. The leak rate did not exceed 0.02 cfm at this vacuum and, acceptable by EPA therefore. was standards.

The probe then was inserted into the duct and attached to the cyclone and filter system, and the probe heater was adjusted to provide a gas temperature of approximately 250° F at the probe outlet. The filter heating system was turned on to prevent moisture from condensing out of the gas sample.

Sample Collection

Particulate sampling for the continuous-charging tests was accomplished during the scrap feeding period of the experiments. For the backcharge tests, offgases were sampled during the meltdown of the first of two scrap backcharges.

A velocity traverse of the duct was taken first. To begin sampling, the nozzle was positioned at the first traverse point, with the tip pointing directly into the gas stream. The vacuum pump was

started and the flow adjusted to isokinetic conditions. A nomograph was used with the velocity pressure reading from the type S pitot tube manometer to obtain the correct sampling rate for isokinetic conditions. Each traverse point was sampled for 5 min, and readings were recorded at each point for gas temperature, static pressure, velocity head, orifice pressure differential, gas sample volume, gas sample temperature leaving the last impinger and at the inlet and outlet of the dry gas meter, and vacuum gage. barometric pressure was recorded during the run, and an Orsat gas sample was taken during the run for analyses of CO, CO_2 , O_2 , and N_2 constituents.

The sample train was carefully moved from the test site to a sample recovery area to minimize the loss of collected The impingers were weighed to determine the moisture gain. The filter was placed in a tared beaker and dried to constant weight, while the nozzle, cyclone, and collection beaker probe. were cleaned with reagent-grade acetone. The acetone washings in a tared beaker were evaporated to dryness at ambient temperature and pressure and then dried to a constant weight. The data collected were evaluated using the equations shown in the appendix to obtain the average sample volume, moisture content, particuconcentration, late and isokinetic variation.

RESULTS

CONTINUOUSLY CHARGED PREHEATED SCRAP

The shredded scrap used continuous-feeding tests was carefully sized to remove the oversize pieces as well as the metal coils and springs that could cause plugging of the feed inlet into the preheater unit. The temperature of the furnace offgases introduced into the preheater ranged between 1,020° and 1,200° F. The temperature of the exit gases from the preheater was usually less than 390° F owing to transfer of heat to the preheated scrap and infusion of air into the feed inlet during continuous charging of scrap. Preheated scrap temperatures of 840° to 1,110° F were obtained based on measurement of samples as described previously. The efficiency of heat exchange between the furnace offgases and the scrap was influenced by the air infiltration into the scrap preheater.

Table 3 summarizes the averaged experimental data, including power input, scrap feed rates, and electrode and electrical energy consumption for 12 tests made with preheated scrap. The scrap feed rates ranged from 22.9 to 66.2 lb/min, and the power input ranged from 440 to 682 kW. Observations revealed the scrap did not build up in the bath at the feed rates

TABLE 3. — Experimental data for steelmaking tests using continuously charged preheated scrap

Test	1	2	3	4	5	6
Overall data:						
Total test timemin	114.7	81.7	87.2	91.9	83.1	99.2
Av power inputkW	463	594	602	583	597	493
Specific energy consumption						}
KW•h/st scrap	924	785	806	876	81.5	787
Specific electrode consumption						
1b/st scrap	25.2	16.3	15	16.1	13.2	14.5
Meltdown period:		Ì				
Av power inputkW	533	564	563	468	560	478
Specific energy consumption						
kW•h/1b scrap	0.424	0.418	0.409	0.411	0.400	0.471
Continuous-addition period:						
Av scrap feed ratelb/min	22.9	66.2	57.9	44.7	45.2	39.7
Av power inputkW	440	611	613	626	628	527
Specific energy consumption		-	0.00			5_,
kW·h/lb scrap	0.405	0.296	0.361	0.325	0.330	0.338
Refining period:	00 105	0.270	00001	00025	0.330	0.330
Av power inputkW	489	605	588	683	562	314
Tap temp°F	3, 100	3,040	3, 100	3, 095	3,090	2,850
Products, 1b/st scrap:	3, 100	3, 0 10	3, 200	3, 0,5	3,070	2,030
Metal	1,767	1,840	1,848	1,872	1,810	1,825
Slag	132	262	199	216	241	208
Stack dust	42	45	31	50	83	47
Test	7	8	9	10	11	12
Overall data:	····′	- 0	,	10	11	12
Total test timemin.	83.0	76.4	86.3	89.7	83.1	85.0
	584	634	617	549	609	604
Av power inputkW	304	0.54	017	349	009	004
Specific energy consumption	791	812	837	817	815	880
kW•h/st scrap	/91	012	037	017	013	000
Specific electrode consumption	1 / 0	15 1	12 1	1 / 2	1/. 7	1 5 7
1b/st scrap	14.8	15.1	13.1	14.2	14.7	15.7
Meltdown period:	540		F 2 1	410	540	540
Av power inputkW	563	648	531	410	569	563
Specific energy consumption						
kW•h/lb scrap	0.402	0.360	0.411	0.431	0.449	0.444
Continuous-addition period:				[
Av scrap feed ratelb/min	41.5	41.3	42.3	41	41.9	38.4
Av power inputkW	614	650	672	630	636	682
Specific energy consumption kW·h/lb						
scrap	0.356	0.386	0.381	0.344	0.343	0.391
Refining period:						
Av power inputkW	468	500	547	626	591	534
- 0-	2,920	3,010	2,750	2,950	2,920	3,085
Tap temp°F						
	,					
Tap tempF Products, 1b/st scrap: Metal	1,803	1,903	1,940	1,926	1,846	1,893
Products, 1b/st scrap:		1,903 373	1,940 236	1,926 206	1,846 256	1,893 256

used. For the continuous-feeding period, electrical energy consumption ranged from 0.296 to 0.405 kW h per pound of charge. Overall energy consumption for the tests ranged from 0.392 to 0.462 kW h per pound of scrap. The large differences noted in specific energy consumption from test to test are primarily due to the large variations in scrap feedrates. Problems were encountered with maintaining continuous flow of scrap from its charge bin. Electrode consumptions of 13.1 to 25.2 1b per short ton of scrap were noted in these tests.

