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## Ferronickel shot — Sampling for analysis

*Ferro-nickel en grenailles — Échantillonnage pour analyse*

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## Foreword

ISO (the International Organization for Standardization) is a worldwide federation of national standards bodies (ISO member bodies). The work of preparing International Standards is normally carried out through ISO technical committees. Each member body interested in a subject for which a technical committee has been established has the right to be represented on that committee. International organizations, governmental and non-governmental, in liaison with ISO, also take part in the work. ISO collaborates closely with the International Electrotechnical Commission (IEC) on all matters of electrotechnical standardization.

Draft International Standards adopted by the technical committees are circulated to the member bodies for approval before their acceptance as International Standards by the ISO Council. They are approved in accordance with ISO procedures requiring at least 75 % approval by the member bodies voting.

International Standard ISO 8049 was prepared by Technical Committee ISO/TC 155, *Nickel and nickel alloys*.

Annexes A, B and C of this International Standard are for information only.

# Ferronickel shot — Sampling for analysis

## 1 Scope

This International Standard defines a method of sampling for analysis of ferronickel lots in the form of shot as specified in ISO 6501 in those cases where lots are constituted either heat by heat or by taking from blended stock.

The purpose is to determine the contents of the various elements :

- either from slugs by physical analysis methods (such as X-ray fluorescence or emission spectral analysis);
- or from chips by dry methods (carbon, sulfur) or chemical analysis (other elements).

## 2 Normative references

The following standards contain provisions which, through reference in this text, constitute provisions of this International Standard. At the time of publication, the editions indicated were valid. All standards are subject to revision, and parties to agreements based on this International Standard are encouraged to investigate the possibility of applying the most recent editions of the standards listed below. Members of IEC and ISO maintain registers of currently valid International Standards.

ISO 513 : 1975, *Application of carbides for machining by chip removal — Designation of the main groups of chip removal and groups of application.*

ISO 3855 : 1977, *Milling cutters — Nomenclature.*

ISO 4957 : 1980, *Tool steels.*

ISO 6352 : 1985, *Ferronickel — Determination of nickel content — Dimethylglyoxime gravimetric method.*

ISO 6501 : 1988, *Ferronickel — Specification and delivery requirements.*

## 3 Form and packaging of product

Grain size : between 2 and 50 mm

Lot tonnage : equal to or greater than 5 t

In the case of lots taken from blended stock, the nickel content range  $k$  to  $(k + n)$  % of the blended heats shall be chosen as

$$\begin{aligned} 15 &\leq k \leq 59 \\ 1 &\leq n \leq 5 \\ 16 &\leq k + n \leq 60^{1)} \end{aligned}$$

The ferronickel shot is generally delivered in bulk form in units which may be trucks, containers, or railroad cars, of which the contained masses normally range from 5 to 30 t, although in the case of railroad cars, loads may have masses up to 60 t.

This type of ferronickel can also be delivered drum-packed (the contained mass of which may be 250 kg).

## 4 Principle

In a single heat, intergrain homogeneity is practically ensured. It is therefore very easy to obtain a representative "primary sample" from a small number of "primary increments".

In the case of a blended lot composed of several heats, a greater number of primary increments  $N_p$  has to be taken, but the whole still constitutes the primary sample.

After blending and mass division of the primary sample, an "intermediate sample" is obtained having a reasonable mass for laboratory treatment. The treatment of the intermediate sample gives a "secondary sample", which may be divided in  $N_s$  "secondary increments" not exceeding a mass of 1 kg individually. Each secondary increment is then remelted under appropriate conditions so that no variation in composition can be observed and that  $N_s$  homogeneous small ingots<sup>2)</sup> be obtained (within-small-ingot homogeneity).

1) The case of non-blended lots (case  $n \leq 1$ ) is not dealt with in this International Standard.

2) It is generally accepted that 1 kg is the maximum mass which can be accommodated in a laboratory furnace for re-casting under the required conditions. According to the grain size distribution of shot, it is often necessary for the secondary sample to exceed 1 kg in order to be representative. Hence the necessity of melting several small ingots. See the statistical justification in annex A.

The small ingots are then either used for physico-chemical analysis or machined into chips for chemical analysis. (This procedure is summed up in figure A.1 of annex A.)

## 5 Taking of the primary sample and then of the intermediate sample

### 5.1 Blended lots

#### 5.1.1 Bulk sampling in the case of a suitable system for taking the primary sample

This can be performed, for example, by emptying the shot into a bin with reclaim by belt conveyor. From the conveyor discharge, two possibilities are

- to have a true sampling system respecting the rules of the art for sampling of particulate material (such as a cross-stream sampler);
- to take increments at regularly spaced intervals, using a power shovel with a dipper intercepting the shot stream in a representative manner.

The mass of each primary increment shall be, in this case, not less than 20 kg, and is generally between 20 and 50 kg.

The number of primary increments  $N_p$  to be selected is shown in table 1.

The primary sample shall then be mass-divided into smaller units, in order to obtain an intermediate sample having a mass which can reasonably be sent to the laboratory for further preparation, 20 to 50 kg, say.

This can be accomplished with automatic mass dividers (such as rotary dividers) of suitable size with respect to the particle size of the product being handled.

Failing such equipment, the division can be made by alternate shovelling from the primary sample stockpile. As a precaution

against material spill during shovelling, it is recommended that a scoop or coal-miner's-type shovel be used.

For example, every fifth shovelful or less would be taken and this division would be repeated a sufficient number of times to obtain the desired sample mass of 20 to 50 kg.

#### 5.1.2 Sampling of bulk material when no adequate primary sampling system is available

In this case, hand sampling shall be performed by alternate shovelling on each unit to be checked (truck, railroad car, container, etc.). The number of units to be checked is the number  $N_p$  in table 1 or the total number of units if it is less than  $N_p$ .<sup>1)</sup>

For example, when unloading a 20 t truck on to the ground, sampling could proceed as follows :

- Shovel the 20 t, setting aside every fifth shovelful.
- From the 4 t obtained, set aside every fifth shovelful.
- From the 800 kg obtained, set aside every fifth shovelful.
- From the 160 kg obtained, set aside every fifth shovelful.
- Send the 32 kg obtained to the laboratory.

In this example, an intermediate sample is obtained for the checked unit.

If more than one unit is checked in the same lot, intermediate samples in each unit can be blended and mass division carried out again until an intermediate sample representative of the lot is obtained. In this case, the intermediate sample mass can be reduced to 10 to 20 kg.

#### 5.1.3 Sampling of a drum-packed lot

The number of drums from which increments have to be taken is the number  $N_p$  in table 1 or the total number of drums if this is less than  $N_p$ .<sup>2)</sup>

Table 1 — Minimum number of primary increments to be selected

Numbers of primary increments $N_p$	Tonnage	Range of nickel contents $n$				
		$n < 1$	$1 < n < 2$	$2 < n < 3$	$3 < n < 4$	$4 < n < 5$
	5 to 50	5	10	15	20	30
	50 to 200	7	12	17	22	35
	200 to 500	10	15	20	25	40
	500 to 2 500	15	20	25	30	45
Number of secondary increments $N_s$ <sup>1)</sup>		1	2	3	4	5

1) This indicates the number of small ingots to be remelted in the hypothesis of 1 kg per small ingot. (If the maximum mass which can be remelted is  $1/x$  kg, the number of small ingots to be remelted is  $x \cdot N_s$ .)

1) For this purpose, the rules for random sampling given in annex B can be applied.

