

Basic considerations for pneumatic conveying against high back pressures

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Pneumatic conveying systems for injecting solids against very high back pressures (sometimes up to $\cong 50$ bar) are used particularly in special gasification systems, but also in other processes. The following text provides a general insight into the aspects to be considered in the system concept and design for such requirements and provides appropriate calculation approaches.

1 Introduction

Various plant manufacturers are currently planning process plants for special bulk materials with operating pressures of $p_G = (40 - 50)$ bar, which are to be fed with finely ground solids by pneumatic conveying systems. For energetic and economic reasons, higher loads are required of these conveyor systems, $\mu = \dot{M}_S / \dot{M}_F > 20$, with: \dot{M}_S, \dot{M}_F = mass flows of solids (S) and conveying gas (F). It is well known that larger loads result in a reduced energy requirement for the solids transport. Compared to the loads of pneumatic conveying systems working with almost atmospheric pressure, i.e. systems conveying against $p_G \cong 1$ bar, this does not appear to present a problem. However, the fact that the gas phase of the 2-phase flow of gas/solids is compressible is often underestimated or not sufficiently taken into account. Increasing conveying pressure or back pressure results in an increasingly stronger compression of the conveying gas and thus, at a given solids throughput \dot{M}_S , in a decrease in the local distances between the bulk solids particles, i.e. in a mixture compression, and thus in an increase in the local solids volume fraction. Ultimately, this also results in a larger required conveying gas mass flow \dot{M}_F to provide the required conveying gas velocity for a given solids mass flow \dot{M}_S and thus in a reduction in the maximum load μ that can be achieved.

Since the load μ along a conveying section L_R is constant, it does not provide any information about the profile of the local spatial density of the gas/solids mixture ρ_b in the pipe and therefore also provides no specific information about the locally occurring flow patterns. As a general rule, for describing the gas /solid interactions, i.e. particle /particle and particle / pipe wall collisions with corresponding effects on the delivery pipe pressure loss Δp_R , other parameters, e.g. the use of the volume concentrations of one of the two phases, are more

suitable. In the case of the high-pressure conveying systems considered below, a number of simplifications can be made, as will be shown in the following.

With high back pressures, the conditions at the start of the delivery line are critical, as this is where the highest pressure p_{in} and therefore the lowest gas velocity v_{in} occur. If at this point the mixture density exceeds a limit value $\rho_{b,crit}$, that corresponds approximately to that of the loose bulk density ρ_{ss} , the solids then form a plug that fills the entire cross-section of the conveyor pipe, totally blocking it due to wedging.

The following paragraphs first describe in detail the dependence of the conveying gas velocity v_F on the pressure ratios p_{in}/p_{out} prevailing in the pipe, followed by that of the load $\mu(p_{in})$ and the limit load μ_{crit} specified by $\rho_{b,crit}$ and discuss their consequences. Next, some suitable and proven economic design variants for pneumatic solids injection at high back pressures are presented.

Further considerations are based on the example of the pneumatic conveyor system shown in Figure 1, which transports from the "lock hopper system" to the "feed bin" = feed into the process plant.

The following stipulations apply: True velocities are indicated by " u_x " and so-called empty pipe velocities by " v_x ". For the gas phase in pneumatic conveying systems, the following applies, for example: $v_F = \varepsilon_F u_F$, with ε_F = relative gap volume of the gas. As ε_F has values $\varepsilon_F \geq 0.90$ in such systems, it is generally possible to specify $u_F = v_F$ with reasonable accuracy. The relative gap volume ε_F corresponds to the gas volume fraction in the pipe element ΔV_R that is currently under consideration, while the corresponding volume fraction of the solid material is represented by ε_S . The following therefore applies: $\varepsilon_F + \varepsilon_S = 1$.