Table 4 summarizes the experimental data obtained during measurements of stack gas temperatures, flows, and particulate concentrations. Temperatures of the gases ranged from 101° to 135° F, while the flows were 1,382 to 1,783 scfm. The stack gas contained 0.03 to 2.24 gr/dscf of particulate.

Table 5 shows a material balance for a test made with preheated scrap and Sorel-flux B as a slag conditioner. The data in table 5 include the weights and analyses of the charge materials and products, as well as the furnace dust recovered from the two cyclones and baghouse. Orsat and particulate data are not included in the material balance because these data were obtained only during a portion of the test. The metal product contained relatively high levels of copper, silicon, aluminum, and carbon since

no attempts were made to produce a particular grade of steel. Almost half of the furnace dust consisted of zinc, along with lesser amounts of lead, iron, magnesium and other constituents.

CONTINUOUSLY CHARGED COLD SCRAP

The experimental data for 14 tests made with continuously charged cold scrap are shown in table 6. The scrap feed rate ranged from 30.5 to 63.7 lb/min, and the power input ranged from 498 to 657 kW. Electrical energy consumption for the feeding period ranged from 0.315 to 0.431 kW.h per pound of charge. Overall, the energy consumption was 819 to 1,000 kW.h per short ton of charge, and the electrode usage was 10.5 to 19.1 lb per short ton of scrap.

Data for the measurements and particulate concentrations of stack gases for four cold charge tests are summarized in table 7. Stack gas data for the other 10 tests in this group were not obtained because adequate ductwork and an adequate gas sampling location were not available when these tests were conducted. stack gas temperatures ranged from 190.9° to 270.1° F, with flow rates of 1,220 to Particulate concentrations 1,377 scfm. ranged from 0,01 to 0.53 gr/dscf. data indicated the stack gas contained mostly air, with small concentrations of CO and CO2 noted.

TABLE 4. - Stack gas data for tests made with continuously charged preheated scrap

	Sampling	Av	Av gas	Av gas	Moisture	Particu-	Or	sat an	alyse	s,
Test	duration,	temp,	velocity,	flow,	content,	late conc,		vo1	pct	
	min	°F	ft/s	scfm	pct	gr/dscf	CO ₂	02	CO	N ₂
1	40.0	122	31.1	1,783	0.28	0.83	NA	NA	NA	NA
2	40.0	117	23.9	1,382	.09	.03	NA	NA	NA	NA
3	15	129	24.5	1,386	1.09	2.24	0.2	18.4	0.4	81
4	24	120	26.8	1,543	1.23	-41	NA	NA	NA	NA
5	11.2	127	26.5	1,507	1.03	1.96	. 3	20.1	. 4	79.2
6	20.3	123	30.9	1,768	. 90	. 87	. 2	20.4	• 2	79.2
7	20	125	25.8	1,470	1.07	.10	. 4	19.6	. 8	79.2
8	29	101	29.3	1,745	1.03	. 27	NA	NA	NA	NA
9	15	111	29.7	1,733	1.27	•91	. 4	20.2	• 5	78.8
10	30	121	27.1	1,559	1.67	. 27	• 2	20	. 2	79.6
11	17.7	135	28.8	1,613	1.92	. 68	. 7	20.1	. 2	79
12	34.3	115	27.7	1,606	1.66	. 58	. 4	20.6	• 4	78.6

NA Not analyzed.

TABLE 5. - Partial material balance for typical test made with continuously charged preheated scrap

1.0 11.3 25	1. 0 NAp		Ca NAp	Fe	Mg	Mn	P	S	Si	Ti
11.3 25	NAp		NAp							
11.3 25	NAp		NAp			1				
25	_	11 2		NAp	NAp	NAp	NAp	NAp	NAp	NAp
	27.4	11.3	NAp	NAp	NAp	NAp	NAp	NAp	NAp	NAp
	NAp	23.9	0.1	NAp	NAp	NAp	NAp	0.25	NAp	NAp
62	NAp	NAp	41.4	0.1	0.3	NAp	NAp	.02	0.4	NAp
40	.1	NAp	NAp	. 5	NAp	NAp	NAp	NAp	18.2	NAp
1,697.4	14.1	9.5	NAp	1,623.6	NAp	7.3	0.58	.76	14.4	NAp
0.004				30						
10	NAp	. 2	NAp	1	NAp	6.1	.01	NAp	1.7	NAp
7.5	• 2	. 1	NAp	2.9	.1	NAp	NAp	NAp	. 2	1.5
, 854.2	15.4	45	41.5	1,628.1	. 4	13.4	. 59	1.03	34. 9	1.5
, 529.8	4.3	29.2	NAp	1,425.1	NAp	13.6	0.79	0.32	22.4	0.4
302.3	36.9	. 2	29.8	12.8	59	. 9	.07	.30	20.7	3.4
50.3	. 6	NAp	1.4	4.8	2.2	. 3	NAp	NAp	1.3	NAp
, 882.4	41.8	29.4	31.2	1,442.7	61.2	14.8	. 86	. 62	44.4	3.8
101.5	1271.4	65.3	75.2	88.6	(2)	110.5	145.8	60.2	127.2	253.3
Í										
NAp	0.28	0.91	NAp	93.16	NAp	0.89	0.05	0.02	1.46	NAp
NAp	12.2	0.1	9.9	4.2	19.5	0.3	0.02	0.10	6.9	NAp
NAp	1.2	3.6	2.8	9.5	4.4	0.6	NAp	NAp	2.6	NAp
	40 ,697.4 10 7.5 ,854.2 ,529.8 302.3 50.3 ,882.4 101.5	40 .1 ,697.4 14.1 10 NAp 7.5 .2 ,854.2 15.4 ,529.8 4.3 302.3 36.9 50.3 .6 ,882.4 41.8 101.5 1271.4 NAp 0.28 NAp 12.2 NAp 1.2	40 .1 NAp ,697.4 14.1 9.5 10 NAp .2 7.5 .2 .1 ,854.2 15.4 45 ,529.8 4.3 29.2 302.3 36.9 .2 50.3 .6 NAp ,882.4 41.8 29.4 101.5 1271.4 65.3 NAp 0.28 0.91 NAp 12.2 0.1 NAp 1.2 3.6	40	40	40	40	40	40	40