2) In general, drum-packaging is used for low-tonnage lots. The first line of the table is therefore applicable in most cases.

A minimum of 1 kg of shot or more, if required, per selected drum is taken to obtain a mass in excess of 20 kg, generally between 20 and 50 kg.

If the contents of each drum are assumed to be homogeneous, the sample may be taken from the top of the drum. If not, the drums shall be emptied and the sample taken by alternate shovelling.

## 5.2 Particular case of a lot made up of one single heat

As inter-grain homogeneity is ensured, it is sufficient to take the minimum quantity of material for small ingot remelting (1 kg for example).

To have more adequate guarantee, a small number of primary samples, for example 3 to 5, can be taken (either by bulk sampling or sampling from drums), then blended and mass-divided in order to obtain an intermediate sample of 5 to 10 kg.

If the lot is not assumed to be made up of a single heat, one of the procedures described in 5.1 shall be applied.

## 6 Treatment of the intermediate sample and taking of the secondary sample

This is generally carried out in the laboratory sampling shop.

### 6.1 Blended lot

The intermediate sample is blended, then mass-divided, preferably using a riffle divider of appropriate dimensions<sup>1)</sup> or, failing this, by alternate shovelling, until a mass equal to or slightly exceeding the mass, in kilograms, in the last line of table 1 is obtained. In the table,  $N_s$  is the number of small ingots to be remelted when 1 kg of material can be melted in one operation. (If melting is achieved by masses of  $1/x$  kg, the number of small ingots to be remelted is  $x \cdot N_s$ .)

The mass defined by the rule above is the mass to be remelted and to be used for representative analysis. If a sampling reject or second unmelted secondary sample is to be kept, the corresponding quantity of material shall be set aside at the time of mass divisions.

### 6.2 Lot made up of a single heat

To be representative, a small ingot having a mass of 250 to 1 000 g shall be obtained. This is obtained by blending and mass division of the intermediate sample made in accordance with 5.2 until the mass required for remelting is obtained.

## 7 Remelting of the secondary sample

Remelting shall be performed in conditions such that no variation in content (of either Ni or the impurities to be checked) occurs either during melting or casting of the final sample (slugs, rondelles or small ingots).

In practice, the melting shall be done by induction heating in order to be carried out rapidly; it generally requires argon protection. The melted sample can be cooled and solidified in the melting pot itself, provided that argon protection is provided. However, it is much better to cast after melting by centrifuging. This ensures

- an excellent homogeneity of the sample produced as a result of mixing the molten metal during its injection into the mould;
- a uniform crystalline structure which fosters a good repeatability of the measurements for physical analysis methods. The argon protection should preferably be maintained during centrifuging.

It is recommended that a reagent (such as aluminium chips in a proportion of 1 to 2 g/kg) be introduced to kill the shot to be remelted. Naturally, the dilution undergone by the sample can be taken into account to correct the nickel content found during final analysis.

## 8 Use of small ingots (secondary increments)

**8.1** The small ingots produced are truncated near their base to obtain a slice having a thickness of some 15 to 20 mm.

The slices obtained can be used for physical analysis and the average value of the analyses is calculated.

**8.2** It is also possible to take chips by drilling or milling on the remaining parts of the small ingots. Chips coming from all the small ingots are conditioned for analysis by dry methods (sulfur and carbon) or chemical analysis (for the other elements).

### 8.2.1 Precautions for chip machining

Machining (and preferably milling) shall be carried out in such a way that chips cannot be contaminated (either by cutting tool wear or by dust or grease). In particular, the work shall be carried out under dry conditions.

For the detailed technical conditions of machining, see annex C.

Some ferronickel types are very hard, hence the need to select appropriate cutting tools and cutting conditions with great care.

Machining will generally be easier if the small ingot is previously annealed.

### 8.2.2 Treatment of chips

#### 8.2.2.1 Washing

When surface contamination of chips (by lubricants, dust, etc., inevitably present when working with machine tools) is feared,

1) The colander width shall be at least three times the mean diameter of the largest shot.

it is strongly recommended that the chips be washed twice in pure acetone (or once in pure acetone and once in pure ether).

The solvent is drained off. Residual solvent is then evaporated in the air and the sample is dried for a minimum of 0,5 h in an oven maintained at 100 to 110 °C.<sup>1)</sup>

#### 8.2.2.2 Crushing

If chips come from a single small ingot, due to the fact that cast small ingots are very homogeneous, it is not necessary to crush the chips.<sup>2)</sup>

If several small ingots have been cast it is useful, when possible, to crush the chips in order to achieve homogeneity between the chips from various small ingots.

In practice, crushability depends on

- the nickel content : if it exceeds 35 %, the alloy becomes ductile and is difficult to crush;
- the impurity contents (above all carbon) : high-carbon ferronickels can be crushed much finer than low-carbon ferronickels.

In the case of crushable ferronickels, a suitable crusher shall be used which does not introduce contamination with iron. Vibration mill laboratory crushers used for a duration of 10 to 30 s are suitable. It is desirable that the crushing container be of tungsten carbide or, if this is not possible, of special anti-wear steel (ball-type or bar-type crushers are not permissible).

In the case of ferronickels having nickel contents less than 35 %, 30 s crushing gives such fine material that almost all can be considered as undersize in case of sieving

— on a sieve having a 2,5 mm aperture size (8 mesh), for low-carbon ferronickel (LC);

— on a sieve having a 0,8 mm aperture size (20 mesh) for medium- and high-carbon ferronickels (MC and HC).

#### 8.2.2.3 Homogenization and bottling

When the chips derive from several small ingots, it is necessary to achieve homogenization (using a mechanical homogenizer or repeated alternative shovelling, or several passes through a riffle divider keeping all the material, etc.).

The sample shall be subdivided in several portions using a riffle divider or a sample distributor. The number of fractions will depend on the required number of test samples for analysis to be kept by the interested parties.

The minimum distribution shall be

- 1 for the purchaser,
- 1 for the vendor,
- 1 for the referee,
- 1 reserved.

For low-carbon ferronickels (LC), all handling operations shall be carried out so that no carbon contamination can occur (no contact with paper, cardboard, rubber, cork or plastics; metallic materials and aluminium foils can be used).

The same care shall be exercised for bottling.

For medium- and high-carbon ferronickels (MC and HC), samples can be stored in bottles of, for example, glass or aluminium, or in thick, heavy-quality polyethylene bags.

1) The use of pure organic solvents and their utmost removal is required for later determination of carbon and sulfur with automatic devices according to dry instrumental techniques.

2) This is all the more valid the finer the chips. Millings are finer than drillings.

## Annex A (informative)

### Justification of the number of primary and secondary increments

#### A.1 General

The reasoning below applies to blended lots. The definition of the product is given in clause 3 of this International Standard.

The adopted procedure is derived from the following preliminary considerations :

- a) excellent homogeneity within a granulated heat. No content variation is detected (for nickel and the various impurities : carbon, cobalt, chromium, sulfur, silicon, ...) either among granules of the same particle size range or among the various particle size ranges within the same heat;
- b) particle size distributions may vary considerably from one heat to another in a blended lot;
- c) it is possible to remelt ferronickel shot under argon without varying the content of nickel, cobalt, chromium, silicon, sulfur, ... However, slight reductions in carbon contents have been observed.

In practice, the maximum known capacity of remelting furnaces is 1 kg and the numerical values of  $N_s$  in table 1 have been chosen on this basis.

The study has been carried out mainly on nickel contents, which are the figures on which the greatest accuracy is sought.