2 Conveying gas velocities

The conveying gas velocities at the beginning (in) and end (out) of an (unstaggered) conveying section with a constant mixture temperature T_M along the pipe follow with the ideal gas law from

$$\frac{v_{F,out}}{v_{F,in}} = \frac{p_{in}}{p_{out}} \rightarrow p_{out} = p_G, p_{in} = p_G + \Delta p_R \quad (1)$$

$$v_{F,out} = v_{F,in} \cdot \frac{p_{in}}{p_{out}} \quad (2)$$

Example 1:

$$v_{F,in} = 10.0 \text{ m/s}, \Delta p_R = 1.0 \text{ bar}, p_G = 1.0 \text{ bar}$$

$$v_{F,out} = 10.0 \frac{\text{m}}{\text{s}} \cdot \frac{2.0 \text{ bar}}{1.0 \text{ bar}} = 20.0 \frac{\text{m}}{\text{s}}$$

Example 2:

$$v_{F,in} = 10.0 \text{ m/s}, \Delta p_R = 1.0 \text{ bar} \\ (\text{alternatively: } 5.0 \text{ bar}), p_G = 40.0 \text{ bar}$$

$$v_{F,out} = 10.0 \frac{\text{m}}{\text{s}} \cdot \frac{41.0 \text{ bar}}{40.0 \text{ bar}} = 10.25 \frac{\text{m}}{\text{s}}$$

alternatively:

$$v_{F,out} = 10.0 \frac{\text{m}}{\text{s}} \cdot \frac{45.0 \text{ bar}}{40.0 \text{ bar}} = 11.25 \frac{\text{m}}{\text{s}}$$

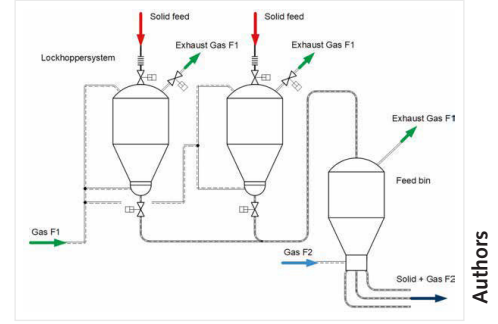
At the high conveying pressures/counterpressures considered here, an approximately constant conveying gas velocity v_F is established along the conveying section. This means that at the selected gas inlet velocity $v_{F,in}$ the flow pattern of the gas/ solids mixture is largely maintained over the entire conveying distance. Critical conveying conditions, e.g. plug formation in the case of fine-grained bulk solids, therefore no longer resolve themselves. In atmospheric pressure operation, the considerable gas expansion would support the plug dispersion process.

3 Pressure dependence $\mu(p)$ of the load

For a given solids mass flow rate \dot{M}_S the continuity equation of the conveying gas for the load is as follows:

$$\mu = \frac{\dot{M}_S}{\dot{M}_F} = \frac{\dot{M}_S}{\frac{\pi}{4} \cdot D_R^2 \cdot \rho_F(p) \cdot v_F(p)} \quad (3)$$

In the following, the operating conditions at the critical starting point of the conveying pipe (in) are considered and the condition of the gas is described using the ideal gas law



1 System setup

$$\rho_F(p_{in}) = \frac{p_{in}}{R_F \cdot T_M} \quad (4)$$

with: ρ_F - gas density,

p_{in} - absolute pressure at the beginning of the conveying line,

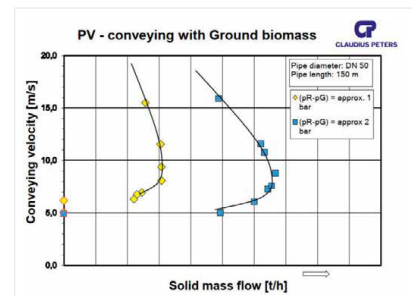
R_F - specific gas constant,

T_M - absolute mixture temperature,

Furthermore, the initial conveying gas velocity $v_{F,in}$ is divided into a minimum velocity $v_{F,in,min}$ and the distance to this minimum velocity $\Delta v_{F,in}$ that is currently selected for the system design. The following applies:

$$v_{F,in} = v_{F,in,min} + \Delta v_{F,in} \quad (5)$$

$V_{F,in,min}$ is the minimum conveying gas velocity below which no solids can be conveyed. It can be derived from conveying diagrams showing the respective bulk material to be conveyed. As an example, Figure 2 shows the pressure dependence of the solids throughput \dot{M}_S relating to ground biomass for systematically varied conveying gas velocities $V_{F,in}$ at two different conveying pressure differences Δp_R at the conveying section $D_R = 54.5 \text{ mm} \times L_R = 155.3 \text{ m}$, back pressure $p_G = 1.0 \text{ bar}$. The ordinate intersections of the \dot{M}_S curves correspond to the minimum velocities $V_{F,in,min}$ at the respective pressure difference Δp_R , i.e. each pressure loss curve Δp_R is assigned a $V_{F,in,min}$ value. If measurements are carried out with the same operating settings on a conveyor section with a larger pipe diameter D_R , this generally leads to larger