NAp Not applicable.

Alumina refractory material from the eroded furnace roof dissolved in the slag and contributed to the high aluminum accountability.

Metal also contained 0.28 wt pct Cr and 0.78 wt pct Cu.

A material balance including the weights and analyses of the charge materials, products, and furnace dusts is shown in table 8 for a test made with continuously fed cold scrap with natural fluorspar as a slag conditioner. The metal product contained relatively high

levels of copper, aluminum, silicon, and carbon. The slag product contained 0.63 wt pct F, along with 19.2 wt pct Mg and other constituents. Analytical data also indicated the dust contained (wt pct) 0.49 F, 52.6 Zn, 2.11 Pb, and other elements.

^{2&}gt;1,000. Magnesite refractory eroded from the furnace sidewall dissolved in the slag and contributed to the high magnesium accountability.

⁴Dust also contained 2.4 wt pct Pb and 41.5 wt pct Zn.

TABLE 6. - Experimental data for steelmaking tests using continuously charged cold scrap

Test	13	14	15	16	17	18	19
Overall data:	13		13			10	17
Total test timemin.	102.2	104.1	94.7	96.9	108.4	98.0	92.5
Av power inputkW	539	543	569	580	595	581	624
Specific energy consumption	337	743	507	300	373	501	024
1 00 1	896	902	890	932	1,000	862	931
kW·h/st scrap	090	902	090	934	1,000	002	931
Specific electrode consumption	17.0	10 5	16.0	10 1	10 1	16.4	1.2
lb/st scrap	17.3	10.5	16.8	19.1	19.1	16.4	12
Meltdown period:	/05	1.61	F/0	F10	F7(500	50/
Av power inputkW	495	461	540	519	576	580	584
Specific energy consumption	0 //0	0 / 0 0	0 / 10	0.400	0 500	0.507	0 / 00
kW·h/lb scrap	0.462	0.420	0.418	0.482	0.529	0.524	0.489
Continuous-addition period:							
Av scrap feed ratelb/min	33.8	32	34.3	31.5	32.2	40.9	42.3
Av power inputkW	591	582	612	633	633	595	648
Specific energy consumption		k	NO. NO. COS. PART		VIN DAS ADDITION	50% sage als Nr. 14	50 MH 50
kW•h/lb scrap	0.380	0.396	0.408	0.375	0.431	0.365	0.358
Refining period:							
Av power inputkW	395	476	547	585	500	486	610
Tap temp°F	2,975	3,090	3,040	3,075	3,085	3,000	3,245
Products, 1b/st scrap:			1				
Metal	1,925	1,913	1,849	1,858	1,980	1,892	1,922
Slag	159	215	175	241	243	170	345
Stack dust	NAp	NAp	NAp	NAp	NAp	NAp	NAp
Test	20	21	22	- 23	24	25	26
Overall data:							
Total test timemin	92.4	87.0	110.1	97.7	84.6	91.6	96.3
Av power inputkW	566	611	515	608	602	605	F (2
				000	002	603	563
specific energy consumption				000	002	603	563
Specific energy consumption kW•h/st scrap	823	821				860	No. 44 145 1900
kW•h/st scrap	823	821	908	923	819	27.00 SSH-1 6669 93	862
kW•h/st scrap Specific electrode consumption			908	923	819	860	862
kW•h/st scrap Specific electrode consumption lb/st scrap	823 15.5	821 13.9				27.00 SSH-1 6669 93	No. 44 145 1900
kW•h/st scrap Specific electrode consumption lb/st scrap Meltdown period:	15.5	13.9	908	923 14.5	819 12.3	860 14.5	862 15
kW•h/st scrap Specific electrode consumption lb/st scrap Meltdown period: Av power inputkW			908	923	819	860	862
kW•h/st scrap Specific electrode consumption lb/st scrap Meltdown period: Av power inputkW Specific energy consumption	15.5 500	13.9	908 18 527	923 14.5 580	819 12.3 579	860 14.5 550	862 15 438
kW•h/st scrap Specific electrode consumption lb/st scrap Meltdown period: Av power inputkW Specific energy consumption kW•h/lb scrap	15.5	13.9	908	923 14.5	819 12.3	860 14.5	862 15
kW•h/st scrap Specific electrode consumption lb/st scrap Meltdown period: Av power inputkW Specific energy consumption kW•h/lb scrap Continuous-addition period:	15.5 500 0.391	13.9 600 0.542	908 18 527 0•529	923 14.5 580 0.462	819 12.3 579 0.487	860 14.5 550 0.393	862 15 438 0•491
kW•h/st scrap Specific electrode consumption lb/st scrap Meltdown period: Av power inputkW Specific energy consumption kW•h/lb scrap Continuous-addition period: Av scrap feed ratelb/min	15.5 500 0.391 36.9	13.9 600 0.542 63.7	908 18 527 0.529 30.5	923 14.5 580 0.462 31.1	819 12.3 579 0.487 39.6	860 14.5 550 0.393 40.3	862 15 438 0.491 38.8