#### A.2 Sampling scheme

The general principle adopted is summed up in figure A.1.

The following notation is used :

$N_p$  is the number of primary increments.

$V_p$  is the primary variance estimating the scatter of nickel contents among primary increments.

This variance therefore gives an estimate of the slight heterogeneity which may be observed in the whole lot when it is discharged at the purchaser's.

This quantity is by nature a variance of integration (zero if homogeneity is perfect among primary increments). It is generally not computable from a theoretical scheme. Only empirical observations can be made when the lots are discharged.

$N_s$  is the number of secondary increments. This is the number of remelted small ingots.

$V_s$  is the secondary variance estimating the scatter of nickel contents among secondary increments (mass < 1 kg).

The fact that the material is made up of separate shot particles and that the heats selected within a defined range of nickel contents are blended means that a minimum mass of blended material must be kept for the secondary sample to be representative.

This quantity is by nature a fundamental variance [which always exists in any fragmented material undergoing content variations from fragments to fragments, even if homogenization (or blending) is perfect].

There are mathematical models for calculating this quantity. Their application to ferronickel lots generally implies keeping a mass in excess of 1 kg. This is the main reason why several small ingots are remelted and why  $N_s$  and  $V_s$  are to be evaluated.

$V_e$  is the sampling variance.

We have

$$V_e = \frac{V_p}{N_p} + \frac{V_s}{N_s} \quad \dots (1)$$

NOTE — The intermediate sample is not taken into consideration in the procedure principle. It is only one sample among all the others handled during blending and mass-division operations. It is to be selected with a reasonable mass for transportation between the primary sampling location at the works and the laboratory.

If one analysis is carried out per remelted sample ingot,

$$V_{Ni} = \frac{V_p}{N_p} + \frac{V_s}{N_s} + \frac{V_r}{N_s} + V_A + V_L \quad \dots (2)$$

where

$V_{Ni}$  is the variance of nickel content over all "sampling and analysis" operations when several laboratories are compared;

$V_r$  is the variance of analytical repeatability<sup>1)</sup>;

$V_A$  is the variance of inter-analyst fluctuation (analytical)<sup>1)</sup>;

$V_L$  is the variance of inter-laboratory fluctuation (analytical)<sup>1)</sup>.

1) For the detailed significance of  $V_r$ ,  $V_A$ ,  $V_L$ , see annex B of ISO 6352.

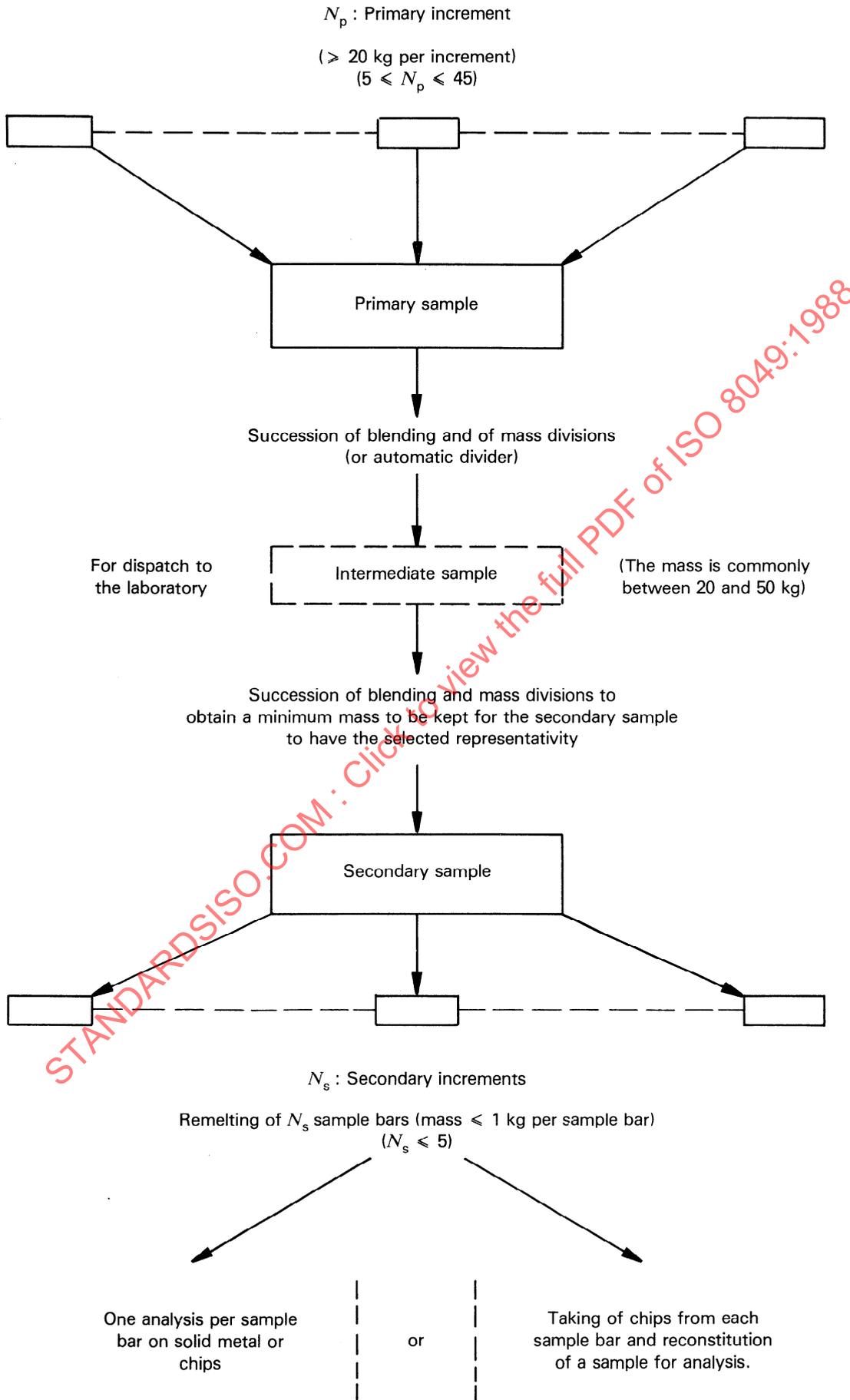


Figure A.1 — General scheme of increments

### A.3 Estimate of primary variance

Observations have been made by some producers either at dispatch from the manufacturing plant or at the time of delivery at the purchaser's.

Even in the case  $n = 5$  (maximum range of nickel content), very low values of  $V_p$  have always been observed.

The value  $V_p = 0,01$  has been retained for  $n = 5$ , which is noticeably in excess of any collected concrete observations. It can therefore be considered as a safe value.

Moreover, it has been assumed that the upper limit adopted for  $V_p$  could be reduced when  $n$  was reduced.

Again a slight diminution of the value of  $S_p = \sqrt{V_p}$  has been admitted by prudence, according to table A.1.

Table A.1 — Variation of  $S_p$

$n$	$S_p = \sqrt{V_p}$	$V_p$
1	0,06	0,003 6
2	0,07	0,004 9
3	0,08	0,006 4
4	0,09	0,008 1
5	0,10	0,010 0

### A.4 Estimate of secondary variance

The main conclusions drawn from a mathematical model for the estimation of this fundamental variance are :

**A.4.1** The fact that heats with different particle size ranges and nickel content are blended has no effect on the value of the fundamental variance if no correlation is found between the nickel content and the mass of each shot particle.