2 Conveying diagram $v_{F,in}=f(\dot{M}_S, \Delta p_R)$, $D_R = 54.5 \text{ mm}$, $L_R \approx 155 \text{ m}$

$v_{F,in,min}$ values. This special behaviour is specific to each individual solid. The dependence of the minimum velocity $v_{F,in,min}$ for any given bulk material must be determined experimentally and can be represented with sufficient accuracy for practical use using the empirical power approach

$$v_{F,in,min} = K_v \cdot \frac{D_R^k}{\rho_{F,in}^\ell} \quad (6)$$

In Equation (6), the gas density $\rho_{F,in}$ has a decisive influence. The greater this is, the greater the ability of the conveying gas to carry solids. Using the ideal gas law, the following applies

$$v_{F,in,min} = K_v \cdot \frac{D_R^k}{p_{in}^\ell} \cdot (R_F \cdot T_M)^\ell \quad (7)$$

with: K_v, k, ℓ - solid-specific constants/exponents, to be determined by experiment.

Equation (7) describes the relevant influencing variables of pipe diameter D_R , type of gas R_F , operating temperature T_M , conveying pressure p_{in} , type of bulk material and the condition of the gas, i.e. also the influence of local altitude, relative gas humidity ϕ etc. Details regarding the minimum velocity can be found in [1, 2].

Note: For a selected pressure difference Δp_R , the conveying diagram in Figure 2 provides a simple way of determining the energetically optimum operating point = state of minimum specific energy demand $P_{spez} = (P_v / (\dot{M}_S \cdot L_R))_{min}$ - usually stated in [kWh/(t · 100m)]. This corresponds to the point of contact of the tangent from the coordinate origin with the respective curve $\Delta p_R = konst.$ At this operating point, the associated maximum possible load μ_{max} occurs. It is evident that the corresponding conveying gas velocities $v_{F,in}$, especially in the case of fine-grained bulk solids and high conveying pressures, are too close to the conveying limit $v_{F,in,min}$ to be of use for practical operation designs.

For a specific calculation task (\rightarrow solid and conveying gas given), the following results from equation (7)

$$v_{F,in,min} = K_v^* \cdot \frac{D_R^k}{p_{in}^\ell} \quad (8)$$

p_{in} is to be used as the absolute pressure. By applying a double logarithmic plot of $v_{F,in,min}(p_{in})$ at $D_R = konst.$ or $v_{F,in,min}(D_R)$ at $p_{in} = konst.$ the exponents k and ℓ , as well as the constant K_v^* can be determined from their respective gradients. The exponents determined in this way are in the range $(k, \ell) \cong 0 \dots 1$. It can be seen that the (k, ℓ) values are systematically dependent on the flow behaviour of the bulk solids currently being conveyed [1, 2]. Equation (8) is currently confirmed by practice up to $p_{in} \cong 20$ bar.

The distance $\Delta v_{F,in}$ of the conveying gas velocity from the current applicable minimum velocity $v_{F,in,min}$, equation (5), can be freely selected taking into account the characteristic curve/behaviour of the lock system and the requirements of the downstream plant equipment, e.g. with regard to freedom from pulsation. If pneumatic conveying, i.e. transport of solids takes place at gas velocities $v_{F,in} \geq v_{F,in,min} + \Delta v_{F,in}$ then

$$\Delta v_{F,in} = u_{S,in} \quad (9)$$

can be applied, see Figure 3.

From the measurement results in Figure 2 and Figure 3, the resistance coefficients for calculating the conveying pipe pressure loss Δp_R can also be determined. This is discussed in section 7.