kW•h/st scrap Specific electrode consumption	15.5 500 0.391	13.9 600 0.542	908 18 527 0•529	923 14.5 580 0.462	819 12.3 579 0.487	860 14.5 550 0.393	862 15 438 0•491
kW•h/st scrap Specific electrode consumption lb/st scrap Meltdown period: Av power inputkW Specific energy consumption kW•h/lb scrap Continuous-addition period: Av scrap feed ratelb/min Av power inputkW Specific energy consumption	15.5 500 0.391 36.9 604	13.9 600 0.542 63.7 627	908 18 527 0.529 30.5 498	923 14.5 580 0.462 31.1 633	819 12.3 579 0.487 39.6 645	860 14.5 550 0.393 40.3 657	862 15 438 0.491 38.8 636
kW•h/st scrap Specific electrode consumption	15.5 500 0.391 36.9	13.9 600 0.542 63.7	908 18 527 0.529 30.5	923 14.5 580 0.462 31.1	819 12.3 579 0.487 39.6	860 14.5 550 0.393 40.3	862 15 438 0.491 38.8
kW•h/st scrap Specific electrode consumption lb/st scrap Meltdown period: Av power inputkW Specific energy consumption kW•h/lb scrap Continuous-addition period: Av scrap feed ratelb/min Av power inputkW Specific energy consumption kW•h/lb scrap Refining period:	15.5 500 0.391 36.9 604 0.356	13.9 600 0.542 63.7 627 0.315	908 18 527 0.529 30.5 498 0.332	923 14.5 580 0.462 31.1 633 0.423	819 12.3 579 0.487 39.6 645 0.319	860 14.5 550 0.393 40.3 657 0.410	862 15 438 0.491 38.8 636 0.376
kW•h/st scrap Specific electrode consumption lb/st scrap Meltdown period: Av power inputkW Specific energy consumption kW•h/lb scrap Continuous-addition period: Av scrap feed ratelb/min Av power inputkW Specific energy consumption kW•h/lb scrap Refining period: Av power inputkW	15.5 500 0.391 36.9 604 0.356 593	13.9 600 0.542 63.7 627 0.315 553	908 18 527 0.529 30.5 498 0.332 546	923 14.5 580 0.462 31.1 633 0.423 480	819 12.3 579 0.487 39.6 645 0.319 492	860 14.5 550 0.393 40.3 657 0.410	862 15 438 0.491 38.8 636 0.376 588
kW•h/st scrap Specific electrode consumption	15.5 500 0.391 36.9 604 0.356	13.9 600 0.542 63.7 627 0.315	908 18 527 0.529 30.5 498 0.332	923 14.5 580 0.462 31.1 633 0.423	819 12.3 579 0.487 39.6 645 0.319	860 14.5 550 0.393 40.3 657 0.410	862 15 438 0.491 38.8 636 0.376 588
kW•h/st scrap Specific electrode consumption lb/st scrap Meltdown period: Av power inputkW Specific energy consumption kW•h/lb scrap Continuous-addition period: Av scrap feed ratelb/min Av power inputkW Specific energy consumption kW•h/lb scrap Refining period: Av power inputkW Tap tempkW Products, lb/st scrap:	15.5 500 0.391 36.9 604 0.356 593 3,055	13.9 600 0.542 63.7 627 0.315 553 2,965	908 18 527 0.529 30.5 498 0.332 546 3,130	923 14.5 580 0.462 31.1 633 0.423 480 3,135	819 12.3 579 0.487 39.6 645 0.319 492 2,815	860 14.5 550 0.393 40.3 657 0.410 369 2,965	862 15 438 0.491 38.8 636 0.376 588 2,840
kW•h/st scrap Specific electrode consumption lb/st scrap Meltdown period: Av power inputkW Specific energy consumption kW•h/lb scrap Continuous-addition period: Av scrap feed ratelb/min Av power inputkW Specific energy consumption kW•h/lb scrap Refining period: Av power inputkW Tap tempkW Products, lb/st scrap: Metal	15.5 500 0.391 36.9 604 0.356 593 3,055 2,009	13.9 600 0.542 63.7 627 0.315 553 2,965 1,916	908 18 527 0.529 30.5 498 0.332 546 3,130 1,848	923 14.5 580 0.462 31.1 633 0.423 480 3,135 1,836	819 12.3 579 0.487 39.6 645 0.319 492 2,815 1,855	860 14.5 550 0.393 40.3 657 0.410 369 2,965 1,714	862 15 438 0.491 38.8 636 0.376 588 2,840 1,910
kW•h/st scrap Specific electrode consumption lb/st scrap Meltdown period: Av power inputkW Specific energy consumption kW•h/lb scrap Continuous-addition period: Av scrap feed ratelb/min Av power inputkW Specific energy consumption kW•h/lb scrap Refining period: Av power inputkW Tap tempkW Products, lb/st scrap:	15.5 500 0.391 36.9 604 0.356 593 3,055	13.9 600 0.542 63.7 627 0.315 553 2,965	908 18 527 0.529 30.5 498 0.332 546 3,130	923 14.5 580 0.462 31.1 633 0.423 480 3,135	819 12.3 579 0.487 39.6 645 0.319 492 2,815	860 14.5 550 0.393 40.3 657 0.410 369 2,965	862 15 438 0.491 38.8 636 0.376 588 2,840

NAp Not applicable. A suitable stack dust collection system was not available at the time.