#### A.4.2 Estimation of the between-heats variance $V_c$

This is the variance which estimates the variations due only to the difference in nickel contents between heats (without taking into account the granular nature of the material).<sup>1)</sup>

This variance is of the form

$$V_c = \frac{(n + 2\varepsilon)^2}{\alpha} \times \frac{M - N}{M - 1} \quad \dots (3)$$

where

$\varepsilon$  is the numerical value expressing the uncertainty in the determination of the nickel content of a heat, which means that the range defined by  $k$  to  $(k + n)$  % as planned to constitute the blended batch may have to be extended in practice from  $(k - \varepsilon)$  to  $(k + n + \varepsilon)$  %. The theoretical range is  $n$ . The practical range is  $n + 2\varepsilon$ ;

$M$  is the total number of shot particles in the lot;

$N$  is the number of shot particles in the primary sample;

$\alpha$  is the coefficient, the numerical values of which have been calculated under three hypothetical conditions of nickel content distribution within the range, as follows :

Hypothesis 1 (most unfavourable case) : Half of the heats are at the lower limit  $(k - \varepsilon)$  % and the other half at the upper limit  $(k + n + \varepsilon)$  % :  $\alpha = 4$

Hypothesis 2 (less unfavourable case) : The contents, heat after heat, are uniformly distributed between the two limits :  $\alpha = 12$

Hypothesis 3 (more favourable case) : Normal (Gaussian) distribution of the mean in the middle of the range and the standard deviation at 1/6th of the range :  $\alpha = 36$

In practice, a producer tries to centre his production of shot on a desirable content for the heats intended for constituting the blended lot; the hypothesis of a normal distribution is then logical. For shot, hypothesis 2 remains consequently an unfavourable hypothesis.

#### A.4.3 Estimation of the required secondary sample mass for the desired uncertainty to be achieved (fundamental error)

The number  $N$  of shot particles to be taken for achieving an uncertainty characterized by standard deviation  $s_s$  is given by the formula :

$$N = \frac{(1 + \varrho_m^2) V_c}{s_s^2} + \varrho_m^2 \quad \dots (4)$$

in which parameter  $\varrho_m$  represents the coefficient of variation of shot mass  $m$ . This parameter, which would disappear if all shot particles had the same mass ( $\varrho_m = 0$ ), is used to account for the fact that the sampling variance depends on the product particle size heterogeneity. The presence of larger particles beside smaller particles has an unfavourable influence on the fundamental variance for a given total number  $N$  of particles in the sample.

The log-normal representation of the particle size distribution curve (i.e. of the cumulative proportion of undersize — or oversize — on a normal scale versus particle size on a logarithmic scale) is used to evaluate the quantity  $(1 + \varrho_m^2)$  as well as the quantities  $\tilde{m}$  and  $\sigma_{\ln m}^2$  which appear in the following equation for mean particle mass.

$$E(m) = \tilde{m} \exp \left( - \frac{\sigma_{\ln m}^2}{2} \right) \quad \dots (5)$$

1) This is the same case as for the sampling of ferronickels in ingots or pieces as dealt with in ISO 8050.

2) In  $m$  is the natural logarithm of  $m$ .

The mass  $\tilde{m}$  of particle, such that particles of mass less than  $\tilde{m}$  represent as a whole 50 % of the product mass, and consequently as a correlation, that particles of mass greater than  $\tilde{m}$  also represent 50 % of the total mass, can be derived from the cutting size, which defines the same proportions. Similarly,  $\sigma_{\ln m}$ , the standard deviation of the natural logarithm of shot masses, can be derived from the slope of the particle size distribution curve.

Finally,  $(1 + \varrho_m^2)$  is derived from  $\sigma_{\ln m}$  using the equation

$$1 + \varrho_m^2 = \exp(\sigma_{\ln m}^2)$$

From equations (4) and (5), the mass  $M_e$  of the required shot sample for secondary sample representativity is derived.

$$M_e = N \cdot E(m) \quad \dots (6)$$

$$M_e = \tilde{m} \left( \frac{(1 + \varrho_m^2) V_c}{s_S^2} + \varrho_m^2 \right) \exp\left(-\frac{\sigma_{\ln m}^2}{2}\right) \quad \dots (6a)$$

Conversely,  $s_S^2$  can be expressed in terms of  $M_e$  from equation (6a). If  $s_S^2$  is small in relation to  $V_c$ , which is generally the case, the second term of the second member of equation (4)<sup>1)</sup> can be neglected and a relationship can be obtained which is of the form

$$s_S^2 = \frac{\text{Constant}}{M_e} \quad \dots (7)$$

since parameters  $\tilde{m}$ ,  $\sigma_{\ln m}$ ,  $\varrho_m$  and  $V_c$  are characteristic constants of the sampled product.

Whatever the approximation, the number of required small ingots can be derived in any case from equation (6a) for  $M_e$  and from the mass  $M_1$  of a small ingot obtained by melting.

$$N_s = \frac{M_e}{M_1} \quad \dots (8)$$

The ratio  $M_e/M_1$  is obviously rounded up in the upper unit.

### A.5 Examples of numerical applications

Table A.2 gives the values of the variations of the variance between heats  $V_c$ .

Table A.3 gives the values of  $N$  and  $M_e$  in a large range of hypotheses.

The numbers of increments  $N_p$  and  $N_s$  to be found in table A.1 are deduced from the following numerical values :

$$V_c = \frac{(n + 2\varepsilon)^2}{24}$$

that is

$$\alpha = 24 \text{ 2)}$$

$$1 + \varrho_m^2 = 4,5 \text{ 3)}$$

$$\varepsilon = 0,10$$

Table A.2 — Variation of the variance between heats  $V_c$

		Hypothesis for the distribution of heat contents in the range of the lot Nickel contents			
		Equal distribution at both limits	Uniform distribution over the whole range	Intermediate distribution	Normal (Gaussian) distribution with 6 standard deviations from $(k - \varepsilon)$ to $(k + n + \varepsilon)$
$\varepsilon = 0,10$	$V_c =$	$\frac{(n + 2\varepsilon)^2}{4}$	$\frac{(n + 2\varepsilon)^2}{12}$	$\frac{(n + 2\varepsilon)^2}{24}$	$\frac{(n + 2\varepsilon)^2}{36}$
	$n = 1$	0,360	0,120	0,060	0,040
	$n = 2$	1,210	0,403	0,202	0,134
	$n = 3$	2,560	0,853	0,426	0,284
	$n = 4$	4,410	1,470	0,735	0,490
	$n = 5$	6,760	2,253	1,127	0,751

1) In other words, in this case  $\varrho_m^2$  is negligible with respect to  $\frac{(1 + \varrho_m^2) V_c}{s_S^2}$

2)  $\alpha = 12$  (see A.4.2) corresponds to an unfavourable hypothesis in the selection of these heats constituting a lot, and  $\alpha = 36$  corresponds to a normal hypothesis. The value  $\alpha = 24$  is an intermediate value adopted for safety reasons.

3) These numerical values correspond to a rather large particle size, which should not be exceeded in normal production ( $d_{50} \approx 12$  mm and  $d_{95} \approx 25$  mm).