The equations (4, 5, 7) inserted into equation (3) provide the pressure dependence of the load

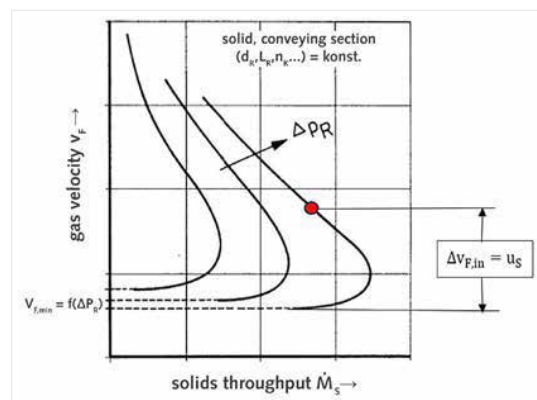
$$\mu = \frac{\dot{M}_S}{\frac{\pi}{4} \cdot D_R^2 \cdot \frac{p_{in}}{R_F \cdot T_M} \cdot \left(K_v \cdot \frac{D_R^k}{p_{in}^\ell} \cdot (R_F \cdot T_M)^\ell + \Delta v_{F,in} \right)} \quad (10)$$

It can be seen from this equation that for given boundary conditions ($\rightarrow \dot{M}_S, D_R, \Delta v_{F,in}$) the possible load μ in each case decreases approximately inversely, proportionally with increasing conveying pressure p_{in} . The counteracting term p_{in}^ℓ that describes the pressure dependence of the minimum velocity has an exponent $\ell \cong 0.5$ for fine-grained bulk solids and is only of minor influence.

4 Critical load μ_{crit}

The pressure dependence of the load μ that is described by equation (10) provides no information as to whether it can actually be attained in practice.

At the inlet to a pneumatic conveying section charged with an increasingly large mass flow \dot{M}_S a limit state (crit) is reached at a certain \dot{M}_S . In this state the solid particles fill the entire



3 Conveying diagram, schematic

pipe cross-section and are in permanent contact with each other, i.e. they form a solid structure that cannot be compressed any further. This condition corresponds approximately to that of a loosely piled heap and can be described, for example, by the relative volume proportion of the solid material in the pipe element under consideration

$$\varepsilon_{S,crit} = (1 - \varepsilon_{F,crit}) = \frac{Q_{b,crit}}{Q_S} \cong \frac{Q_{SS}}{Q_S} \quad (11)$$

with: $\varepsilon_s, \varepsilon_f$ - volume fractions of solids and gas, $(\varepsilon_s + \varepsilon_f) = 1$
 $Q_{b,crit}$ - critical bulk density at the start of the pipe,
 Q_{ss} - loosely heaped bulk density of the bulk material,
 Q_s - particle density of the solid, it is often identical with the specific density of the solid

In order to achieve the desired conveyance with the solids volume fraction at the beginning of the pipe, the condition

$$\varepsilon_{S,in} \leq \varepsilon_{S,crit} \quad \text{alternativ:} \quad \varepsilon_{F,in} \geq \varepsilon_{F,crit} \quad (12)$$

must therefore be fulfilled. The current solids volume fraction at the conveying pipe inlet follows from the continuity equations of the two phases $\dot{M}_S = \varepsilon_{s,in} \cdot A_R \cdot Q_S \cdot u_{s,in}$ and $\dot{M}_F = (1 - \varepsilon_{s,in}) \cdot A_R \cdot Q_{F,in} \cdot u_{F,in} = A_R \cdot Q_{F,in} \cdot v_{F,in}$, leading to

$$\varepsilon_{S,in} = \mu \cdot \frac{Q_{F,in}}{Q_S} \cdot \frac{v_{F,in}}{u_{s,in}} \quad (13)$$

Consideration of the equations (5, 9) and the explanations in section 3 and in [1] lead with $u_{s,in} = \Delta v_{F,in}$ to equation (14)

$$\varepsilon_{S,in} = \mu \cdot \frac{Q_{F,in}}{Q_S} \cdot \left(\frac{v_{F,in,min}}{\Delta v_{F,in}} + 1 \right) \quad (14)$$

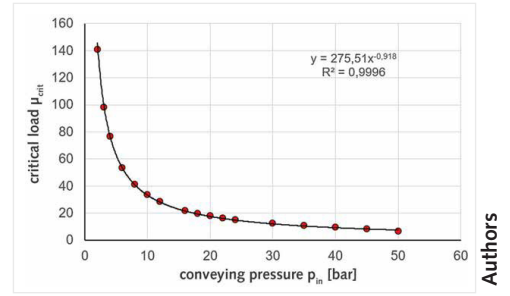
Evaluation of equation (12) by inserting equations (11) and (14) yields the critical = maximum possible load μ_{crit} of the planned conveyance. The following applies:

$$\mu \leq \mu_{crit} = \frac{Q_{SS}}{Q_{F,in}} \cdot \frac{1}{\left(\frac{v_{F,in,min}}{\Delta v_{F,in}} + 1 \right)} \quad (15)$$