TABLE 7. - Stack gas data for tests made with continuously charged cold scrap

	Sampling	Av	Av gas	Av gas	Moisture	Particu-	0rs	Orsat analyses,		
Test	duration,	temp,	velocity,	flow,	content,	late conc,		vol pc	t	
W-	min	°F	ft/s	scfm	pct	gr/dscf	CO ₂	02	CO	N ₂
23	20.0	208.0	27.0	1,349	2.10	0.34	0.40	20.0	0.4	75.2
24	37.9	190.9	26.8	1,377	.86	.07	•40	20.6	. 4	78.6
25	25.1	270.1	27.6	1,260	1.35	• 53	NA	NA	NA	NA
26	39	210.4	24.5	1,220	2.15	.01	. 8	19.2	. 2	79.8

NA Not analyzed.

TABLE 8. - Partial material balance for typical test made with continuously fed cold scrap

<i> </i>	Weight	1				Constit	uent				
		A1	C	Ca	F	Fe	Mg	Mn	P	S	Si
Charge, 1b:										3 A St. C. A	
Aluminum	1.0	1.0	NAp	NAp	NAp	NAp	NAp	NAp	NAp	NAp	NAp
Electrodes	10.3	NAp	10.3	NAp	NAp	NAp	NAp	NAp	NAp	NAp	NAp
Fluorspar	5	. 1	.1	2.0	1.7	NAp	NAp	NAp	NAp	0.02	0.2
Coke	25	NAp	23.9	.1	NAp	NAp	NAp	NAp	NAp	. 25	NAp
Lime	62	NAp	NAp	41.4	NAp	0.1	0.3	NAp	NAp	NAp	.4
Quartz	40	.1	NAp	NAp	NAp	.5	NAp	NAp	NAp	NAp	18.2
Shredded scrap	1,800	14.9	10.1	NAp	NAp	1,721.7	NAp	7.7	0.61	.81	15.3
Silico-				_	_		_		1		
mangannese	10	NAp	. 2	NAp	NAp	1	NAp	6.1	.01	NAp	1.7
Total	1,953.3		44.6	43.5	1.7			13.8	.62	1.08	
Products, 1b:											
Metal	1,669.9	3.1	32.7	NAp	NAp	1,571.4	NAp	14.4	0.80	0.41	20.1
Slag	269	28.2	. 1	30.5	1.7	9.9		• 7	.03	.18	20.5
Dust	20.5	. 2	. 2	. 4	. 1.	1.3	. 7	NAp	NAp	NAp	. 3
Total	1, 959.4	31.5	33	30.9	1.8	1,582.6	52.3		.83	. 59	40.9
Accountability	1										
pct	100.3	1195.7	74	71	105.9	91.8	(²)	109.4	133.9	54.6	114.3
Product analy-											
ses, wt pct:	ļ										
Metal ³	NAp	0.19	1.96	NAp	NAp	94.10	NAp	0.86	0.05	0.02	1.20
Slag	NAp	10.48	0.04	11.34	0.63	3.68	19.2	0.26	0.01	0.07	7.62
Dust 4	NAp	0.98	0.98	1.95	0.49	6.34	3.41	NAp	NAp	NAp	1.46

NAp Not applicable.

Alumina refractory material from the eroded furnace roof dissolved in the slag and contributed to the high aluminum accountability.

^{2&}gt;1,000. Magnesite refractory eroded from the furnace sidewall dissolved in the slag and contributed to the high magnesium accountability.

Metal also contained 0.33 wt pct Cr and 0.69 wt pct Cu.

³Dust also contained 2.11 wt pct Pb and 52.6 wt pct Zn.

SCRAP BACKCHARGING TESTS

The experimental data for the three scrap backcharge tests are shown in table 9. In these tests, the power input ranged from 533 to 579 kW for the melt-down of the 1,800-1b scrap charges. Electrical energy consumption ranged from 612 to 658 kW h per short ton of charge. Electrode consumption of 10.2 lb or less per short ton of scrap was noted in these tests.

The data for the stack gas measurements and particulate concentrations are summarized in table 10. The stack gas having temperatures of 126° to 146° F flowed at a rate of 1,385 to 1,490 scfm. Particulate concentration ranged from 0.11 to 0.64 gr/dscf, and the moisture content of the gases was 1.31 to 1.75 wt pct.

Table 11 shows a material balance for a test made with backcharged scrap consisting mainly of shredded automotive body steel and Girstley borate as a slag conditioner. The metal product contained (wt pct) 0.22 Cu, 0.73 C, and 0.001 B, along with other constituents. The slag

product contained 15.1 pct Mg, and other elements as shown in table 11. The dust contained (wt pct) 0.03 B, along with 32.5 Zn, 1.99 Pb, and other constituents.

SLAG CONDITIONERS

Of the three types of slag conditioners evaluated on the basis of visual observations, the Girstley borate and B203 group appeared the most effective in increasing the fluidity of the individual slags. resultant slags contained up to 0.5 wt pct B, while the tapped metal prodicts contained 0.001 to 0.002 wt pct B. Boron at these levels is not considered a harmful constituent in most low- or medium-carbon steels. The constituent markedly increases the hardenability of steels at higher levels approaching 0.007 The fluorine-containing materiwt pct. als were also effective in increasing In the tests made with slag fluidity. natural and synthetic fluorspar slag conditioners, the tapped metal products contained between 0.03 and 0.05 wt pct P.