Table A.3 — Variation of values  $N$  and  $M_e$

Magnitude of range	Particle size distribution hypothesis	Ni content distribution within the range							
		$V_c = \frac{(n + 2\varepsilon)^2}{4}$		$V_c = \frac{(n + 2\varepsilon)^2}{12}$		$V_c = \frac{(n + 2\varepsilon)^2}{24}$		$V_c = \frac{(n + 2\varepsilon)^2}{36}$	
		$N$	$M_e$ (kg)	$N$	$M_e$ (kg)	$N$	$M_e$ (kg)	$N$	$M_e$ (kg)
$n = 1$	G-1	475	0,636	160	0,214	81	0,109	55	0,073
	G-2	740	1,68	249	0,566	127	0,288	86	0,194
	G-3	614	3,34	207	1,12	105	0,571	71	0,387
$n = 2$	G-1	1 590	2,13	531	0,712	267	0,358	178	0,239
	G-2	2 477	5,62	828	1,90	417	0,947	278	0,631
	G-3	2 055	11,18	687	3,74	346	1,88	231	1,25
$n = 3$	G-1	3 361	4,50	1 119	1,50	559	0,749	375	0,502
	G-2	5 237	11,89	1 748	3,97	875	1,99	585	1,33
	G-3	4 345	23,64	1 450	7,89	726	3,95	485	2,64
$n = 4$	G-1	5 788	7,76	1 931	2,59	967	1,30	645	0,864
	G-2	9 018	20,47	3 009	6,83	1 506	3,42	1 006	2,28
	G-3	7 483	40,71	2 496	13,58	1 249	6,80	834	4,54
$n = 5$	G-1	8 871	11,89	2 958	3,96	1 481	1,98	988	1,32
	G-2	13 822	31,37	4 609	10,46	2 308	5,24	1 539	3,49
	G-3	11 468	62,39	3 824	20,80	1 915	10,42	1 277	6,95

$N$  = number of shot particles to be taken

$M_e$  = mass of secondary sample in order to achieve  $s_s = 0,05$  and  $V_s = 0,002 5$

$\varepsilon = 0,10$

The calculation is carried out for three hypotheses of particle size distribution.

G-1 : favourable hypothesis

G-2 : normal hypothesis

G-3 : very unfavourable hypothesis

Mass per 1 kg small ingot (or 1 000 g for the numerical application) :

$$E(m) = 2,0 \text{ g } ^1$$

That is

$$V_s = s_s^2 = 0,375 \times 10^{-3} (n + 0,2)^2$$

which gives the values indicated in table A.4.

Table A.4 — Variation of  $V_s$

$n$	$V_s$	$s_s$
1	0,000 54	0,023 2
2	0,001 82	0,042 6
3	0,003 84	0,062 0
4	0,006 62	0,081 3
5	0,010 14	0,100 7

Selection of the numbers of primary and secondary increments ( $N_p$  and  $N_s$ ).

In clause A.2, equation (1) showed that the variance in the nickel analysis due to sampling is  $V_e$  where

$$V_e = \frac{V_p}{N_p} + \frac{V_s}{N_s} \dots (1)$$

If one selects an acceptable value for this sampling variance,  $V_e$ , then the known values of the primary variance,  $V_p$ , and of the secondary variance,  $V_s$ , will establish acceptable values for the number of primary increments,  $N_p$ , and the number of secondary increments,  $N_s$ .

For trade purposes one would like to set the sampling variance,  $V_e$ , as low as possible. However, there is no advantage in setting the sampling variance smaller than the variance in the analytical methods. The variances in the analytical methods (between analysts, within laboratory and between laboratories) usually range between 0,002 5 and 0,010. So it is reasonable to set a target value for the sampling variance of about 0,002 5 or less.

$$V_e = \frac{V_p}{N_p} + \frac{V_s}{N_s} \leq 0,002 5 \dots (1a)$$

1) These numerical values correspond to a rather large particle size, which should not be exceeded in normal production ( $d_{50} \approx 12 \text{ mm}$  and  $d_{95} \approx 25 \text{ mm}$ ).

Values for  $V_p$  can be taken from table A.1 and values for  $V_s$  can be taken from table A.4. In both tables, the variances are shown to depend upon the range of nickel content,  $n$ . These tables can be combined:

$n$	$V_p$ (from table A.1)	$V_s$ (from table A.4)
1	0,003 6	0,000 54
2	0,004 9	0,001 82
3	0,006 4	0,003 84
4	0,008 1	0,006 62
5	0,010 0	0,010 14

Clearly, equation (1a) allows a wide range in choosing  $N_p$  and  $N_s$  when  $V_p$  and  $V_s$  are fixed. However, the following recommendations are made in table A.5.

The values in table A.5 are as shown in table 1, taking the first line for  $N_p$  and the only line for  $N_s$ .

For  $N_p$  slightly greater values have been retained in this International Standard when the tonnage of the lot is increased, although theoretically the mass of the lot has no influence on the representativity of sampling.

It was, however, considered that such increased values of  $N_p$ , depending on the lot tonnage, could be recommended because

- when the duration of discharge increased, the duration of the sample handling made it easier;
- this involved additional safeguards for the purchaser if the latter fears segregations due to transport (this phenomenon has never been observed in the investigations made while developing this International Standard).

Excellent representativity can thus be obtained with a small number of primary increments even with lots of a high tonnage (up to 2 500 t for example).

**Table A.5 — Recommended primary and secondary increments  
(to achieve  $V_e$  less than 0,002 5)**

Nickel range $n$	Primary variance $V_p$	Secondary variance $V_s$	Recommended sampling increments		Calculated sampling variance $V_e$
			primary $N_p$	secondary $N_s$	
1	0,003 6	0,000 54	5	1	0,001 26
2	0,004 9	0,001 82	10	2	0,001 40
3	0,006 4	0,003 84	15	3	0,001 71
4	0,008 1	0,006 62	20	4	0,002 06
5	0,010 0	0,010 14	30	5	0,002 36

## Annex B (informative)

### Methods for taking a sample of size $N$ in a supply of $M$ items

#### B.1 General

It should be noted from the outset that in any method for drawing a sample from a population two stages can be distinguished :

- a) the definition of the items to be sampled;
- b) the process of sampling itself.

It should also be recalled that in order to be representative a sample has to be drawn in such a way that any item of the sampled population has the same probability of being drawn.

#### B.2 Methods for defining the items constituting the sample

Two methods can be contemplated : one is random sampling of all the items of the sample; the other is systematic periodic sampling, only the first item being designated at random.

##### B.2.1 Random sampling of items

In this method, all possible samples of  $N$  items (or combinations of  $N$  objects taken among  $M$ ) really have an equal probability.

Let us assume that the  $M$  items of the consignment bear some kind of identification which can be translated by a special numbering from 1 to  $M$ . The problem is then reduced to drawing  $N$  distinct integers at random among the first  $M$  integers.

To this purpose  $N$  random numbers shall first be selected from the uniform distribution in the interval 0 to 1. Some tables give such numbers directly. Others (such as table B.1) only give rows of numbers from 0 to 9, in random order, and real uniformly distributed numbers can easily be selected by taking an integer part equal to zero completed by a sequence of  $n$  decimals represented by  $n$  figures of the table.

*Example :*

Table B.1 is an extract from a table of random numbers which allows all the concrete cases to be found in this International Standard to be dealt with.

If numbers of the uniform distribution from 0 to 1 are needed with, say, five decimal places, groups of five digits shall be taken either by column or by row or by any other systematic

means. Thus, by taking the first five digits of each line for example, the following sequences of digits are obtained :

10275  
28415  
34214  
61817  
etc.

and the numbers are : 0,102 75 — 0,284 15 — 0,342 14 — 0,618 17, etc.

NOTE — In table B.1, the spaces between rows and columns are only for improved readability of the table, which regroups digits from 0 to 9 in a random order.