Note: In the above considerations, see equation (11), the gas density $Q_{F,X}$ was neglected compared to the densities $Q_{b,x}$ and Q_s . If very light bulk solids and very high conveying pressures occur at the same time, the solids volume fraction $\varepsilon_{s,x}$ should be calculated using the more accurate approach

$$Q_{b,x} = \varepsilon_{s,x} \cdot Q_s + (1 - \varepsilon_{s,x}) \cdot Q_{F,X} \quad \text{as follows} \quad (16)$$

$$\varepsilon_{s,x} = \frac{Q_{b,x} - Q_{F,X}}{Q_s - Q_{F,X}}$$

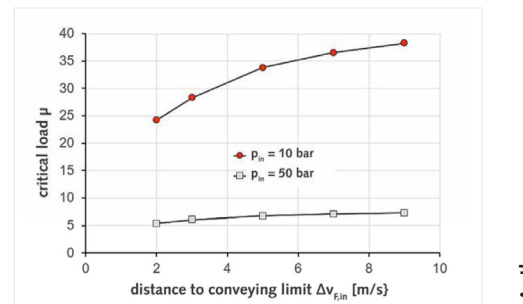


4 Dependence of the critical load μ_{crit} on the conveying pressure p_{in} ; at $\Delta v_{F,in} = 5.0$ m/s

In equation (15), the variables $Q_{F,in}$ and $v_{F,in,min}$ are dependent on the conveying pressure p_{in} , i.e. it is not possible to make a direct μ_{crit} calculation without knowing the pressure loss $|\Delta p_R|$ of the conveying section (L_R, D_R) at a given back pressure p_G . In practice, the desired operational setting is used to first calculate the conveying distance and the resultant load μ , and its achievability is then checked with the critical load μ_{crit} determined using equation (15). As starting values for this iteration, the parameters corresponding to the back pressure $p_G = p_{out}$ can be used. Any necessary corrections then require a change in the operating conditions in the conveying pipe, e.g. by increasing the distance $\Delta v_{F,in}$ to the conveying limit $v_{F,in,min}$ etc. The iteration is stable.

Figure 4 presents an example of a special fine-grained hard coal dust with the characteristics of mean particle diameter $d_{S,50} = 27 \mu m$, solid \cong particle density $Q_s = 1650 \text{ kg/m}^3$, bulk density $Q_{ss} = 550 \text{ kg/m}^3$ as well as the associated constants of $v_{F,in,min}$ equation (8) $k = 0, \ell = 0,35, K_v = 4.0 \text{ bar}^{0,35}$, m/s showing the dependence of the critical = maximum load μ_{crit} for the constant distance $\Delta v_{F,in} = 5.0$ m/s to the current conveying limit $v_{F,in}$.

The conveying gas is air, the operating temperature $T_B = 20 \text{ }^\circ\text{C}$. It can be seen that, for example, the maximum possible load of $\mu_{crit} \cong 34 \text{ kg}_s/\text{kg}_F$ at $p_{in} = 10$ bar falls to $\mu_{crit} \cong 7 \text{ kg}_s/\text{kg}_F$ at $p_{in} = 50$ bar. The exponent of the power function balancing the calculated values in Figure 4 has a value $n = 0.918$ that is close to 1, as was to be expected on the basis of the previous explanations, equation (10).



5 Dependence of the critical load μ on the distance to the conveying limit $\Delta v_{F,in}$ at conveying pressures $p_{in} = 10$ bar and 50 bar

Figure 5 shows, for the same application, the dependence of the load μ_{crit} on the distance $\Delta v_{F,in}$ at conveying pressures $p_{in} = 10$ bar and 50 bar. μ_{crit} increases, as can already be seen from equation (15), with greater $\Delta v_{F,in}$. As this raises the distance to the conveying limit, the gas/solids mixture can be compressed more strongly.

As a general rule, the greater the back pressure $p_G = p_{out}$ and thus the conveying pressure p_{in} , the lower the maximum possible limit load μ_{crit} . Increasing the distance $\Delta v_{F,in}$ of the conveying gas velocity relative to the conveying limit $\Delta v_{F,min,in}$ leads to higher critical loads μ_{crit} . The limit curves $\mu_{crit}(p_{in})$ are specific to the particular solid.