TABLE 9. - Experimental data for steelmaking tests using backcharged scrap

Test	27	28	29
Overall data:			
Total test timemin	82.2	79.8	77.2
Av power inputkW	575	579	553
Specific energy consumptionkW*h/st scrap	640	658	612
Specific electrode consumptionlb/st scrap	8.6	8.6	10.2
Meltdown period:			
Av power inputkW	575	579	533
Specific energy consumptionkW.h/1b scrap	0.320	0.329	0.306
Refining period:			
Av power inputkW	520	467	567
Tap temperature°F	2,985	2,915	2,775
Products, 1b/st scrap:			
Metal	1,908	1,920	1,750
Slag	147	145	127
Stack dust	12	8	22

TABLE 10. - Stack gas data for tests made with scrap backcharges

	Sampling	Av	Av gas	Av gas	Moisture	Particu-	Orsat analyses,				
Test	duration,	temp,	velocity,	flow, content,		late conc,	vol pct				
	min	°F	ft/s	scfm	wt pct	gr/dscf	CO ₂	02	CO	N ₂	
27	18.5	146	27.0	1,490	1.31	0.11	0.4	19.2	0.4	80.0	
28	19.5	130	24.5	1.385	1.47	.11	.5	19	.3	80.2	
29	17	126	25	1,484	1.75	•64	. 2	20	.6	79.2	

TABLE 11. - Partial material balance for typical test made with backcharged scrap.

	Weight		Constituent								
		A1	В	C	Ca	Fe	Mg	Mn	P	S	Si
Charge, 1b:			Ī								
Aluminum	1.0	1.0	NAp	NAp	NAp	NAp	NAp	NAp	NAp	NAp	NAp
Electrodes	7.4	NAp	NAp	7.4	NAp	NAp	NAp	NAp	NAp	NAp	NAp
Girstley borate	11.4	.1	1.1	NAp	1.5	NAp	0.2	NAp	NAp	NAp	0.5
Coke	25		NAp	23.9	. 1	NAp	NAp	NAp	NAp	0.25	NAp
Lime	62		NAp	NAp	41.4	0.1	. 3	NAp	NAp	. 02	. 4
Quartz	40	. 1	NAp	NAp	NAp	. 5	NAp	NAp	NAp	NAp	18.2
Shredded scrap	1,800	23.6	.06	20	NAp	1,696.9	NAp	8.6	0.85	.70	20.9
Silicomanganese	10	NAp	NAp	. 2	NAp	1	NAp	6.1	.01	NAp	1.7
Total	1,956.8	24.8	1.16	51.5	43	1,698.5	• 5	14.7	. 86	. 97	41.7
Products, 1b:											
Metal	1,728	0.6		12.7	NAp		NAp			0.58	1.73
Slag	130.6	7.5	. 65	. 1	19.3	10.7	19.7	8.1	.03	.09	12.6
Dust	7.2	• 5	.02	. 1	. 1	1.4	. 1	NAp	NAp	NAp	.1
Total	1,865.8	8.2	. 69	12.9	19.4	1,708.9	19.8	14.2	0.24	0.67	14.4
Accountability											9
pct	95.4	33.1	59.5	25.1	45.1	100.6	(1)	96.6	27.9	69.1	34.6
Product analyses,											
wt pct:	}										
Metal ²	NAp	0.03	0.001	0.73	NAp	98.19	NAp	0.35	0.01	0.03	0.10
S1ag	NAp	5.74	0.5	0.08	14.8	8.19	15.1	6.2	0.02	0.07	9.7
Dust 3	NAp	0.69	0.03	0.69	1.11	19.9	1.39	NAp	NAp	NAp	1.25

NA Not applicable.

The synthetic fluorspar made from a phosphate plant waste was not a source of phosphorus contamination in the tapped metal products. The sodium constituent in aluminum potlining waste presented a slight fuming problem during scrap meltdown. In the third group, Sorelflux B

was the least effective under the conditions used. Only visual comparisons can be given since slag viscosity data were Moreover, the slag componot obtained. sitions varied from test to test owing to erosion of furnace lining material.

DISCUSSION OF RESULTS

Comparisons of the pertinent averaged shredded scrap charging experiments are shown in table 12. In the continuousportion of the addition tests, the

electrical energy consumption averaged experimental data for the three types of 0.355 kW h per pound of preheated charge and 0.375 kW h per pound of cold charge, using average power inputs ranging from 515 to 624 and 463 to 634 kW,

 $^{^{1}}$ > 1, 000. Magnesite refractory eroded from the furnace sidewall dissolved in the slag and contributed to the high magnesium accountability.

²Metal also contained 0.14 wt pct Cr and 0.22 wt pct Cu.

³Dust also contained 1.99 wt pct Pb and 32.5 wt pct Zn.

respectively. The data indicated that preheated scrap continuously fed at rates averaging 22.9 to 66.2 lb/min consumed 5.3 pct less electrical energy than did cold scrap fed at 30.5 to 63.7 lb/min. The wide variation in feed rates from test to test was caused by problems encountered with frequent hangup of shredded scrap in the feed bin. Tests with continuously fed preheated and cold scrap charges consumed overall 829 and 888 kW·h per short ton of scrap, respectively. Overall, the preheated scrap charges consumed 6.6 pct less electrical energy than did the cold-charged scrap. Figure 6 shows the actual electrical energy consumed in the continuous addition, meltdown, and refining periods of the two types of scrap charging tests, along with tests for conventional backcharged scrap. It can be readily seen that both types of continuous-charging tests consumed more electrical energy than did the test made conventional scrap backcharges. Higher energy consumption was attributed to the inability to continuously charge

at a rate fast enough to use all of the available energy. This explanation is supported by the open-bath conditions and the higher bath temperature observed during continuous charging. No significant differences were noted in the quantity of metal produced in the three types of From 92.9 to 94.3 pct of the tests. scrap charged was recovered as tapped metal.