Let  $x_1, x_2, \dots, x_N$  be a series of  $N$  numbers of the uniform distribution thus obtained. All these (real) numbers are multiplied by the integer  $M$ , which gives real numbers selected at random in the interval 0 to  $M$ .

$$Mx_1, Mx_2, \dots, Mx_N$$

These real numbers are rounded up to the next highest integer :

$$E_1 = [Mx_1] + 1$$

$$E_2 = [Mx_2] + 1$$

...

$$E_N = [Mx_N] + 1$$

where  $[Mx_i]$  = integer part of  $Mx_i$

The integers  $E_1, E_2, \dots, E_N$  then identify the items to be drawn from the population of  $M$  objects.

If this procedure results in drawing some equal numbers  $E_i$ , additional  $x_i$  numbers shall be drawn until  $N$  different values of  $E_i$  have been obtained.

##### B.2.2 Systematic periodic sampling of items

In this method, not all samples of  $N$  items constituted from  $M$  items of the supply have an equal probability of being obtained. Actually, this probability is zero for a very large number of them, although any specific item can have (at least approximately) the same probability of being part of the sample. This somewhat paradoxical result is explained by the non-independence of individual increments.

A whole quotient of  $M$  by  $N$ , i.e.  $Q$ , is calculated and, if the division gives a remainder  $R$  (less than  $N$ ), it is neglected.

An integer is then chosen at random in the sequence 1, 2, ...,  $Q-1$ ,  $Q$ , for example by the method described in B.2.1. Let this number be  $H$ . The items composing the sample are then defined by integers :

$$H, Q + H, 2Q + H, \dots, (N - 1)Q + H$$

It can be seen that by this method  $M - NQ$  items are ignored by sampling but that only one drawing is necessary from the table of random numbers.

Because of the unequal probability of all possible samples of items being drawn, it is also necessary to specify that the theoretical formulae for calculating the sampling variance do

not apply in this case, except if the lot of items has been mixed carefully, which is hardly practicable.

### B.3 Drawing of $N$ identified items

$N$  items have been theoretically identified by integers  $E_1, E_2, \dots, E_N$  among the  $M$  items constituting the lot. The physical operation of sampling remains to be carried out without losing sight of the fact that the items generally do not bear an identification mark. As these items are trucks, wagons, containers, drums, etc., they can be numbered<sup>1)</sup> from 1 to  $M$ , and among them the  $N$  items designated by integers  $E_1, \dots, E_N$  shall be sampled according to the procedure described in 5.1.2 or 5.1.3, then clause 6 and the following.

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1) The term "item" (a general concept of statistics) has the same meaning as the term "unit", which is used in the main text to designate the various forms of conditioning.

Table B.1 – Table of random numbers

10 27 53 96 23	71 50 54 36 23	54 31 04 82 98	04 14 12 15 09	26 78 25 47 47
28 41 50 61 88	64 85 27 20 18	83 36 36 05 56	39 71 65 09 62	94 76 62 11 89
34 21 42 57 02	59 19 18 97 48	80 30 03 30 98	05 24 67 70 07	84 97 50 87 46
61 81 77 23 23	82 82 11 54 08	53 28 70 58 96	44 07 39 55 43	42 34 43 39 28
61 15 18 13 54	16 86 20 26 88	90 74 80 55 09	14 53 90 51 17	52 01 63 01 59
91 76 21 64 64	44 91 13 32 97	75 31 62 66 54	84 80 32 75 77	56 08 25 70 29
00 97 79 08 06	37 30 28 59 85	53 56 68 53 40	01 74 39 59 73	30 19 99 85 48
36 46 18 34 94	75 20 80 27 77	78 91 69 16 00	08 43 18 73 68	67 69 61 34 25
88 98 99 60 50	65 95 79 42 94	93 62 40 89 96	43 56 47 71 66	46 76 29 67 02
04 37 59 87 21	05 02 03 24 17	47 97 81 56 51	92 34 86 01 82	55 51 33 12 91
63 62 06 34 41	94 21 78 55 09	72 76 45 16 94	29 95 81 83 83	79 88 01 97 30
78 47 23 53 90	34 41 92 45 71	09 23 70 70 07	12 38 92 79 43	14 85 11 47 23
87 68 62 15 43	53 14 36 59 25	54 47 33 70 15	59 24 48 40 35	50 03 42 99 36
47 60 92 10 77	88 59 53 11 52	66 25 69 07 04	48 68 64 71 06	61 65 70 22 12
56 88 87 59 41	65 28 04 67 53	95 79 88 37 31	50 41 06 94 76	81 83 17 16 33
02 57 45 86 67	73 43 07 34 48	44 26 87 93 29	77 09 61 67 84	06 69 44 77 75
31 54 14 13 17	48 62 11 90 60	68 12 93 64 28	46 24 79 16 76	14 60 25 51 01
28 50 16 43 36	28 97 85 58 99	67 22 52 76 23	24 70 36 54 54	59 28 61 71 96
63 29 62 66 50	02 63 45 52 38	67 63 47 54 75	83 24 78 43 20	92 63 13 47 48
45 65 58 26 51	76 96 59 38 72	86 57 45 71 46	44 67 76 14 55	44 88 01 62 12
39 65 36 63 70	77 45 85 50 51	74 13 39 35 22	30 53 36 02 95	49 34 88 73 61
73 71 98 16 04	29 18 94 51 23	76 51 94 84 86	79 93 96 38 63	08 58 25 58 94
72 20 56 20 11	72 65 71 08 86	79 57 95 13 91	97 48 72 66 48	09 71 17 24 89
75 17 26 99 76	89 37 20 70 01	77 31 61 95 46	26 97 05 73 51	53 33 18 72 87
37 48 60 82 29	81 30 15 39 14	48 38 75 93 29	06 87 37 78 48	45 56 00 84 47
68 08 02 80 72	83 71 46 30 49	89 17 95 88 29	02 39 56 03 46	97 74 06 56 17
14 23 98 61 67	70 52 85 01 50	01 84 02 78 43	10 62 98 19 41	18 83 99 47 99
49 08 96 21 44	25 27 99 41 28	07 41 08 34 66	19 42 74 39 91	41 96 53 78 72
78 37 06 08 43	63 61 62 42 29	39 68 95 10 96	09 24 23 00 62	56 12 80 73 16
37 21 34 17 68	68 96 83 23 56	32 84 60 15 31	44 73 67 34 77	91 15 79 74 58
14 29 09 34 04	87 83 07 55 07	76 58 30 83 64	87 29 25 58 84	86 50 60 00 25
58 43 28 06 36	49 52 83 51 14	47 56 91 29 34	05 87 31 06 95	12 45 57 09 09
10 43 67 29 70	80 62 80 03 42	10 80 21 38 84	90 56 35 03 09	43 12 74 49 14
44 38 88 39 54	86 97 37 44 22	00 95 01 31 76	17 16 29 56 63	38 78 94 49 81
90 69 59 19 51	85 39 52 85 13	07 28 37 07 61	11 16 36 27 03	78 86 72 04 95
41 47 10 25 62	97 05 31 03 61	20 26 36 31 62	68 69 86 95 44	84 95 48 46 45
91 94 14 63 19	75 89 11 47 11	31 56 34 19 09	79 57 92 36 59	14 93 87 81 40
80 06 54 18 66	09 18 94 06 19	98 40 07 17 81	22 45 44 84 11	24 62 20 42 31
67 72 77 63 48	84 08 31 55 58	24 33 45 77 58	80 45 67 93 82	75 70 16 08 24
59 40 24 13 27	79 26 88 86 30	01 31 60 10 39	53 58 47 70 93	85 81 56 39 38
05 90 35 89 95	01 61 16 96 94	50 78 13 69 36	37 68 53 37 31	71 26 35 03 71
44 43 80 69 98	46 68 05 14 82	90 78 50 05 62	77 79 13 57 44	59 60 10 39 66
61 81 31 96 98	00 57 25 60 59	46 72 60 18 77	55 66 12 62 11	08 99 55 64 57
42 88 07 10 05	24 98 65 63 21	47 21 61 88 32	27 80 30 21 60	10 92 35 36 12
77 94 30 05 39	28 10 99 00 27	12 73 73 99 12	49 99 57 94 82	96 88 57 17 91
78 83 19 76 16	94 11 68 84 26	23 54 20 86 85	23 86 66 99 07	36 37 34 92 09
87 76 59 61 81	43 63 64 61 61	65 76 36 95 90	18 48 27 45 68	27 23 65 30 72
91 43 05 96 47	55 78 99 95 24	37 55 85 78 78	01 48 41 19 10	35 19 54 07 73
84 97 77 72 73	09 62 06 65 72	87 12 49 03 60	41 15 20 76 27	50 47 02 29 16
87 41 60 76 83	44 88 96 07 80	83 05 83 38 96	73 70 66 81 90	30 56 10 48 59