Alternative approach: As can be seen from equation (12), the volume fraction of the solid $\varepsilon_{s,in}$ at the beginning of the conveying pipe must remain lower than/equal to a critical volume fraction $\varepsilon_{s,crit}$ in order to achieve pneumatic conveyance. If $\varepsilon_{s,x}$ is replaced on both sides of equation (12) by $Q_{b,x} / Q_s$ (\rightarrow corresponding to equation (11)) and Q_s is removed, the following results:

$$Q_{b,in} \leq Q_{b,crit} \quad (17)$$

i.e. the bulk density $Q_{b,in}$ of the solids entering the conveyor pipe must not exceed a limit value $Q_{b,crit} \cong Q_{ss}$ (\rightarrow loosely tipped bulk density of the current bulk material). Based on the above assumptions, the following applies for density $Q_{b,in}$ from the continuity equation of the solid:

$$Q_{b,in} = \dot{M}_S / (A_R \cdot u_{S,in}) = \dot{M}_S / (A_R \cdot \Delta v_{F,in}) \quad (18)$$

If A_R is back-calculated from the continuity equation of the conveying gas $\dot{M}_F = A_R \cdot Q_{F,in} \cdot v_{F,in}$ and introduced together with equation (18) into equation (17), this results, after a few elementary conversions, in the dependency equation (15) for the permitted or maximum possible load μ_{crit} (\rightarrow this also results with permissible simplifications when using equation (16)). The presented approaches are therefore largely identical

5 Energetic consideration

The back pressure of the infeed, p_G , to the “feed bin” is determined by the downstream process system. This means that the pressure generator used to compress the gas flow \dot{M}_F to the pressure $p_{in} = p_G + |\Delta p_R|$ needs the power P_V .

Generally, $|\Delta p_R| \ll p_G$ under the operating conditions considered here, which means that P_V is hardly determined by the pressure loss Δp_R of the actual conveying section, but is essentially governed only by the size of the gas mass flow \dot{M}_F . For an (idealised) isothermal compression (\rightarrow permanent cooling of the compression chamber), the following then applies, e.g.:

$$P_V = \dot{M}_F \cdot R_F \cdot T_{F,0} \cdot \ln \frac{p_{in}}{p_0} \quad (19)$$

with: $p_0, T_{F,0} = 1.0$ bar, 20 °C, intake suction condition.

Practical calculations show that there are optimum conveying pipe diameters D_R in which the energy requirement P_V due to the pipe diameter D_R (\rightarrow a reduction in diameter leads to an increase in $|\Delta p_R|$ and thus also in p_{in}) and also due to the gas mass flow \dot{M}_F (\rightarrow a reduction in diameter leads to a reduction in \dot{M}_F and the other way round) is minimized. Calculations show that the total energy requirement P_V is increased to a lesser extent by increasing the pressure loss $|\Delta p_R|$ than by increasing the gas mass flow rate \dot{M}_F . The aim of a system design must therefore be to reduce the gas mass flow \dot{M}_F to a level required for the stable attainment of the downstream process (\rightarrow distance $\Delta v_{F,in}$). General tendency: smaller conveying pipe diameters are preferable. This optimisation is specific to both the solids involved and the type of compressor. It is always necessary to check the permissible limit load μ_{crit} .

Example: A hard coal conveyor system with $\dot{M}_S = 18.0$ t/h, $L_R = 50$ m, $p_G = 40$ bar, $D_{R,1} = 50$ mm leads to a pressure loss $\Delta p_{R,1} = 3.00$ bar. A reduction in the pipe diameter with the same $\Delta v_{F,in}$ to $D_{R,2} = 40$ mm results in a pressure loss $\Delta p_{R,2} = 5.34$ bar. With equation (19) the following applies:

$$\frac{P_{V,2}}{P_{V,1}} = \left(\frac{D_{R,2}}{D_{R,1}} \right)^2 \cdot \left(\frac{p_{in,2}}{p_{in,1}} \right) \cdot \frac{v_{F,in,min,2} + \Delta v_{F,in}}{v_{F,in,min,1} + \Delta v_{F,in}} \cdot \frac{\ln \left(\frac{p_{in,2}}{p_0} \right)}{\ln \left(\frac{p_{in,1}}{p_0} \right)}$$

mit: $v_{F,in,min,1} \cong v_{F,in,min,2}$ (20)

For the operating data given above, the following extremely optimistic value is calculated for the selected $\Delta v_{F,in}$:

$$\frac{P_{V,2}}{P_{V,1}} \cong 0,69$$

Independently of the planned system concept, it follows from the considerations presented here that the systematic optimisation of energy expenditure is obviously a sensible approach to the system design.