Analytical data from backcharge tests indicated the furnace dust contained less zinc (32.5 wt pct) than the dusts derived from the continuous feeding charges. Both types of continuous-charging tests averaged approximately 100 lb more slag than did tests made with conventional scrap backcharges.

Indications were that this large quantity of slag was due to dissolved refractory material, principally from the magnesite lining. Observations made during the tests revealed that use of open-bath conditions during the continuous-feeding period caused part of the arc energy to be radiated across the top of the bath to

Table 12. - Comparisons of averaged experimental data for steelmaking tests using three types of scrap charging practices

	Continu	ous	Continu	ious	Backcha	rge	
	prehea	ited	cold ch	arge	only		
	charg	je					
	Average	Std	Average	Std	Average	Std	
		dev		dev		dev	
Overall data:							
Total test timemin	88.4	10.1	96.9	7.4	79.7	2.5	
Av power inputkW	579	31	577	51	569	14	
Specific energy consumption							
kW•h/st scrap	829	43	888	51	637	23	
Specific electrode consumption							
lb/st scrap	15.6	3.2	15.1	2.4	9.1	0.9	
Meldtown period:							
Av power input, kW	538	61	539	49	574	24	
Continuous addition period:		70%					
Av scrap feed ratelb/min	43.7	11	37.5	9	NAp	NAp	
Av power inputkW	581	6	615	30	NAp	NAp	
Specific energy consumption			600 000 200		1		
kW•h/1b scrap	0.355	0.032	0.375	0.037	NAp	NAp	
Products, 1b/st scrap:					•		
Metal	1,861	52	1,887	71	1,859	96	
Slag	245	66	241	70	140	12	
Stack dust	54	27	NAp	NAp	13	6	
Metal recoverypct	93	NAp	94.3	NAp	92.9	NAp	
NAp Not applicable.	•						

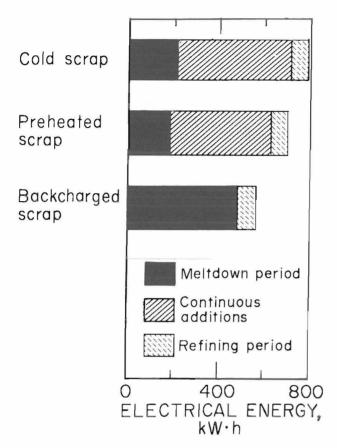


FIGURE 6.—Actual electrical consumption for the three types of scrap charging tests.

the furnace sidewall. This resulted in excessive energy consumption at the maximum power inputs used, as well as considerable refractory consumption. slags from the continuous-feeding tests had magnesium levels averaging 15.5 pct cold charges and 16.6 pct for prefor Earlier investigators heated charges. experienced similar problems with excessive energy and refractory consumption when continuously feeding charge materials such as prereduced iron into an electric arc furnace (37-38). In the case of conventional scrap backcharge practice, the furnace sidewall is shielded most of the time by the unmelted scrap. In the absence of an open bath, less of the arc energy is radiated to the sidewall.

It became apparent that the scrap feed rate was not balanced with the high power input used. The scrap charging and preheating equipment was not large enough to feed the scrap at a rate corresponding to the power input used. One suggested

technique to address this problem is choke feeding, whereby the scrap feed rate is sufficient to allow an accumulation of scrap in the furnace to prevent open-bath conditions. The choke feeding technique used for prereduced iron ore pellets is reported to result in higher melt rates and increased productivity (39). Another technique to employ during continuous feeding is use of a foamy slag (40). The radiation of arc energy to the sidewalls is reduced since the tips of the electrodes are surrounded by scrap or In the present case, the a foamy slag. equipment and electric arc furnace were not large enough to permit testing of either technique.

An alternate procedure was tested to demonstrate that energy consumption could be decreased during continuous feeding of scrap into the 1-st furnace. continuous-feeding period was started during the meltdown of the initial 450-1b charge. In this manner, the preheated scrap was fed at a faster rate of 49.6 lb/min, with a power input 587 kW. The overall energy consumption of 698 kW·h per short ton of charge was 15.8 pct lower than the 829 kW h per short ton of charge shown in table 12. Energy consumption in the meltdown of preheated scrap with the modified procedure was 0.306 kW h per pound of charge.

Some differences in stack gas temperatures and flows were observed between the three types of scrap charging tests shown in table 13. The stack gas of preheated scrap tests exhibited the lowest temperatures (about 120° F) and the highest flow (1,615 scfm), while that of tests using continuously fed cold scrap averaged 220° F and flowed at a rate of 1,302 scfm. In the scrap backcharge tests, the stack gas temperature averaged 135° F and the gas flow rate averaged 1.433 scfm. Stack gas in the preheating tests was of lower temperature because heat was absorbed by the scrap fed into the preheater unit. The higher gas flow was attributed to the ingress of air into the open scrap feed inlet of the prethe continuous-addition heater during Stack gas from the preheated period. scrap tests also contained the highest concentrations of particulate

TABLE 13	Averaged	data	for	stack	gas	measurements	of	the	three	types	of	scrap
charging t	ests											

	Continu	ous	Continu	ious	Backcha	_	
	prehea	ted	cold ch	arge	only		
	charg	ge					
	Average	Std	Average	Std	Average	Std	
		dev		dev		dev	
Sampling durationmin.	21.6	8.4	33.0	8.8	18.3	1.3	
Temperature of gas°F	120	11	220	35	135	10	
Gas velocityft/s	28	2	26.5	1.4	25.5	1.3	
Flow ratescfm.	1,615	132	1,302	73.4	1,433	53.2	
Moisture contentwt pct	1.3	0.4	1.6	0.6	1.5	0.2	
Particulate concgr/dscf	0.97	0.74	0.24	0.24	0.29	0.3	
Orsat analyses, vol pct:							
CO ₂	0.3	0.2	0.5	0.2	0.4	0.2	
02	20	0.7	19.9	0.7	19.4	0.5	
co	0.3	0.2	0.3	0.1	0.4	0.2	
N ₂	79.4	0.8	79.3	0.6	79.8	0.5	

(0.97 gr/dscf). A portion of the organic material (such as upholstery), iron oxide scale, and other materials that mechanically separated from the continuously fed scrap as it tumbled in the rotating preheater became entrained as particulate in the furnace offgas stream. In the other type of test, the scrap was charged directly into the furnace, and part of the dirt, glass, etc., reported to the slag phase.