## Annex C (informative)

### Technical conditions for drilling and milling

#### C.1 General

The hardness of ferronickel small ingots may vary considerably, depending on the grades (nickel content) and especially on the quantity of impurities present (mainly C and Si).

The small ingot must be considered very hard when it falls between 180 and 600 in Vickers hardness (or equivalent hardness scales).

The cutting tools used and the conditions under which they are used must be carefully selected. Cutting is extremely difficult, because the work must always be done dry to avoid any contamination.

NOTE — Machining by drilling or milling is the most common procedure. Shaping with planing tools meeting the general requirements below is also possible. Work is then done on cut-off faces as in milling. This type of machining is, however, slower.

#### C.2 Case of very hard ferronickel

When machining is very difficult and involves tool wear and subsequent significant chip contamination or even impossible sampling, it is advisable to heat treat (temper) the material. The practical procedure is dependent on the metal hardness and crystalline structure. For guidance, tempering is generally useful when hardness exceeds 180 Vickers. Tempering may be carried out either on small ingots or on pieces cut off small ingots, as follows :

The small ingots or pieces are placed in a furnace at 650 to 800 °C for 2 to 4 h. The furnace is then turned off and allowed to cool slowly overnight.

If a shorter period is required, the small ingots or pieces may be cooled in a few hours by immersing them in sand down to a temperature less than 200 °C.

This treatment induces oxidation of the surface when conducted in the presence of air, and decarburization to a depth of 0,5 to 1 mm, whatever the atmosphere in the furnace. Chips shall therefore not be taken from the surface of the metal after heat treatment. Either a 2 to 3 mm thickness should be cut from the surfaces and the resultant block machined, or the chips obtained from the first 2 to 3 mm of the sample during machining should be discarded.

#### C.3 Selection of cutting tools

The cutting tools used shall be made of suitable types and grades of steel, so that tools wear and resultant contamination of the chips will be reduced as much as possible.

For high-speed steel tools, table C.1 gives the specifications from ISO 4957, *Tool steels*.

The high carbon, chromium and cobalt contents ensure tool hardness; the molybdenum prevents the chips from sticking to the tool.

For hard ferronickels (for example of hardness greater than 180 Vickers), experience has shown that it is indispensable to have a steel with a cobalt content greater than or equal to 7,5 %; type S 11 appears to be the most suitable.

For less hard ferronickels, one may be satisfied with a quality containing about 5 % cobalt, for example type S 12.

Table C.1 — Steels for high-speed tools

Grade	S 9	S 10	S 11	S 12
Designation	HS 12-1-5-5	HS 10-4-3-10	HS 2-9-1-8	HS 7-4-2-5
C %	1,45 to 1,60	1,20 to 1,35	1,05 to 1,20	1,05 to 1,20
Co %	4,70 to 5,20	9,50 to 10,5	7,50 to 8,50	4,70 to 5,20
Cr %	3,50 to 4,50	3,50 to 4,50	3,50 to 4,50	3,50 to 4,50
Mo %	0,70 to 1,00	3,20 to 3,90	9,00 to 10,0	3,50 to 4,20
V %	4,75 to 5,55	3,00 to 3,50	0,90 to 1,40	1,70 to 2,20
W %	11,5 to 13,0	9,00 to 10,0	1,30 to 1,90	6,40 to 7,40
Minimum Rockwell hardness after tempering (HRC) <sup>1)</sup>	65	66	66	66

1) 66 HRC is equivalent to approximately 900 Vickers.

For tungsten carbide tools : a type shall be chosen that affords a compromise between resistance to wear and toughness in order to avoid tool chafing or rupture. Selection has therefore to be made between types M 10, M 20 or M 30 of ISO 513.

NOTE — The data in this clause are for guidance only. They correspond to present experience of laboratories having experimented in the field.

### C.4 Additional precautions

Vibration between the cutting tool and the metal being machined shall be avoided as far as possible.

When the metal is to be drilled, very short drills that are not too thin shall be used (the diameter should be no less than 12 mm and preferably between 15 and 20 mm). A low helix angle is also favourable; for example, a drill with a 15° helix angle could be used, as opposed to a standard drill with a helix angle of 30°.

The use of a taper-shank drill (Morse taper socket No. 2 or 3) is highly desirable.

Milling cutters shall also be short in relation to their diameter.

Finally, the machine shall be very rigid. This is readily achieved with a milling machine but more difficult to accomplish with a drilling machine, whether equipped with a milling table or not.

In either case, it shall be possible to mount the tool on the spindle using a sufficiently solid intermediate : standard ISO taper SA 40 or SA 50.

### C.5 Machining parameters

The machining specifications shall be such that

- there is very little heating of the tool, so that it does not become worn; tool wear can be detected by examining the

chips produced during machining : slight yellowing is acceptable, but they must never be blue;

- the feed per tooth does not fall below a minimum value during either drilling or milling, so that the material does not undergo work hardening; this is necessary for normal operation of the tool, free from vibration, wear and abnormal heating.

The parameters given in table C.2 must be considered to achieve a good compromise.

The relationships of these parameters are expressed in the following equations :

$$V_1 = \frac{\pi DN}{1\ 000}$$

$$a = \frac{V_2}{N_d}$$

(valid with the above-mentioned units)

Good machining conditions are obtained by selecting appropriate values for  $V_1$  and  $a$  and then adjusting  $N$  and  $V_2$  accordingly on each machine<sup>1)</sup>.

Examples of recommended specifications are given in table C.3.

Understandably, for low-hardness metal machining, the maximum values given in table C.3 may be increased.