6 Gas exchange in the conveying system

If an inert gas F1, compare Figure 1, is used for conveying solids from the “lock hopper system” to the “feed bin” for reasons of safety engineering or explosion protection, but an oxidising gas medium F2 for heat generation, e.g. air, oxygen, carbon dioxide or water vapour, is used for the feed into the process plant, e.g. a gasification plant, the gas flow F2 mixes at the outlet of the “feed bin” with the gas flow F1, as this splits into the exhaust gas flow $\dot{M}_{F1,Abgas}$ and the gap gas flow $\dot{M}_{F1,c}$ at the head of the “feed bin”. The volume fraction $\epsilon_{F,feed}$ of the gas F1 that arises as the gap volume of the bulk material in the “feed bin” is conveyed with the solid to its outlet ($\rightarrow u_s = u_F$ applies) and mixes

there with the gas F2. The gas mass flow $\dot{M}_{F1,\epsilon}$ can be calculated using the continuity equations of the two phases involved as follows
with: ρ_{F1} - density of gas F1 at pressure p_G .

$$\dot{M}_{F1,\epsilon} = \frac{\epsilon_{F,feed}}{1 - \epsilon_{F,feed}} \cdot \frac{\rho_{F1}}{\rho_S} \cdot \dot{M}_S \quad (21)$$

Note: In moving bulk quantities of solids, the relative gas gap volume ϵ_F is generally greater than it is in a stationary bulk quantity ($\rightarrow \epsilon_{F,feed}$) and the process of feeding into a container can lead to further loosening or fluidisation of the bulk solids, especially in the case of fine-grained materials.

These effects and pressure differences over the height of the “feed bin” must be taken into account separately.

It is not possible to prevent the mixing of gases F1 and F2 at the outlet from the “feed bin” in the design variant shown in Figure 1. It can only be eliminated, if necessary, by suitable flushing with F2 gas upstream of or integrated into the “feed bin”. The F1/F2 mixture from the flushing process must then be discharged separately.

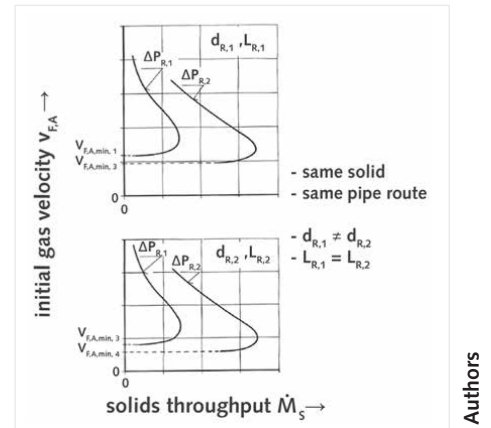
Example: Under consideration is a gasification plant with a solids throughput of $\dot{M}_S = \dot{M}_{S,feed} = 18.0$ t/h of pulverised hard coal with the parameters particle diameter $d_{s,50} = 27 \mu\text{m}$, solids density $\rho_S = 1650 \text{ kg/m}^3$, bulk density $\rho_{ss} = 550 \text{ kg/m}^3$, pressure in the “feed bin” $p_G = p_{feed} = 40.0$ bar, and operating temperature $T_S = 20 \text{ }^\circ\text{C}$. The F1 gas used is CO_2 with $\rho_{\text{CO}_2} = 1.842 \text{ kg/m}^3$ at $20 \text{ }^\circ\text{C}$, 1 bar. Using equation (11), the resulting value for the relative void volume of the gas in the “feed bin” is $\epsilon_{F,feed} = 1 - \rho_{ss}/\rho_S = 1 - 550 \text{ kg/m}^3 / 1650 \text{ kg/m}^3 = 0.666$. The density of the F1 gas at the operating pressure p_G is: $\rho_{F,1} = \rho_{\text{CO}_2}(20^\circ\text{C}, 1.0 \text{ bar}) \cdot p_G / p_0 = 1.842 \text{ kg/m}^3 \cdot 40.0 \text{ bar} / 1.0 \text{ bar} = 73.68 \text{ kg/m}^3$. Equation (21) then provides the F1 gas mass flow in the gap volume of the downward-moving bulk solid as:

$$\dot{M}_{F1,\epsilon} = \frac{0,666}{1 - 0,666} \cdot \frac{73,68 \text{ kg/m}^3}{1650 \text{ kg/m}^3} \cdot 18,0 \frac{\text{t}}{\text{h}} = 1,608 \frac{\text{t}}{\text{h}}$$