The EPA sampling procedures used in the meltdown periods of the tests were established for sampling stack gases such as those from coal-fired electric power-plants that operate continuously under

steady-state conditions. In the scrap meltdown period, gas sampling under steady-state conditions is difficult owing to the relatively short sampling times available.

It was impossible to make a quantitative evaluation of the seven different slag conditioners used because the chemical compositions of the steelmaking slags varied considerably owing to dissolved refractory materials from the magnesite lining. As previously mentioned, the furnace refractories suffered considerable damage due to open-bath conditions.

SUMMARY AND CONCLUSIONS

Results of this investigation showed that continuously fed fragmented scrap preheated by furnace offgases consumed about 7 pct less electrical energy overall than did similarly charged cold Meltdown of scrap preheated to 840° to 1,110° F in an electric arc furnace consumed 0.355 kW h per pound charge, compared with 0.375 kW h using cold scrap. Data indicated the preheated and cold charges of fragmented scrap continuously fed into the furnace consumed more electrical energy and electrode materials than did conventional backcharged scrap, which

0.318 kW.h per pound of charge. The higher energy consumption is not necessarily characteristic of continuous feeding but may be due to the way the testing was conducted. The procedure used to feed scrap into the bath resulted in high refractory consumptions, with the resultant slags having magnesium levels as high as 16.6 pct.

Three groups of slag conditioners that contained boron, fluorine, or titanium were used as substitutes for natural fluorspar. The boron—and fluorine—containing additives were effective in increasing the fluidity of the slags.

Use of the boron-containing conditioners did not increase the boron levels in the meltdown metal. The tapped metal products contained 0.001 to 0.002 wt pct B.

The stack gases from the scrap melt-downs were measured and sampled, and the results were compared. Significant differences in stack gas temperatures,

flows, and particulate concentrations were noted between the three types of scrap charging tests. Stack gases from scrap preheating tests exhibited the highest flow rate, which averaged 1,615 scfm, and the lowest temperature, which averaged about 120° F.

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APPENDIX. --EQUATIONS USED TO PROCESS DATA FOR STACK GAS MEASUREMENTS

The sample volume measured by the dry gas meter was corrected to standard conditions, 70° F and 29.92 in Hg, by equation A-1:

$$V_{m_{s+d}} = V_m \left(\frac{T_{s+d}}{T_m}\right) \frac{P_{bar} + \frac{\Delta H}{13.6}}{P_{s+d}}$$
 (A·1)

where Vmstd = volume of gas sample through the dry gas meter, scf,

Vm = volume of gas sample through the dry gas meter (meter conditions), ft³,

Tstd = absolute temperature at standard conditions, 530° R,

Tm = average dry gas meter temperature, °R,

Pbar = barometric pressure at the orifice, in Hg,

ΔH = average pressure drop across the orifice meter, in H₂O,

13.6 = specific gravity of mercury,

and Pstd = absolute pressure at standard condition, 29.92 in Hg.

The volume of water vapor was calculated by equation A-2:

$$V_{w_{std}} = V_{lc} \left[\left(\frac{\rho_{H_20}}{M_{H_20}} \right) \left(\frac{RTstd}{Pstd} \right) \right],$$
 (A-2)

where Vwstd = volume of water vapor in the gas sample, scf,

V_{|c} = total volume of liquid collected in impingers and silica gel, mL,

 $\rho_{\rm H_2O}$ = density of water, 1 g/mL,

 $M_{\rm H_2O}$ = molecular weight of water, 18 lb/lb mol,

and R = ideal gas constant (1.83 in Hg - ft³)/(1b mol - °R).

The percent of moisture content was calculated by equation A-3:

$$B_{WO} = \frac{V_{W_{s+d}}}{V_{m_{s+d}} + V_{W_{s+d}}} (100), \qquad (A-3)$$

where Bwo = water vapor in the gas stream, vol pct.

Concentration of particulate was given by equation A-4:

$$Cs = 0.0154 \frac{gr}{mg} \left[\frac{Mn}{Vm_{s+d}} \right], \qquad (A-4)$$

where Cs = concentration of particulate matter in stack gas, gr/dscf,

Mn = total amount of particulate matter collected, mg,

gr = grains of particulate,

and mg = milligrams of particulate.

Isokinetic variation was calculated by equation A-5:

$$I = \frac{Ts \left[\frac{(V_{1c})(\rho_{H_2O})(R)}{M_{H_2O}} + \frac{Vm}{Tm} \left(Pbar + \frac{\Delta H}{13.6} \right) \right]}{\theta Vs Ps An}$$
(A-5)

where I = isokinetic sampling, pct,

Ts = absolute average stack gas temperature, °R,

 θ = total sampling time, min,

Vs = stack gas velocity, ft/s,

Ps = absolute stack gas pressure, in Hg,

and An = cross-sectional area of nozzle, ft2.

For the test data to be acceptable, the isokinetic sampling must be within the 90- to 110-pct range.