To achieve these specifications in practice, machining shall be done within the following limits :

$$N = 30 \text{ to } 100 \text{ rpm when drilling}$$

$$40 \text{ to } 100 \text{ rpm when milling}$$

$$V_2 = 3 \text{ to } 10 \text{ mm/min when drilling}$$

$$5 \text{ to } 20 \text{ mm/min when milling}$$

Table C.2 — Parameters to be considered

Symbol	Parameter	Unit of measurement
$N$	Speed of rotation of the tool	revolution per minute
$D$	Diameter of drill or milling cutter	millimetre
$d$	Number of teeth <sup>1)</sup>	
$V_1$	Linear cutting speed	metre per minute
$V_2$	Rate of longitudinal feed or cross-feed (milling) or of vertical feed (drilling)	millimetre per minute
$a$	Feed per tooth	millimetre per tooth

1) In mechanical terms, each tooth corresponds, in the case of a drill (examples in C.6.1, C.6.2 and C.6.3) or in the case of end mill cutters (examples in C.6.4, C.6.5 and C.6.6) to one flute.

1) In the case of milling, the linear cutting speed is applied in a rigorous manner; the above-mentioned diameter  $D$  is always the diameter corresponding to the cutting point of the rotating tool. In the case of a drill, the formula  $V_1$  normally applies to the cutting portion of the drill which is located on the outer diameter; but  $V_1$  has bad significance for the working zones of the drill on a smaller diameter and above all for the point, which penetrates by pressure rather than by cutting. This is why the set of cutting parameters can be controlled better in milling than in drilling.

Table C.3 — Examples of recommended specifications

Tool	$V_1$	$V_1$	$a$	$a$	$a$
	Maximum m/min	Desirable m/min	Maximum mm/tooth	Desirable mm/tooth	Minimum mm/tooth
High-speed steel drill	4	2 to 3	0,05	0,04	0,03
Tungsten carbide drill	10 to 12	4 to 7	0,03	0,02	0,015
High-speed steel chip-breaking milling cutter	6	2 to 3	0,03	0,015 to 0,02	0,01
High-speed steel end milling cutter	6	2 to 4	0,05	0,03 to 0,04	0,02

**C.6 Examples of suitable tools**

The descriptions below are given for guidance only. They should facilitate research for the reader and also testing for selecting the most appropriate cutting tools from the various producers in each country.

Only a life duration test with macrographical cutting lip examination can be used for drawing reliable conclusions. A high-performing tool type or sharpening obtained at one supplier may perform less well at another. The following are examples only.

**C.6.1 High-speed steel drill**

- Diameter : 15 to 20 mm
- Morse taper : No. 2 or 3
- Useful length : 60 to 70 mm
- Helix angle : 15° (or, failing this, 30°)
- Point angle : 140° (or, failing this, 130°, but no less)
- Clearance angle : 5° to 7°
- Back-clearance angle : About 15°
- Rake angle : Positive (i.e. in the same direction as the helix), but may be 3° to 6°
- Drill sharpening : 3 rake angles : clearance, back-clearance and web clearance

Web thinning permits having a cross-edge of 1 to 2 mm.

**C.6.2 Carbide drill**

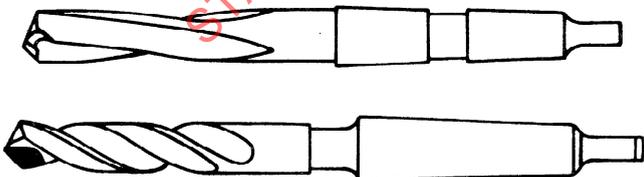


Figure C.1 — Carbide drill

- Diameter : Approximately 15 mm
- Useful length : Approximately 35 mm
- Morse taper : No. 2
- Helix angle : 10° to 15° (or, failing these, 30°)
- Point angle : 130°
- Clearance angle : 2° to 4°
- Back-clearance angle : About 15°

Rake angle : Positive (i.e. in the same direction as the helix) but may be 2° to 5°

Sharpening : 3 rake angles : clearance, back-clearance and web thinning.

The remark for web thinning as in C.6.1 applies here. Here it is very important not to reduce the cross-edge to less than 1 mm. Otherwise, the risk of rupture of the point is very high.

This type of drill is not to be used with the hardest ferronickels. The risk of carbide flaking or point rupture increases when hardness increases. And the steel land wear beyond the carbide tip becomes very important due to blocked chips inside the hole inducing friction. (The latter disadvantage is suppressed when using solid carbide tools.)

**C.6.3 Oil-hole drill**

(See table C.4 and figure C.2)

These drills are used with compressed air instead of oil feeding. A special coupling ring is used for this purpose.

Such drills are not made of steels with high cobalt and molybdenum contents. Hence, they cannot be used for very hard ferronickels.

Chip recovery as depicted in figure C.6 shall be used with chips drilling technique.

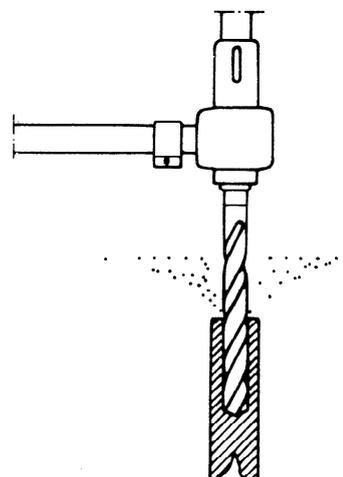


Figure C.2 — Oil-hole drill

Table C.4 — Characteristics of oil-hole drills

	Drill diameter	
	15,875 mm	19,050 mm
Total length	241,3 mm	266,7 mm
Groove length	123,825 mm	149,225 mm
Helix angle	34°	34°
Point angle	118°	118°
Clearance angle <sup>1)</sup>	10°	10°
Back-clearance angle <sup>2)</sup> (approximately)	10°	10°

1) As measured at the intersection of the side flank with the land (margin).  
2) This angle shall be altered to improve performance. Such sharpening or so-called "flattened-lip" reduces the effective inclination of the axis with respect to the side flank and has proved advantageous for many applications.

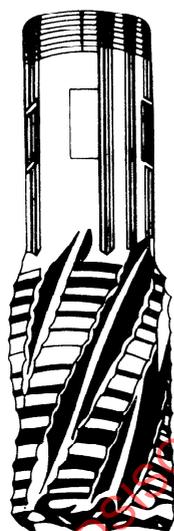
C.6.4 Chip-breaking mill cutter<sup>1)</sup>

Figure C.3 — Chip-breaking mill cutter

Diameter : 20 to 30 mm  
Number of teeth (or flutes) : 4, 5 or 6  
Profile of the longitudinal edge of the length of each flute : Rough-formed cutting edge, round profile  
Useful length : 30 to 45 mm  
Clearance angle : 3° to 4°  
Rake angle : 2° to 5°  
Mounting : By an intermediate socket with SA 40 or SA 50 taper

This type of milling cutter is suitable for working with the cylindrical part.

Cut depth : 0,5 to 2 mm, depending on the hardness of the material

1) For the terminology of milling cutters, see ISO 3855.

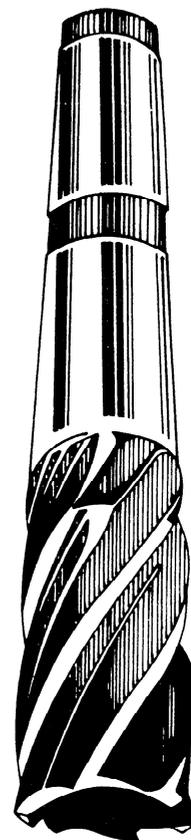
C.6.5 End-mill cutter<sup>1)</sup>

Figure C.4 — End-mill cutter

Diameter : 20 to 50 mm  
Morse taper : No. 3 or 4  
Number of teeth (or flutes) : 4, 5 or 6  
Useful length : 35 to 75 mm  
Clearance angle : 4° to 6°  
Rake angle : 2° to 5°