This gas flow must be included in the design of the process reactor. In practice, as indicated above, the involved $\dot{M}_{F1,\epsilon}$ value will be somewhat greater than calculated here [3].

7 Required extent of test work

In order to be able to carry out the examinations described in sections 2, 3 and 4, it is necessary to know or determine by means of experimentation not only the dependency $v_{F,in,min}$ for the respective bulk solids, but also the dependency for determining the conveying pipe pressure loss Δp_R . For an



6 Required extent of test work for a reliable system design: ($v_{F,A} = v_{F,in}$)

unknown/new bulk material, this at least requires measurements on two conveyor sections of the same length, running as parallel as possible, i.e. geometrically similar conveying sections $L_{R,1} = L_{R,2}$ with very different diameters $D_{R,1} \neq D_{R,2}$, compare Figure 6. At least two conveyance characteristic curves $(\Delta p_{R,1}, \Delta p_{R,2}) = \text{konst.}$ must be recorded on

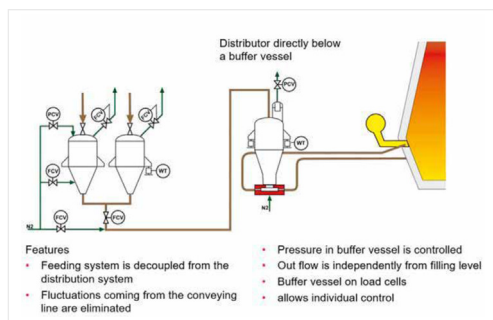
both pipes all the way down to the conveyance limit $v_{F,in,min}$. The pressure differences $\Delta p_{R,1}$ and $\Delta p_{R,2}$ should be the same in both conveying sections, but should be as far apart from each other as possible. Details on the procedure can be found in [2, 4] and other sources.

The data then available can be used to back-calculate the dependencies of the minimum velocity characterising the respective bulk solids, as well as the resistance coefficients describing the pressure loss of the pipe. The latter can be performed using different models, e.g. the λ_s -standard model or company-specific approaches, compare [1, 4, 5] and other sources.

8 Design variants

The high back pressures in pneumatic conveying systems mentioned here naturally occur not only in gasification plants but also in other processes, e.g. in the charging of blast furnaces with pulverised coal as a heat transfer and reduction medium. The feeding / lock systems use pressure vessels only.

The “lock hopper system” is usually designed as a parallel configuration of two feeders (\rightarrow one feeds, while the other is being filled, i.e. quasi-continuous operation), but could also be designed as a series configuration of two feeders arranged one above the other, i.e. absolutely continuous operation. Both processes have their advantages and disadvantages. The latter solution generally results in a very high overall installation height. Depending on the design, the bulk material is often fed from the feed bin into the downstream reactor in parallel at several feed points, see Figure 1. Figure 7 shows an example of a system for injecting coal into a blast furnace.



Authors

7 Design of a system for injecting coal into blast furnaces by Claudius Peters Projects [6]

Typical conveying pressures range up to $p_{in} \cong 20$ bar. From the distributor below the “buffer vessel” (= “feed bin”) the coal can be distributed in parallel to any number of outgoing conveying pipes.

The solids throughput \dot{M}_s is regulated via the top pressure in the “buffer vessel”. Compared to the design of a feed bin with, for example, multi-cone discharge, which is difficult to implement in terms of construction and strength verification, the design of the buffer vessel with integrated

distributor shown in Figure 7 appears to be advantageous. This design allows easy adaptation to downstream gasification/reactor systems of various designs, as the number of conveying pipes leading from the manifold can be individually customised from 1 ... n. A tried and tested resistance equalisation of the “n” individual pipes leaving the distributor allows the same or also defined different solids flow rates in these pipes. [7].

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